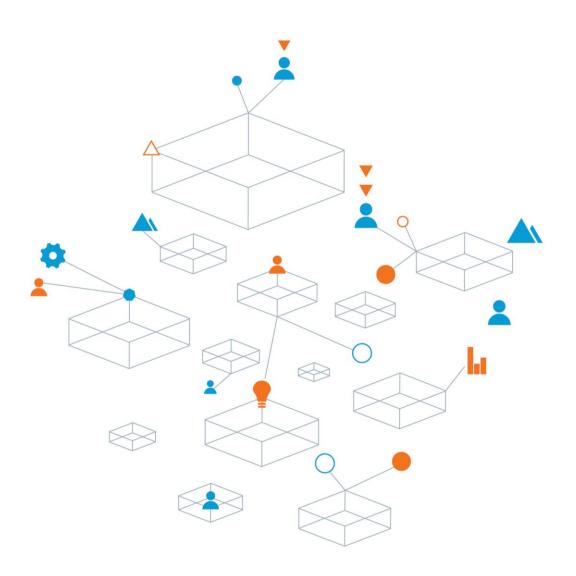


Xtra-Gold Resources Corporation

Kibi Gold Project

Effective Date:30 September 2024



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DOCUMENT INFORMATION

Author(s):

Dr Corné Koegelenberg TECT – Principal Geoscientist Pr.Sci.Nat., MGSSA, MSEG

(Structural Geology, 3D Modelling)

Dr Ian Basson TECT – Principal Geoscientist Pr.Sci.Nat., FGSSA, MSEG, AMSAIEG

(Structural Geology, 3D Modelling)

Ken Lomberg Director (Geology and Resources)

BSc (Hons) Geology, BCom, MEng (Mining),

Pr.Sci.Nat.

Effective Date: 30 September 2024

Project Number: XTR002

Document Review and Sign Off

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Author Ken Lomberg

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Author

Dr Ian Basson

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Author

Dr Corné Koegelenberg

This is a scanned signature held on file by Pivot. The person and signatury corperts to its use only for the purpose of this document.

Reviewer

Chris Fredericks

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Appendix A – Authors Certificate

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Appendix B – Apapam Mining Lease Status Opinion prepared by REM Law Consultancy

1 SUMMARY

TECT Geological Consulting (TECT) and Pivot Mining Consultants (Pty) Limited (Pivot) were requested by Xtra-Gold Resources Corp. (Xtra-Gold) to prepare an Independent Technical Report consistent with the Canadian Securities Administrators National Instrument 43-101 and Form 43-101F1 for the Apapam Concession (LVB 5191/09). The Apapam Concession forms part of Xtra-Gold's Kibi Gold Project in southern Ghana.

The purpose of this report is to publish an Independent Technical Report summarizing the geology, past exploration activities and Mineral Resource Estimate on the Kibi Gold Project with the updated mineral resource estimate based on drilling completed since the October 2012 to January 2024.

Through its subsidiary companies, Xtra-Gold holds three (3) concessions that are contiguous in the Kibi Gold Project, including the Apapam Mining Lease. The Apapam Concession is a granted mining lease, whereas the other two titles are currently applications that have been submitted to the Minerals Commission.

1.1 Project Area and Location

The Apapam Concession is located approximately 75 km north-northwest of the capital city of Accra, in the East Akim District of the Eastern Region of Ghana, on the eastern flank of the Atewa Range near the headwaters of the Birim River. The centre of the concession is situated at approximately 6° 09' 30" West Longitude and 0° 34' 15" North Latitude (WGS 84). The Kibi Project area covers 3,365 ha and is located at the northern extremity of the Kibi Winneba Greenstone Belt.

1.2 Accessibility, Climate, Local Resources, Infrastructure and Physiography

The concession is located approximately 75 km north-northwest of Accra and can be accessed by two asphalted secondary highways. Ghana has a fairly good network of paved highways and roads. Within the Apapam Concession, numerous tracks and paths are available for easy access to most points.

The climate within the area is equatorial with relatively high humidity throughout the year. Rainfall is mainly confined to two periods, being high and unpredictable especially during the peak period which falls in May/June with a second peak in September/October. The dry season is generally from January to February. Temperatures ranges between 22°C and 35°C. Operations can be undertaken throughout the year.

The topography of the Apapam Concession is characterized by steeply sloping ridges and undulating mountain sides and hills due to the prominent, NNE-trending Atewa Range which is approximately 50 km long and 10 km - 15 km wide, that dominates the area. The range consists of steep-sided hills with fairly flat summits which indicates the last remnants of the Cenozoic peneplain that once covered southern Ghana. The peneplain locally contains ancient bauxitic soils. The steep flanks feature a wide variety of high canopy tropical

hardwoods typical of south-western Ghana whereas the summit has a diverse flora, including extensive hanging vines. The range is the site of the country's largest Forest Reserve.

The infrastructure in the Kibi District is fairly well-developed. The town of Kyebi (Kibi) is a major regional centre with a population of 11,700 (2013). Kibi is connected to the national electricity supply network, and hospital, postal and other community facilities are available. Extensive mining infrastructure is in place in all of the major gold producing areas of Ghana.

1.3 History

Virtually all of the past gold mining activity that is locally evident has focused on the alluvial gold occurrences found in the many river valleys throughout the area.

A London-based junior, Akim (1928), Ltd is known to have carried out some exploration and development work on the quartz vein at the Kibi Mine on the outskirts of the town of Kibi. It is likely that some of the other known vein occurrences in the area were explored but apparently with little economic success.

Exploration was undertaken a by a number of companies in the latter part of the 20th century. Sun Gold Group were granted two concessions in the area, which were taken over by Shefford Resources of Toronto. Shefford did some initial work on the hard rock potential including a Bankable Feasibility Study in 1989. After corporate restructuring the mining leases were put into a new company, Goldenrae Mining Company, and Sikaman Gold Resources of Toronto amalgamated with Shefford. Sikaman then brought in a senior partner, the London-based ITM group who financed an alluvial operation which eventually closed down in 1993.

In the early 1990s, the EQ Resources group of Toronto also picked up a large concession (Apapam) on the eastern flank of the Atewa range, covering the drainage of the upper Birim River in the vicinity of the town of Kibi. EQ carried out a successful pitting program in cooperation with Goldenrae, with the intention of setting up a satellite production unit under Goldenrae management.

The remainder of the work was focused on the potential and profitability of other alluvial deposits.

Xtra-Gold Mining's interest in Kibi Gold Project was previously via a prospecting licence granted by the Government of Ghana on March 29, 2004, covering a licensed area of 33.65 km². In May 2008, Xtra-Gold Mining applied to the Government of Ghana to convert the Kibi prospecting license to a mining lease, which was granted. The mining lease expired in December 2015 but all required documentation to extend the lease for 15 years from December 17, 2015 has been submitted to the Ghana Minerals Commission. As these extensions generally take years for the regulatory review to be completed, Xtra-Gold is not yet in receipt of the extension approval. However, until Xtra-Gold receives the extension documents, the old lease remains in force under the mineral laws. The extension is in accordance with the terms of application and payment of fees to the Minerals Commission of Ghana.

Xtra-Gold have been granted surface and mining rights by the Government of Ghana to work, develop and produce gold in and from the Apapam lease area (including the processing, storing and transportation of ore and materials).

In 2012 a maiden Mineral Resource was declared. This included the Big Bend, Double 19, East Dyke, Mushroom and South Ridge targets (Table 1.1).

Table 1.1: Kibi Project Mineral Resource 2012 Declared in terms of the guidelines of the CIM Standards													
	Category	Tonnage (Mt)	Gold Grade (g/t)	Gold oz									
Total	Indicated	3.38	2.56	277,000									
Total	Inferred	2.35	1.94	147,000									

In 2021 a revised Mineral Resource was declared. This included the Big Bend, Double 19, East Dyke, Mushroom, South Ridge and Gate House and Gold Mountain targets (Table 1.2).

Table 1.2: Kibi Project Mineral Resource 2021 Declared in terms of the guidelines of the CIM Standards											
Indicated	Tonnage (t)	Density (t/m3)	Grade – Au (g/t)	Au (oz)							
Big Bend	6,472,000	2.78	1.48	307,400							
Double 19	1,584,000	2.62	1.38	70,400							
East Dyke	3,102,000	2.72	1.49	148,800							
Mushroom	505,000	2.64	1.37	22,300							
Road Cut	225,000	2.80	0.85	6,100							
South Ridge	2,005,000	2.70	1.07	68,700							
Total	13,893,000	2.73	1.40	623,700							
Inferred	Tonnage (t)	Density (t/m3)	Grade – Au (g/t)	Au (oz)							
Big Bend	1,257,000	2.82	1.03	41,400							
East Dyke	1,128,000	2.84	1.19	43,300							
South Ridge	943,000	2.82	1.02	30,800							
Gate House and Gold Mountain	2,366,000	2.76	0.79	65,200							
Total	5,694,000	2.80	0.99	180,700							

1.4 Geological Setting

Xtra-Gold's Kibi Project is hosted within the Kibi Belt, which forms the northern continuation of the Paleoproterozoic Kibi-Winneba Greenstone Belt (KWB), located in south-western Ghana. The KWB is correlated to the west with the well-documented Ashanti-, Asankrangwa- and Sefwi Biriman Fold Belts, which are hosts to world-class gold mines and deposits. These include, but are not limited to the Akyem Newmont, Obuasi Anglo Gold Ashanti, Tarkwa Goldfields and Ahafo Newmont Mines.

The Kibi Project targets are hosted in intensely-sheared and tightly-folded volcanometasedimentary strata with interlayered felsic- to intermediate- to mafic-dykes and sills, that is regionally correlated with the Birimian Super Group and Tarkwa Group.

Targets are predominantly hosted within, or in the vicinity of, felsic- to intermediate (e.g., granite to granodiorite) dykes or sills, in contact with metasediments (e.g., phyllites or greywackes), that occupy sheared fold limbs or fold hinge zones. These structural sites are classic type-locations for intensified brittle-ductile deformation and concomitant hydrothermal fluid ingress, with eventual auriferous mineralization. Ongoing exploration is geared to identifying and delineating such sites on a licence- to target scale.

1.5 Deposit Types and Mineralization

The Kibi project hosts primary gold mineralization of economic significance. Auriferous bodies discovered to date on the Apapam Concession by Xtra-Gold, include: Big Bend, East Dyke, Mushroom, Road Cut, and South Ridge in Zone 2, Double 19, Twin Zone and Boomerang in Zone 3, and Gate House and Gold Mountain in Zone 1. Collectively, these ten auriferous bodies lying within approximately 2.7 km of each other are estimated to encompass an Indicated Mineral Resource of 22.2 Mt grading 1.18g/t gold for a contained 842,400 oz and an additional Inferred Mineral Resource of 5,694,000 Mt grading 0.99 g/t gold for a contained 180,700 oz.

These targets can be classified under classic, structurally-controlled mesothermal/orogenic gold. Mineralization is primarily associated with quartz albite-carbonate-sulphide tensional vein stockworks developed in or especially near the margins of sills, dykes, and possibly small plutons (stocks) of granodiorite, quartz diorite, and tonalite bodies.

Generally, in comparison with other relevant gold deposits in Ghana, the Kibi Project gold mineralization is characterized by auriferous quartz vein sets hosted in felsic to intermediate (Belt-type) granitoids that are geologically analogous to other Granitoid-hosted gold deposits of Ghana, including Kinross Gold's Chirano and Newmont Mining's Subika deposits in the Sefwi gold belt.

From 2012 to 2021, exploration efforts were primarily focussed on the continued advancement of early-stage gold shoots and showings within the Zone 1 – Zone 3 maiden mineral resource footprint area.

From 2021 – 2023, exploration efforts were primarily focussed on the continued advancement of early-stage gold shoots and showings within the Zone 3 gold-in-soil target area. Continued reconnaissance work also targeted Zones 4, 5 and 9 (Apapam mining lease), and Zone 6 on the Akim Apapam reconnaissance licence application.

1.6 Exploration and Drilling

1.6.1 2006 – 2007 Exploration Program

Two (2) separate work programs were conducted on the Apapam Concession during 2006-2007. The first work program was undertaken and managed by CME Consultants Inc. (CME); a Canada-based geological consultancy with over 15 years of project management experience in Ghana. The second program was undertaken and managed by Xtra-Gold personnel.

The Phase I exploration program was designed to test the Apapam Concession on a regional scale. The field work was implemented by CME from August 12 to September 23, 2006 and included:

- Concession-Wide Stream Sediment Sampling (88 Samples Collected From 44 Sites);
- Survey Grid Establishment (33.78 Line-Km);
- Soil Sampling (1,306 Samples);
- GPS Surveying (33.78 Line-Km);
- Rock Sampling (89 Samples); And
- Historical adit and bulldozer cut sampling (100 samples).

The Phase II exploration program consisted of a reconnaissance trenching program intermittently implemented by the Xtra-Gold exploration staff from February 2007 to December 2007. The trenching was carried out to test the depth extent of geochemical signature from gold-in-soil anomalies that were detected within the north-western portion of the concession during the Phase I work program. A total of 542 channel samples were collected from 21 trenches totalling 1,090 linear metres. In order to obtain an independent assessment of the 2007 Xtra-Gold trenching results, a NI 43-101 compliant data verification program was undertaken by CME in December 2007. The program involved the re-sampling of selected trenches which yielded exploration-significant gold mineralization intervals.

Gold mineralization on the Apapam Concession was found to occur in several different geological setting, including steeply-dipping and flat-lying quartz veins and alteration haloes proximate to the quartz veining.

No drilling was undertaken at this time.

1.6.2 2008 - 2010 Exploration Program

Exploration work on the Apapam Concession during the 2008-2010 reporting period was aimed at advancing the Kibi Project which encompassed a 5.5 km long mineralized trend delineated from gold-in-soil anomalies, trenching, drilling, and geophysical interpretation along the northwest margin of the Apapam Concession. This trend and extent are characterized by widespread gold occurrences of the granitoid hosted-type.

An extensive soil geochemistry survey covering approximately 47 line-kilometres (1,827 samples) was implemented in early 2008 to further define the extensive gold-in-soil trend. Regolith development in most of the Kibi Project area is favourable for soil sampling. The generally steep topography along the flank of the Atewa Range has resulted in relatively thin

colluvial (lateritic gravel) cover in the project area. As a result, gold-in-soil anomalies on steeper slopes and ridges probably reflect a good, although not exactly quantitative, measure of gold distribution in the underlying saprolite.

The entire Kibi Project grid was also covered by Induced Polarization (IP)/Resistivity (~ 64 km) and ground magnetometer (~79 km) surveys to help define the lithological and structural pattern of the mineralized trend, and prioritize trench/drill targets. Although the widespread granitoid-hosted gold mineralization in Zone 2 is typically characterized by 1-3%, locally up to 5%, disseminated sulphides (pyrrhotite, pyrite, arsenopyrite), the known mineralization occurrences failed to produce a chargeability response. In Zone 3, a moderate chargeability (IP) / very high resistivity anomaly is spatially associated with an approximately 135 m wide, NE-trending, granitoid hosted, structural corridor that appears to encompass at least five (5) distinct, gold-bearing, sheeted to stock worked vein zones. The south-western portion of the gold-in-soil trend is characterized by an approximately 3.5 km long, NE-trending, generally moderate to high chargeability/ moderate to high resistivity anomaly lying along or proximate to the contact between the southern resistive domain and the central conductive corridor. This exhibits a spatial relationship with a geophysically inferred, NE-trending, regional structural trend.

Exploration activities in 2008 included a manual trenching program with the excavation of 18 trenches totalling approximately 1,217 linear-metres, including: 4 trenches (302 m) on Zone 2; and 14 trenches (915 m) on Zone 3 of the 5.5 km long gold-in-soil trend. In addition, 67 mechanical (i.e., excavator) trenches totalling approximately 2,223 m were also excavated in conjunction with the 2008 and 2009 drilling programs. The trenches and trench sampling identified extensive, granitoid-hosted, auriferous quartz vein systems.

A total of 109 grab samples were collected by Xtra-Gold during prospecting and reconnaissance geology traverses that were primarily designed to follow-up on gold-in-soil anomalies. Where anomalous values were returned, the locations were subsequently tested by trenching and/or drilling.

As part of the ongoing exploration efforts, Xtra-Gold commissioned SRK Consulting (Canada) Inc. (SRK) to conduct a structural study of the Apapam Concession. The goal of the study was to investigate key exposures and available drill core to document and understand the structural controls on gold mineralization at the Kibi Project. SRK reviewed 14 diamond drill holes (Zone 1 and 2), as well as available trench exposures (Zone 2 and 3) on the Apapam concession from March 16 to 27, 2010. Due to diamond drilling density and accessible trenches, SRK's structural study focused largely on Zone 2 of the Kibi Project. SRK also reviewed Xtra-Gold's geological and structural mapping to date for zones 1, 2 and 3 of the Kibi Project.

A petrographic study was also implemented in March 2010 to characterize the lithological units and ore mineralogy of the Kibi Project. A total of 36 thin sections and nine (9) polished sections were studied by Professor K. Dzigbodi-Adjimah of the University of Mines and Technology, Tarkwa, Ghana. The findings of the structural and petrographic studies are incorporated in the property structure and mineralization sections of the technical report, respectively.

The first drilling campaign on the Apapam Concession area was conducted by Xtra-Gold in 2008. A total of 68 drill holes totalling 7,716 linear metres were drilled, including 18 diamond core drill holes in 2008 (3,001 m) and 50 reverse circulation (RC) drill holes in 2009 (4,715 m). The drill holes targeted the Zones 1, 2, and 3 gold-in-soil anomalies.

1.6.3 2010 - 2012 Exploration Program

A VTEM survey was flown over the Kibi Gold Project by Geotech Airborne from December 2010 to February 2011. The survey measured ground elevation, radiometrics, magnetic field and electromagnetism (resistivity). Interpretation of the data resulted in an interpretive pseudogeology map of the area. The different geophysical units may be correlated with various geological units. Two kinds of targets were defined: Resistive type ((high-resistivity areas within interpreted conductive graphitic shear zones and graphitic sedimentary units) and Granitoidstype (shear/fracture zones in basin-type granitoids). More detailed processing has highlighted the various faults around the project area.

From June 2011 to April 2012, 3,833 soil samples were collected every 25 m along lines 200 m apart in the south eastern part of the concession, as this had previously not been sampled.

One hundred and fifteen (115) rock grab samples were collected in conjunction with the 2011 – 2012 soil geochemical survey and follow up prospecting of gold-in-soil anomalies.

Two Hundred (200) trenches were excavated at a number of prospects in an effort to help define the extents and geological context of gold mineralisation. Due to the high relief in the area, some of the trenches are actually cleaned or cleared road cuttings on the sides of the hills. A total of 4,346 horizontal channel samples were taken and due to the prevalence of shallowly dipping veins, 509 vertical channel samples were also taken where appropriate.

SRK Consulting (Canada) investigated the structural geology of the Kibi Gold Trend project – Zone 2 in November 2011. The work was focussed on Big Bend, wherein the structural controls on the mineralisation were proposed.

SRK Consulting (Canada) analysed the regional structural geology, regional aeromagnetics and the VTEM data. Areas of structural complexity representing prospective targets for further exploration in the Kibi area, were highlighted.

A total of 188 diamond drill holes were drilled for 41,372 m of drilling with 33,961 samples being taken from the core.

1.6.4 2012 - 2021 Exploration Programme

Further soil sampling was undertaken on the Cobra Creek (Zone 5) and Akwadum South (Zone 7) targets. A total of 10.7 line-kilometres of sampling on a 100 m x 25 m grid was undertaken with a total of 458 samples being collected.

Sampling of outcrop and float material was performed at Cobra Creek (Zone 5) and other locations on the Apapam Concession with a total of 910 samples being collected.

Trenching was undertaken from October 2018 – June 2021 on Zone 1 – Zone 2 – Zone 3 to supplement the drilling of the targets already recognised. Fifty-six (56) trenches (1,577.7 sample metres) were excavated on the majority of the targets that form part of the mineral resource estimate. An additional 38 trenches (1,170.8 sample metres) were also completed on an intermittent basis from September 2012 - March 2016 to delineate the Cobra Creek (Zone 5) auriferous shear system.

A total of approximately 15,000 m² of bedrock exposures were mechanically stripped and power washed at the Cobra Creek target from 2012 – 2016 to permit systematic mapping and channel sampling of the auriferous shear system. With total of 1,312.26 m of saw-cut channel sampling and 71 m of chip-channel sampling completed on the stripped bedrock exposures.

A total of 118 auger drill holes with a cumulative meterage of 390.89 m were sunk at Akwadum South (Zone 7) during 2018 and 2019.

Sixty-two (62) scout pits were excavated manually to follow up on soil geochemistry and hand auger anomalies.

Ground geophysics was conducted at Cobra Creek target, consisting of 32.2 line-kilometres of IP/Resistivity survey and 14.99 line-kilometres of magnetic survey.

A total of 212 diamond drill holes totalling 25,198.55 m was drilled during this reporting period. With most of the drilling,158 drill holes totalling 21,321.45 m (85%), completed from February 2018 – June 2021 on targets within the Zone 1 – Zone 2 – Zone 3 Mineral Resource estimate footprint area.

1.6.5 2021 - 2023 Exploration Programme

Geophysical modelling of an area of approximately 70 km², a subset of an airborne VTEM – Mag survey completed by the Company in 2011, was undertaken as per the Recommendations of the 2021 technical report. Additional structural interpretation and 3D litho-structural modelling covering an area of approximately 41 km² was undertaken, with the study based on additional drilling in combination with the 3D VTEM / TMI inversion models. The intention being to further define the structural controls of the gold mineralization and to generate high-priority exploration targets to help guide ongoing resource expansion drilling. The resultant geological models formed the basis for the 2024 Mineral Resource estimate.

Infill soil sampling was conducted on the Central Fold (Zone 9) target in 2023 with the aim of ground proofing the gold-prospective litho-structural setting generated by the 3D VTEM / TMI inversion modelling. A total of 542 samples were collected from 17.3 line-km of cut lines on a $100 \text{ m} \times 25 \text{ m}$ infill sampling grid.

Traditional prospecting, including the collection of 165 composite chip samples from outcrop and rock float material, was performed at the Central Fold (Zone 9) target to ground proof the inferred first order F2 anticlinorium fold structure generated from the geophysical modelling. A total of 438 composite chip samples was collected during Zone 4 prospecting efforts aimed at

further delineating auriferous silicified siltstone float trains. The sampling of other isolated locations was also undertaken.

Trenching activities during the current reporting period focused primarily on the Boomerang and Twin Zone targets in Zone 3, with a total of 20 trenches (581.8 m) completed to delineate the near surface litho-structural and mineralization nature of these two new mineral resource estimate bodies. Four (4) scout trenches totalling 129.3 m were also excavated on the Central Fold (Zone 9) grassroots target.

A total of 174 diamond drill holes totalling 34,737.1 m were completed by the Company's inhouse drill crews from July 2021 – January 2024 on the Kibi Gold Project. Drilling activities focused primarily on Zone 3 (32,497.3 m), including 144 drill holes (30,052.3 m) dedicated to the advancement of the early-stage Boomerang and Twin Zone targets to the mineral resource phase. Scout drilling on the Cobra Creek (Zone 5) target located at the northeastern extremity of the Apapam concession was also completed.

1.7 Sample Preparation, Analyses and Security

1.7.1 Drill Core Samples

Drill core obtained from diamond drilling is targeted directly from the core tube into wooden core boxes, marked with the drill hole number and depth information. In the case of saprolite material the core is laid directly onto a strip of plastic wrap placed inside the box and then securely wrapped around the core to stabilize and prevent the dehydration of the saprolite.

At the Kwabeng exploration camp core shack the core is laid out along an angle iron on a work bench and meticulously re-assembled piece by piece with the core aligned with the orientation marks at the bottom of each 3 m drill run. The core is then measured, core recovery and RQD information collected, and photographs of each individual box taken. A company geologist subsequently conducts geological logging of the core and marks the sample intervals. The core is sampled over nominal 1 m intervals; with adjustments where necessary for mineralized structures.

The diamond drill core is then saw-split lengthwise and half the core is immediately placed into a labelled plastic bag with a unique sample ticket stapled to the inside lip of the bag, and securely sealed by staples. The remaining half of core is returned to the core box and the box stored in a secure facility.

1.7.2 Reverse Circulation (RC) Drill Samples

Reverse circulation drill samples are collected immediately at the drill hole site. The drill sample cuttings are collected in a cyclone over one (1) meter sample intervals; with the cyclone being purged after every 6 m drill run. The dry RC bulk chip sample (\sim 25 to 30 kg) is then weighted and passed through a two – stage riffle splitter to produce a nominal 2 – 3 kg sample for assay which is also weighed on site.

Drill cuttings from each sample interval are screened – washed and a quick log of the rock chips completed at the drill site by a company geologist; noting amongst other things the sample quality/recovery, weathering profile, main lithologies, prominent alteration, and the character of the mineralization (i.e., oxide versus sulphide).

1.7.3 Auger Sampling

Hand auger sampling is routinely utilized to test the geochemical signature of gold-in-soil anomalies at depth within the saprolite horizon to better define trenching targets. Sampling is typically conducted at one (1) metre intervals with the material from the drill hole's first metre discarded to ensure the collection of in situ material. Auger hole spacing is typically 25 m, with some 12.5 m in-filling. To avoid any contamination only dry samples are collected.

1.7.4 Chain of Custody

A typical chain of custody exists for all samples. Exploration samples, including soil, auger, and trench samples, and Reverse Circulation (RC) drill samples are collected, numbered and bagged in the field and then transported from the field to the Kwabeng camp under Xtra-Gold's supervision. Upon arrival at camp the samples and sample numbers are checked, and the samples secured in rice sacks with a numbered security seal (i.e., nylon zap strap) for shipment to the laboratory. The diamond drill core is collected and transported to the camp's logging facility by Xtra-Gold's in-house drill personnel. All samples are kept in a locked facility at the camp. Samples are transported to the laboratory or collected by the laboratory on a weekly basis. A Tracking Record of all sample deliveries/pickups and pending analytical orders is kept by the Project Geologist, including personnel in custody of samples and time of departure; laboratory drop-off site; status of security seals; and assay turnaround time. Any discrepancies are noted by the laboratory who subsequently advise Xtra-Gold's management to this effect.

1.7.5 Analysis

Pre-2008 samples were analysed by SGS Laboratory Services in Tarkwa. Aqua Regia and fire assay methods were used to determine the gold concentration.

From 2008 – 2017 analyses were undertaken by ALS Chemex at their Kumasi Laboratory. All samples, including drill core and RC chips, trench channel, hand auger, rock, and soil samples, were typically analysed for gold only utilizing ALS Chemex's Au-AA24 method: Fire Assay Fusion with Atomic Absorption Spectroscopy (AAS) Finish.

Since 2017 the analysis has been undertaken by Intertek at their laboratory in Tarkwa. Typically the analysis for gold uses method FA51/AA which is a lead collection fire assay (50 g aliquot) fusion method with an Atomic Absorption Spectroscopy finish (AAS). In March 2023, the analytical methods coding was changed to FA50/AA from FA51/AA, and to FA50L/AA from FA50/AA, by Intertek Minerals. The analytical methods remain the same as the original fire assay with an Atomic Absorption Spectroscopy finish methodology.

1.7.6 Quality Control and Quality Assurance

Quality-Control Programmes have been implemented to ensure best practice in the sampling and analysis of the diamond drill core, reverse circulation (RC) chip samples, saprolite trench and saw-cut channel samples, and soil samples, and that the data can be used to inform subsequent work and the progression of the project.

1.8 Data Verification

The following conclusions and recommendations are made:-

- The operator is diligent in the use of the QA/QC programme with the recording of data for analysis. An important aspect is that an effective and dynamic QC programme is utilised to review data as it comes in from the laboratory; a practice currently being applied on site.
- The assessment of the blanks confirms that there is minimal contamination at the laboratory.
- The assays were undertaken utilising a 50 g aliquot for the fire assay whereas the certified reference materials (CRMs) generally utilised a 30 g aliquot. The expectation is that the larger aliquot should produce results that are better grouped (precision) and more accurate.
- An excessive number of different CRMs have been previously used. It is recommended that fewer are used in the future. It is considered more practical to identify 2 to 5 different CRMs to span the assay range of the expected grades i.e., 0.5 2 ppm. This will allow more control and conformation of the data, i.e., identification of sample swaps in particular.
- In the earlier programmes numerous failures of the Xtra-Gold CRMs have been noted. Most of these have been attributed to the misidentification of the CRMs. The laboratory CRMs demonstrate that the data can be considered to be accurate.
- The precision is tested by analysis of the duplicate data. The results of the duplicate analysis presented suggests that the precision is an issue. This is probably truer for the higher grades and may be related the presence of coarser grains of gold.

It was concluded that the geochemical data used in the resource estimation was satisfactory, with variations most probably due to the nature and deportment of the gold and probably related to the presence of coarse gold in the deposits.

1.9 Mineral Processing and Metallurgical Testing

A Gold Deportment Study which assessed the mineralogical and metallurgical aspects of the gold mineralization in the Kibi Gold Project was completed in October 2011 by SGS South Africa (Pty) Ltd. for both sulphide and oxide composites. The results of the study were:

- Gold in the sulphide samples is highly amenable to cyanidation leaching with ~97% recoverable by means of direct cyanidation.
- The grading analysis on the sulphide sample indicated a very high upgrading of gold in the +106µm size fraction (~69%).
- The direct cyanidation and diagnostic leach indicates that the sulphide sample is highly amenable to cyanide leaching, with ~97% of the gold recovered from the head sample at a grind of 80%-75µm by direct cyanidation and ~96% for the gravity tailings at a grind of~50%-75µm.

- The gold in the composite oxide sample is also highly amenable to cyanidation, with ~97% of the gold recoverable by means of direct cyanidation. The grading analysis on the composite oxide sample indicated a very high upgrading of gold in the +106μm size fraction (~74%).
- The direct cyanidation and diagnostic leach tests indicated that the oxide sample is highly amenable to cyanide leaching, with ~98% of the gold recovered from the head sample at a grind of 80%-75μm and ~99% of the gold in the gravity tailings at a grind of 50%-75μm.

It was concluded that the most simplistic processing option would be to mill the material to ~80%-75µm followed by carbon-in-leach cyanidation.

1.10 Mineral Resource Estimate

The reported mineral resource is presented in Table 1.3.

<u>Cautionary Note:</u> The Mineral Resource Estimate calculation includes the area termed the "Buffer Zone", which encompasses the "open ground" between the existing Apapam mining lease boundary up to the Forest Reserve boundary. Xtra-Gold applied for this Buffer Zone in the original application of extension/renewal of the Apapam mining lease on June 17th, 2015, and also in the updated extension/renewal dated June 28th, 2019. These applications were approved by the Technical Committee of the Minerals Commission and certified to be in conformity with the official cadastral system introduced under Ghana Mineral Law. Xtra-Gold is currently awaiting formal approval of the extension/renewal. Refer to the Apapam mining lease Status Opinion prepared by REM Law Consultancy (Appendix B) for further details on the lease renewal / extension, including the Buffer Zone area.

Although the Company has taken all legal steps to extend the lease with the addition of the Buffer Zone, there is no assurance that such approval will be granted. In the interim, under Ghanaian mining law, the existing lease continues to remain in full force and effect.

Pivot Mining Consultants (Pty) Ltd

Table 1.3: Mineral Resource Declaration - September 2024 Declared in terms of the CIM Standards Cut-off: Au 0.5 g/t

	Арар	oam License a	ınd Buffer Are	as	Buffer Z	one between L Reser		Forest	Apapam License Area Indicated					
		Indica	nted			Indicat	ed							
	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs		
Big Bend	6,472,000	2.78	1.48	307,400	-	-	-	-	6,472,000	2.78	1.48	307,400		
East Dyke	3,102,000	2.72	1.49	148,800	-	-	-	-	3,102,000	2.72	1.49	148,800		
Mushroom	505,000	2.63	1.37	22,200	16,000	2.82	1.18	600	489,000	2.64	1.37	21,600		
South Ridge	2,005,000	2.70	1.07	68,700	-	-	-	-	2,005,000	2.70	1.07	68,700		
Double 19	1,584,000	3	1	70,400	-	-	-	-	-	=	-	-		
GH and GM	-	-	-	-	-	-	-	-	-	-	-	-		
Road Cut	225,000	3	1	6,100	-	-	-	-	225,000	2.80	0.85	6,100		
Boomerang	13,281,000	2.84	0.99	424,300	3,722,000	2.83	1.20	144,100	9,559,000	2.85	0.91	280,200		
Twin Zone	358,000	2.81	0.89	10,300	31,000	2.65	0.68	700	327,000	2.83	0.91	9,600		
	27,532,000	2.79	1.20	1,058,200	3,769,000	2.83	1.20	145,400	22,179,000	2.79	1.18	842,400		
		Inferi	red			Inferre	ed			Inferr	ed			
	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs		
Big Bend	1,257,000	2.82	1.03	41,400	-	-	-	-	1,257,000	2.82	1.03	41,400		
East Dyke	1,128,000	2.84	1.19	43,300	-	-	-	-	1,128,000	2.84	1.19	43,300		
Mushroom	-	-	-	-	-	-	-	-	-	-	-	-		
South Ridge	943,000	2.82	1.02	30,800	-	-	-	-	943,000	2.82	1.02	30,800		
Double 19	-	-	-	-	-	-	-	-	-	-	-	-		
GH and GM	2,366,000	2.76	0.79	65,200	-	-	-	-	2,366,000	2.76	0.79	65,200		
Road Cut	-	-	-	-	-	-	-	-	-	-	-	-		
Boomerang	-	-	-	-	-	-	-	-	-	-	-	-		
Twin Zone	-	-	-	-	-	-	-	-	-	-	-	-		
	5,694,000	2.80	0.99	180,700	-			-	5,694,000	2.80	0.99	180,700		

1.11 Adjacent Properties

Although the Kibi area is blanketed by mining concessions, very little systematic exploration work for bedrock gold targets has been conducted over the years in the Kibi Greenstone Belt. This reflects the fact that the Kibi area has traditionally been recognized as an alluvial gold district, and that the surrounding concessions have been held since the mid-1980s to early 1990s for their alluvial gold potential.

1.12 Other Relevant Data and Information

None.

1.13 Interpretation and Conclusions

The work undertaken has confirmed the presence of a number of auriferous bodies (i.e., Big Bend, East Dyke, Mushroom, Road Cut, South Ridge, Double 19, Gate House and Gold Mountain, Twin Zone and Boomerang) within the concession, as well as providing a structural model that explains the paragenesis of the mineralized bodies. The geological continuity has been demonstrated.

The structural information has been utilised in the mineral resource estimate completed.

The reported mineral resource is presented in Table 1.4.

<u>Cautionary Note:</u> The Mineral Resource Estimate calculation includes the area termed the "Buffer Zone", which encompasses the "open ground" between the existing Apapam mining lease boundary up to the Forest Reserve boundary. Xtra-Gold applied for this Buffer Zone in the original application of extension/renewal of the Apapam mining lease on June 17th, 2015, and also in the updated extension/renewal dated June 28th, 2019. These applications were approved by the Technical Committee of the Minerals Commission and certified to be in conformity with the official cadastral system introduced under Ghana Mineral Law. Xtra-Gold is currently awaiting formal approval of the extension/renewal. Refer to the Apapam mining lease Status Opinion prepared by REM Law Consultancy (Appendix B) for further details on the lease renewal / extension, including the Buffer Zone area.

Table 1.4: Mineral Resource Declaration - Kibi Gold Project - September 2024

Declared in terms of the CIM Standards

Cut-off: Au 0.5 g/t

	Licence	Area and Bu	Licence Area and Buffer Zone (Big Bend)								Buffer Zone(Big Bend)						Licence Area (Big Bend)			
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			
	Measured						Measured						Measured							
Oxide	Indicated	284,000	2.00	2.06	18,800	Oxide	Indicated					Oxide	Indicated	284,000	2.00	2.06	18,800			
ő	M+I	284,000	2.00	2.06	18,800	ŏ	M+I					ŏ	M+I	284,000	2.00	2.06	18,800			
	Inferred	-	-	-	=		Inferred						Inferred	-	-	-	=			
_	Measured				-	_	Measured					_	Measured				-			
sitio	Indicated	115,000	2.41	2.13	7,900	itio	Indicated					sitio	Indicated	115,000	2.41	2.13	7,900			
Transition	M+I	115,000	2.41	2.13	7,900	Transition	M+I					Transition	M+I	115,000	2.41	2.13	7,900			
-	Inferred	-	-	-	-	-	Inferred					-	Inferred	-	-	-	-			
	Measured				-		Measured						Measured				-			
lsh	Indicated	6,072,000	2.82	1.44	280,700	r S	Indicated					rs.	Indicated	6,072,000	2.82	1.44	280,700			
Fresh	M+I	6,072,000	2.82	1.44	280,700	Fresh	M+I					Fresh	M+I	6,072,000	2.82	1.44	280,700			
	Inferred	1,257,000	2.82	1.03	41,400		Inferred						Inferred	1,257,000	2.82	1.03	41,400			
	Measured				-		Measured						Measured				-			
<u>ra</u>	Indicated	6,472,000	2.78	1.48	307,400	豆	Indicated					豆	Indicated	6,472,000	2.78	1.48	307,400			
Total	M+I	6,472,000	2.78	1.48	307,400	Total	M+I					Total	M+I	6,472,000	2.78	1.48	307,400			
	Inferred	1,257,000	2.82	1.03	41,400		Inferred						Inferred	1,257,000	2.82	1.03	41,400			
																	-			
	Licence	Area and Bu	ffer Zone	(East D	yke)	•	•	Buffe	er Zone(l	East Dy	rke)			Licer	nce Area	(East Dy	/ke)			
	Licence	Area and Bu Tonnage (t)	ffer Zone Density (t/m3)	Grade (g/t)	Oyke) Content (ozs)			Buffe	er Zone(I Density (t/m3)	East Dy Grade (g/t)	cke) Content (ozs)			Licer Tonnage (t)	Density (t/m3)	(East Dy Grade (g/t)	/ke) Content (ozs)			
	Licence Measured	Tonnage	Density	Grade	Content		Measured	Tonnage	Density	Grade	Content		Measured	Tonnage	Density	Grade	Content			
qe	1	Tonnage	Density	Grade	Content	ge	Measured Indicated	Tonnage	Density	Grade	Content	de	Measured Indicated	Tonnage	Density	Grade	Content			
Oxide	Measured	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)	Oxide		Tonnage	Density	Grade	Content	Oxide		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			
Oxide	Measured Indicated	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)	Oxide	Indicated	Tonnage	Density	Grade	Content	Oxide	Indicated	Tonnage (t) 269,000	Density (t/m3)	Grade (g/t)	Content (ozs)			
	Measured Indicated M+I Inferred	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)		Indicated M+I	Tonnage	Density	Grade	Content		Indicated M+I	Tonnage (t) 269,000	Density (t/m3)	Grade (g/t)	Content (ozs)			
	Measured Indicated M+I Inferred	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)	ion	Indicated M+I Inferred Measured	Tonnage	Density	Grade	Content	ion	Indicated M+I Inferred	Tonnage (t) 269,000	Density (t/m3)	Grade (g/t)	Content (ozs)			
	Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58 1.58	Content (ozs) 13,700 13,700	ion	Indicated M+I Inferred Measured	Tonnage	Density	Grade	Content	ion	M+I Inferred Measured	Tonnage (t) 269,000 269,000	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58 1.58	Content (ozs) 13,700 13,700			
Transition Oxide	Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - 92,000	2.17 2.17 2.45	1.58 1.58	Content (ozs) 13,700 13,700 - 4,300		Indicated M+I Inferred Measured	Tonnage	Density	Grade	Content		Indicated M+I Inferred Measured Indicated	Tonnage (t) 269,000 269,000 - - 92,000	Density (t/m3) 2.17 2.17 - 2.45	Grade (g/t) 1.58 1.58 - 1.46	13,700 13,700 - 4,300			
	Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 269,000 269,000 - 92,000	2.17 2.17 2.45 2.45	1.58 1.58	Content (ozs) 13,700 13,700 - 4,300	ion	Indicated M+I Inferred Measured Indicated M+I	Tonnage	Density	Grade	Content	ion	Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 269,000 269,000 - - 92,000	Density (t/m3) 2.17 2.17 - 2.45 2.45	Grade (g/t) 1.58 1.58 - 1.46	13,700 13,700 - 4,300			
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - 92,000	2.17 2.17 2.45 2.45	1.58 1.58	Content (ozs) 13,700 13,700 - 4,300	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage	Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - - 92,000	Density (t/m3) 2.17 2.17 - 2.45 2.45	Grade (g/t) 1.58 1.58 - 1.46	13,700 13,700 - 4,300			
	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 269,000 269,000 - 92,000 92,000	2.17 2.17 2.45 2.45	1.58 1.58 1.46 1.46	Content (ozs) 13,700 13,700 - 4,300 4,300	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage	Density	Grade	Content	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 269,000 269,000 - 92,000 92,000 -	Density (t/m3) 2.17 2.17 - 2.45 2.45	Grade (g/t) 1.58 1.58 - 1.46 1.46 1.48	Content (ozs) 13,700 13,700 - 4,300 4,300			
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Tonnage (t) 269,000 269,000 92,000 92,000 - 2,742,000	2.17 2.17 2.45 2.45 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48	Content (ozs) 13,700 13,700 4,300 4,300 - 130,900	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	Tonnage	Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 269,000 269,000 92,000 92,000 - 2,742,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48	Content (ozs) 13,700 13,700 - 4,300 4,300 - 130,900			
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Measured Indicated M+I	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000	2.17 2.17 2.45 2.45 2.79 2.79	1.58 1.58 1.46 1.46 1.48	Content (ozs) 13,700 13,700 - 4,300 4,300 - 130,900 130,900	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Measured Indicated M+I	Tonnage	Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Measured Indicated M+I	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 2.79 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48 1.48	Content (ozs) 13,700 13,700 - 4,300 4,300 - 130,900 130,900			
Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000	2.17 2.17 2.45 2.45 2.79 2.79	1.58 1.58 1.46 1.46 1.48	Content (ozs) 13,700 13,700 - 4,300 4,300 - 130,900 130,900	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated	Tonnage	Density	Grade	Content	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 2.79 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48 1.48	Content (ozs) 13,700 13,700 - 4,300 4,300 - 130,900 130,900			
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 1,128,000 -	2.17 2.17 2.45 2.45 2.79 2.84	1.58 1.58 1.46 1.46 1.48 1.19	Content (ozs) 13,700 13,700 - 4,300 4,300 - 130,900 130,900 43,300	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage	Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000 1,128,000	2.17 2.17 2.45 2.45 2.79 2.79 2.84	1.58 1.58 1.46 1.46 1.48 1.49	Content (ozs) 13,700 13,700 - 4,300 4,300 - 130,900 130,900 43,300			

	Licence A	rea and But	fer Zone	(Mushr	oom)			Buffe	er Zone(l	Mushro	om)			Licer	ice Area	(Mushro	oom)
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
de	Indicated	103,000	2.04	1.50	4,900	ge	Indicated					9	Indicated	103,000	2.04	1.50	4,900
Oxide	M+I	103,000	2.04	1.50	4,900	Oxide	M+I					Oxide	M+I	103,000	2.04	1.50	4,900
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
_	Measured	-			-	_	Measured					_	Measured	-			-
itior	Indicated	17,000	2.05	1.11	600	itio	Indicated					iţi	Indicated	17,000	2.05	1.11	600
Transition	M+I	17,000	2.05	1.11	600	Transition	M+I					Transition	M+I	17,000	2.05	1.11	600
F	Inferred	-	-	-	-	=	Inferred					=	Inferred	-	-	-	-
	Measured	-			-		Measured						Measured	-			-
sh	Indicated	385,000	2.81	1.35	16,600	S.	Indicated	16,000	2.82	1.18	600	rs.	Indicated	369,000	2.81	1.35	16,100
Fresh	M+I	385,000	2.81	1.35	16,600	Fresh	M+I	16,000	2.82	1.18	600	F.	M+I	369,000	2.81	1.35	16,100
	Inferred	-	-	-	-		Inferred	-	-	-	-		Inferred	-	-	-	-
	Measured	-			=		Measured	-			-		Measured	-			-
al	Indicated	505,000	2.63	1.37	22,200	<u>=</u>	Indicated	16,000	2.82	1.18	600	<u>=</u>	Indicated	489,000	2.64	1.37	21,600
Total	M+I	505,000	2.63	1.37	22,200	Total	M+I	16,000	2.82	1.18	600	Total	M+I	489,000	2.64	1.37	21,600
	Inferred	-	-	-	-		Inferred	-	-	-	-		Inferred	-	-	-	-
	Licence Ar	ea and Buff	er Zone	(South F	Ridge)			Buffe	r Zone(S	outh Ri	dge)			Licen	ce Area (South R	idge)
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
e P	Indicated	423,000	2.35	1.15	15,600	e	Indicated					9	Indicated	423,000	2.35	1.15	15,600
Oxide	M+I	423,000	2.35	1.15	15,600	Oxide	M+I					Oxide	M+I	423,000	2.35	1.15	15,600
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
_	Measured	-			-	_	Measured					_	Measured	-			-
sition	Indicated	181,000	2.68	1.16	6,700	sition	Indicated					sition	Indicated	181,000	2.68	1.16	6,700
Transi	M+I	181,000	2.68	1.16	6,700	Transi	M+I					Transi	M+I	181,000	2.68	1.16	6,700
Ė	Inferred	-	-	-	-	=	Inferred					=	Inferred	-	-	-	-
	Measured	-			-		Measured						Measured	-			-
ų,	Indicated	1,402,000	2.81	1.03	46,400	چ	Indicated					S.	Indicated	1,402,000	2.81	1.03	46,400
Fresh	M+I	1,402,000	2.81	1.03	46,400	Fresh	M+I					Fres	M+I	1,402,000	2.81	1.03	46,400
	Inferred	943,000	2.82	1.02	30,800		Inferred						Inferred	943,000	2.82	1.02	30,800
	Measured	-			-		Measured						Measured	-			/
<u></u>	Indicated	2,005,000	2.70	1.07	68,700		Indicated					豆	Indicated	2,005,000	2.70	1.07	68,700
Total	M+I	2,005,000	2.70	1.07	68,700	Total	M+I					Tota	M+I	2,005,000	2.70	1.07	68,700
		943,000	2.82	1.02	30,800									943,000	2.82	1.02	30,800

	Licence /	Area and Bu	ffer Zone	(Doubl	e 19)			Buff	er Zone	(Double	19)			Licer	nce Area	(Double	19)
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
ge	Indicated	152,000	2.11	1.68	8,200	e e	Indicated					de	Indicated	152,000	2.11	1.68	8,200
Oxide	M+I	152,000	2.11	1.68	8,200	Oxide	M+I					Oxide	M+I	152,000	2.11	1.68	8,200
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
_	Measured	-			-	_	Measured						Measured	-			-
itio	Indicated	89,000	2.11	1.67	4,800	iţi	Indicated					itio	Indicated	89,000	2.11	1.67	4,800
Transition	M+I	89,000	2.11	1.67	4,800	Transition	M+I					Transition	M+I	89,000	2.11	1.67	4,800
-	Inferred	-	-	-	-	_	Inferred					_	Inferred	-	-	-	-
	Measured	-			-		Measured						Measured	-			-
Fresh	Indicated	1,343,000	2.71	1.33	57,400	Fresh	Indicated					Fresh	Indicated	1,343,000	2.71	1.33	57,400
F	M+I	1,343,000	2.71	1.33	57,400	F.	M+I					표	M+I	1,343,000	2.71	1.33	57,400
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
	Measured	-			-		Measured						Measured	-			-
Total	Indicated	1,584,000	2.62	1.38	70,400	Total	Indicated					Total	Indicated	1,584,000	2.62	1.38	70,400
₽	M+I	1,584,000	2.62	1.38	70,400	6	M+I					6	M+I	1,584,000	2.62	1.38	70,400
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
	Licence Ar	I D ((-															
	LIOCHIOC AI	ea and Buffe and Gate	er Zone (e House)		ountain			Buffer Z	one(Gold Gate H		tain and			Licence /	Area (Gol Gate Ho		tain and
	Elociloc Al				Content (ozs)			Buffer Zo			Content (ozs)			Tonnage (t)			Content (ozs)
	Measured	and Gate Tonnage	e House) Density	Grade	Content		Measured	Tonnage	Gate H Density	ouse) Grade	Content		Measured	Tonnage	Gate Ho	ouse) Grade	Content
epi		and Gate Tonnage	e House) Density	Grade	Content	e Di		Tonnage	Gate H Density	ouse) Grade	Content	epi	Measured Indicated	Tonnage	Gate Ho	ouse) Grade	Content
Oxide	Measured	and Gate Tonnage	e House) Density	Grade	Content	Oxide		Tonnage	Gate H Density	ouse) Grade	Content	Oxide		Tonnage	Gate Ho	ouse) Grade	Content
Oxide	Measured Indicated	and Gate Tonnage	e House) Density	Grade	Content	Oxide		Tonnage	Gate H Density	ouse) Grade	Content	Oxide	Indicated	Tonnage	Gate Ho	ouse) Grade	Content
	Measured Indicated M+I	and Gate Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)		Indicated M+I Inferred Measured	Tonnage	Gate H Density	ouse) Grade	Content		Indicated M+I	Tonnage (t)	Gate Ho Density (t/m3)	Ouse) Grade (g/t)	Content (ozs)
	Measured Indicated M+I Inferred	and Gate Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)		Indicated M+I Inferred Measured	Tonnage	Gate H Density	ouse) Grade	Content		Indicated M+I Inferred	Tonnage (t)	Gate Ho Density (t/m3)	Ouse) Grade (g/t)	Content (ozs)
sition	Measured Indicated M+I Inferred Measured	and Gate Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)	sition	Indicated M+I Inferred Measured Indicated	Tonnage	Gate H Density	ouse) Grade	Content	sition	Indicated M+I Inferred Measured	Tonnage (t)	Gate Ho Density (t/m3)	Ouse) Grade (g/t)	Content (ozs)
	Measured Indicated M+I Inferred Measured Indicated	and Gate Tonnage (t) 9,000	Density (t/m3)	Grade (g/t)	Content (ozs)		Indicated M+I Inferred Measured Indicated	Tonnage	Gate H Density	ouse) Grade	Content		M+I Inferred Measured Indicated	Tonnage (t)	Gate Ho Density (t/m3)	Ouse) Grade (g/t)	Content (ozs)
sition	Measured Indicated M+I Inferred Measured Indicated M+I	9,000	Density (t/m3)	0.70	200 -	sition	Indicated M+I Inferred Measured Indicated M+I	Tonnage	Gate H Density	ouse) Grade	Content	sition	Indicated M+I Inferred Measured Indicated M+I	9,000 -	Gate Ho Density (t/m3)	Grade (g/t)	200
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred	9,000	Density (t/m3)	0.70	200 -	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage	Gate H Density	ouse) Grade	Content	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred	9,000 - - 192,000	Gate Ho Density (t/m3)	Grade (g/t)	200 -
sition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	9,000	Density (t/m3)	0.70	200 -	sition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage	Gate H Density	ouse) Grade	Content	sition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	9,000 - - 192,000	Gate Ho Density (t/m3)	Grade (g/t)	200 -
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	9,000 - - 192,000	1.50 - 2.40	0.70	200 -	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage	Gate H Density	ouse) Grade	Content	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	9,000 - - 192,000	Gate Ho Density (t/m3)	Grade (g/t)	200 -
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	9,000 - - 192,000	1.50	0.70 - 0.73	200 - - - 4,500	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage	Gate H Density	ouse) Grade	Content	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	9,000 - - - 192,000	1.50 - 2.40	0.70 0.73	200 - - - 4,500
Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated	9,000 - - 192,000	1.50	0.70 - 0.73	200 - - - 4,500	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured MHI Inferred	Tonnage	Gate H Density	ouse) Grade	Content	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated	9,000 - - - 192,000	1.50 - 2.40	0.70 0.73	200 - - - 4,500
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	9,000 - - 192,000	1.50 	0.70 - 0.73	200 - - - 4,500	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured MHI Inferred	Tonnage	Gate H Density	ouse) Grade	Content	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	9,000	1.50 - 2.40	0.70 0.73	200 - - - 4,500

	Licence	Area and Bu	ıffer Zon	e (Road	Cut)			Buff	er Zone((Road C	ut)			Lice	nce Area	(Road C	ut)
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
qe	Indicated					ge	Indicated					g	Indicated				
Oxide	M+I					Oxide	M+I					Oxide	M+I				
	Inferred						Inferred						Inferred				
u	Measured					_	Measured					_	Measured				
itio	Indicated	12,000	2.50	0.74	300	iţi	Indicated					iţi	Indicated	12,000	2.50	0.74	300
Transition	M+I	12,000	2.50	0.74	300	Transition	M+I					Transition	M+I	12,000	2.50	0.74	300
_	Inferred	-	-	-	-	-	Inferred					-	Inferred	-	-	-	-
	Measured	-	-	-	-		Measured						Measured	-	-	-	-
sh	Indicated	213,000	2.82	0.85	5,800	S.	Indicated					- Ly	Indicated	213,000	2.82	0.85	5,800
Fresh	M+I	213,000	2.82	0.85	5,800	Fresh	M+I					Fresh	M+I	213,000	2.82	0.85	5,800
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
	Measured	-	-	-	-		Measured						Measured	-	-	-	-
		225,000	2.80	0.85	6,100	-	Indicated					Total	Indicated	225,000	2.80	0.85	6,100
[a]	Indicated	225,000										ō					
Total	Indicated M+I	225,000	2.80	0.85	6,100	Total	M+I					_	M+I	225,000	2.80	0.85	6,100
Total		•		0.85	6,100	Tot	M+I Inferred						M+I Inferred	225,000	2.80	0.85	6,100
Total	M+I Inferred	225,000	2.80	-	-	Tot		Buffe	r Zone (E	Boomer	ang)			-		-	-
Total	M+I Inferred	225,000	2.80	-	-	Tot		Buffe Tonnage (t)	r Zone (E	Grade (g/t)	ang) Content (ozs)			-	-	-	-
Total	M+I Inferred	225,000 - Area and Buff Tonnage	2.80 - fer Zone Density	(Boome	erang)	Tot		Tonnage	Density	Grade	Content	-		Licen	ce Area (- Boomer _{Grade}	ang)
	M+I Inferred Licence A	225,000 - Area and Buff Tonnage	2.80 - fer Zone Density	(Boome	erang)		Inferred	Tonnage	Density	Grade	Content		Inferred	Licen	ce Area (- Boomer _{Grade}	ang)
Oxide	M+I Inferred Licence A	225,000 - Area and Buff Tonnage (t)	2.80 - fer Zone Density (t/m3)	(Boome Grade (g/t)	Content (ozs)	Oxide	Inferred	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)	Oxide	Inferred Measured	Licend Tonnage (t)	ce Area (Density (t/m3)	Boomer Grade (g/t)	ang) Content (ozs)
	M+I Inferred Licence A Measured Indicated	225,000 - Area and Buff Tonnage (t) 478,000	2.80 - fer Zone Density (t/m3)	(Boome Grade (g/t)	Content (ozs)		Inferred Measured Indicated	Tonnage (t) 164,000	Density (t/m3)	Grade (g/t) 0.94	Content (ozs)		Inferred Measured Indicated	Licence Tonnage (t) 314,000	ce Area (Density (t/m3)	Boomer Grade (g/t) 1.05	ang) Content (ozs)
Oxide	M+I Inferred Licence A Measured Indicated M+I	225,000 - Area and Buff Tonnage (t) 478,000	2.80 - fer Zone Density (t/m3)	(Boome Grade (g/t)	Content (ozs)	Oxide	Inferred Measured Indicated M+I	Tonnage (t) 164,000	Density (t/m3) 1.87	Grade (g/t) 0.94 0.94	Content (ozs)	Oxide	Inferred Measured Indicated M+I	Licence Tonnage (t) 314,000	ce Area (Density (t/m3)	- Boomer Grade (g/t) 1.05	ang) Content (ozs)
Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred	225,000 	2.80 - fer Zone Density (t/m3)	(Boome Grade (g/t)	Content (ozs)	Oxide	Measured Indicated M+I Inferred	Tonnage (t) 164,000 164,000	Density (t/m3) 1.87 1.87	Grade (g/t) 0.94 0.94	Content (ozs)	Oxide	Measured Indicated M+I Inferred	Tonnage (t) 314,000	Density (t/m3) 2.01 2.01	- Boomer Grade (g/t) 1.05	ang) Content (ozs)
sition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured	225,000 - Area and Buff Tonnage (t) 478,000	2.80 - fer Zone Density (t/m3) 1.96 1.96	Grade (g/t) 1.01 1.01	Content (ozs) 15,500 15,500	sition Oxide	Measured Indicated M+I Inferred Measured	Tonnage (t) 164,000 164,000 -	Density (t/m3) 1.87 1.87	Grade (g/t) 0.94 0.94	5,000 5,000	Isition Oxide	Measured Indicated M+I Inferred Measured	Tonnage (t) 314,000	Density (t/m3) 2.01 2.01	- Boomer Grade (g/t) 1.05 1.05	Content (ozs) 10,600 10,600
Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated	225,000 - Area and Buff Tonnage (t) 478,000 242,000	2.80 - fer Zone Density (t/m3) 1.96 1.96 - 2.35	Grade (g/t) 1.01 1.01 - 1.02	Content (ozs) 15,500 15,900	Oxide	Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 164,000 164,000 - 79,000	Density (t/m3) 1.87 1.87 - 2.36	0.94 0.94 - 0.98	Content (ozs) 5,000 5,000 - 2,500	Oxide	Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 314,000	2.01 2.35	Grade (g/t) 1.05 1.05 - 1.04	ang) Content (ozs) 10,600 10,600 - 5,400
sition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I	225,000 - Area and Buff Tonnage (t) 478,000	2.80	Grade (g/t) 1.01 1.01 - 1.02	Content (ozs) 15,500 15,900	sition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 164,000 164,000 - 79,000	Density (t/m3) 1.87 1.87 - 2.36 2.36	Grade (g/t) 0.94 0.94 - 0.98 0.98	Content (ozs) 5,000 5,000 - 2,500	Isition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 314,000	2.01 2.35	Grade (g/t) 1.05 1.05 - 1.04	ang) Content (ozs) 10,600 10,600 - 5,400
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated	225,000 - Area and Buff Tonnage (t) 478,000 - 242,000 242,000	2.80	(Boome Grade (g/t) 1.01 1.01 - - 1.02 1.02	Content (ozs) 15,500 15,900	Transition Oxide	Measured Indicated M+I Inferred Indicated Indicated Indicated Indicated Indicated Indicated Inferred Inferred Inferred Inferred Inferred	Tonnage (t) 164,000 164,000 - 79,000 79,000	Density (t/m3) 1.87 1.87 - 2.36 2.36	Grade (g/t) 0.94 0.94 0.98 0.98	Content (ozs) 5,000 5,000 - 2,500	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 314,000 314,000 - 162,000	2.01 2.35 2.35	1.05 1.05 1.04	ang) Content (ozs) 10,600 10,600 - 5,400
sition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured MHI Inferred Measured	225,000 - Area and Buff Tonnage (t) 478,000 - 242,000 242,000	2.80 fer Zone Density (t/m3) 1.96 2.35	1.01 1.01 1.02 1.02	- Content (ozs) 15,500 15,500 7,900 7,900 -	sition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured M+I Inferred Measured	Tonnage (t) 164,000 164,000 79,000 79,000	Density (t/m3) 1.87 1.87 2.36 2.36	Grade (g/t) 0.94 0.94 - 0.98 0.98	5,000 5,000 2,500 2,500	Isition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured M+I Inferred Measured Measured	Tonnage (t) 314,000 314,000	2.01 2.01 2.35 2.35	1.05 1.05 1.04 1.04	- ang) Content (ozs) 10,600 10,600 - 5,400 5,400
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Inferred Inferred Inferred Inferred Inferred	225,000 - Area and Buff Tonnage (t) 478,000 242,000 242,000 - 12,561,000	2.80	1.01 1.02 1.02 - 0.99	7,900 7,900 400,800	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Miniferred Measured Indicated	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000	Density (t/m3) 1.87 1.87 2.36 2.36 2.88	Grade (g/t) 0.94 0.94 - 0.98 0.98 - 1.22	Content (ozs) 5,000 5,000 - 2,500 2,500 - 136,600	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Measured Inferred	Tonnage (t) 314,000 314,000	2.01 2.01 2.35 2.35 2.89	1.05 1.05 	- ang) Content (ozs) 10,600 10,600 - 5,400 5,400 - 264,200
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	225,000 - Area and Buff Tonnage (t) 478,000 478,000 - 242,000 242,000 - 12,561,000 12,561,000	2.80 fer Zone Density (t/m3) 1.96 1.96 2.35 2.35 2.89 2.89	(Boome (g/t) 1.01 1.01 1.02 1.02 1.02 1.09 0.99	7,900 7,900 400,800	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Measured Indicated M+I	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000 3,478,000	Density (t/m3) 1.87 1.87 - 2.36 2.36 2.88 2.88	Grade (g/t) 0.94 0.94 - 0.98 0.98 - 1.22 1.22	Content (ozs) 5,000 5,000 - 2,500 2,500 - 136,600	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 314,000 314,000 - 162,000 - 9,083,000 9,083,000	2.01 2.01 2.35 2.89 2.89	1.05 1.04 1.04 1.04 0.90	- ang) Content (ozs) 10,600 - 5,400 5,400 - 264,200
Fresh Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Measured Inferred Inferred Inferred Indicated	225,000	2.80	1.01 1.02 1.02 - 0.99 0.99	7,900 7,900 400,800	Fresh Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated M+I Inferred	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000 3,478,000	Density (t/m3) 1.87 1.87 - 2.36 2.36 - 2.88 2.88	Grade (g/t) 0.94 0.94 0.98 0.98 - 1.22 1.22	Content (ozs) 5,000 5,000 - 2,500 2,500 - 136,600	Fresh Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated Indicated Indicated	Tonnage (t) 314,000 314,000 162,000 162,000 9,083,000 9,083,000	2.01 2.01 2.35 2.35 2.89 2.89	1.05 1.05 1.04 1.04 0.90	- ang) Content (ozs) 10,600 - 5,400 5,400 - 264,200
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred	225,000 - Area and Buff Tonnage (t) 478,000 478,000 - 242,000 242,000 - 12,561,000 12,561,000	2.80 fer Zone Density (t/m3) 1.96 2.35 2.35 2.89 2.89	(Boome Grade (g/t) 1.01 1.01 - 1.02 1.02 - 0.99 0.99	7,900 7,900 400,800	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000 3,478,000	Density (t/m3) 1.87 1.87 2.36 2.36 2.88 2.88	Grade (g/t) 0.94 0.94 - 0.98 0.98 - 1.22 1.22 - 1.20	Content (ozs) 5,000 5,000 2,500 2,500 136,600 136,600	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Licence (t) 314,000 314,000	2.01 2.01 2.35 2.89 2.89	1.05 1.05 1.04 1.04 1.04 1.04	- ang) Content (ozs) 10,600 10,600 - 5,400 5,400 - 264,200 264,200

	Licence .	Area and Bu	ffer Zone	(Twin	Zone)			Buffe	r Zone (1	Γwin Zo	ne)			Licen	ce Area (Twin Zo	ne)
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
Oxide	Indicated	47,000	2.41	0.65	1,000	200	Indicated	7,000	1.86	0.54	100	Oxide	Indicated	39,000	2.51	0.67	800
ŏ	M+I	47,000	2.41	0.65	1,000	2	M+I	7,000	1.86	0.54	100	ő	M+I	39,000	2.51	0.67	800
	Inferred	-	-	-	-		Inferred	-	-	-	-		Inferred	-	-	-	1
_	Measured	-	-	-	-	2	Measured	-	-	-	-	_	Measured	-	-	-	1
ansition	Indicated	2,000	2.43	0.74	100	9	Indicated	60	2.75	0.54	1	ansition	Indicated	2,000	2.43	0.74	100
rans	M+I	2,000	2.43	0.74	100		M+I	60	2.75	0.54	1	rans	M+I	2,000	2.43	0.74	100
-	Inferred	-	-	-	-		Inferred	-	-	-	-		Inferred	-	-	-	1
	Measured	-	-	-	-		Measured	-	-	-	-		Measured	-	-	-	
Fresh	Indicated	309,000	2.88	0.93	9,200	2 0 0 1	Indicated	23,000	2.90	0.72	500	Fresh	Indicated	286,000	2.88	0.95	8,700
Fre	M+I	309,000	2.88	0.93	9,200	ثُ ا	M+I	23,000	2.90	0.72	500	Fre	M+I	286,000	2.88	0.95	8,700
	Inferred	-	-	-	-		Inferred	-	-	-	=		Inferred	-	-	-	-
	Measured	-	-	-	-		Measured	-	-	-	-		Measured	-	-	-	
Total	Indicated	358,000	2.81	0.89	10,300	40	Indicated	31,000	2.65	0.68	700	Total	Indicated	327,000	2.83	0.91	9,600
₽	M+I	358,000	2.81	0.89	10,300	5	M+I	31,000	2.65	0.68	700	우	M+I	327,000	2.83	0.91	9,600
	Inferred	-	-	-	-		Inferred	-	-	-	-		Inferred	-	-	-	1
	Lic	Licence Area and Buffer Zone						Buffer	Zone					Licence	e Area		
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content			Tonnage	Density	Grade	Content						
	Measured		` '		(ozs)			(t)	(t/m3)	(g/t)	(ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
Oxide			, ,		(ozs)		Measured	(t)	(t/m3)	(g/t)			Measured				
.=	Indicated	1,755,000	2.12	1.38	77,700	e p	Measured Indicated	(t) 172,000	(t/m3)	(g/t) 0.92		e de					
ô	Indicated M+I	1,755,000 1,755,000	2.12 2.12			Oxide			, ,		(ozs)	Oxide		(t) _	(t/m3)	(g/t)	(ozs)
ô				1.38	77,700	Oxide	Indicated	172,000	1.87	0.92	(ozs) 5,100	Oxide		(t) 1,583,000	(t/m3) 2.15	(g/t) 1.43	(ozs) 72,600
	M+I	1,755,000	2.12	1.38 1.38	77,700 77,700		Indicated M+I	172,000 172,000	1.87	0.92 0.92	(ozs) 5,100		Indicated M+I Inferred Measured	(t) 1,583,000 1,583,000	(t/m3) 2.15 2.15	(g/t) 1.43 1.43	72,600 72,600
	M+I Inferred	1,755,000	2.12	1.38 1.38 0.70	77,700 77,700 200		Indicated M+I Inferred	172,000 172,000	1.87	0.92 0.92	5,100 5,100		Indicated M+I Inferred Measured	(t) 1,583,000 1,583,000	2.15 2.15 1.50	(g/t) 1.43 1.43	72,600 72,600
sition	M+I Inferred Measured	1,755,000 9,000	2.12 1.50	1.38 1.38 0.70	77,700 77,700 200	sition	Indicated M+I Inferred Measured	172,000 172,000 -	1.87 1.87	0.92	5,100 5,100 -	sition	Indicated M+I Inferred Measured Indicated	(t) 1,583,000 1,583,000 9,000	2.15 2.15 1.50	1.43 1.43 0.70	72,600 72,600 200
	M+I Inferred Measured Indicated	1,755,000 9,000 - 750,000	2.12 1.50 - 2.42	1.38 1.38 0.70 - 1.35	77,700 77,700 200 - 32,600		Indicated M+I Inferred Measured Indicated	172,000 172,000 - - 79,000	1.87 1.87 - - 2.36	0.92 0.92 - 0.98	5,100 5,100 - 2,500		Indicated M+I Inferred Measured Indicated	(t) 1,583,000 1,583,000 9,000 - 670,000	2.15 2.15 1.50 - 2.43	1.43 1.43 0.70 - 1.39	72,600 72,600 200 - 30,000
sition	M+I Inferred Measured Indicated M+I	1,755,000 9,000 - 750,000 750,000	2.12 1.50 - 2.42 2.42	1.38 1.38 0.70 - 1.35 1.35	77,700 77,700 200 - 32,600 32,600	sition	Indicated M+I Inferred Measured Indicated M+I	172,000 172,000 - - 79,000	1.87 1.87 - - 2.36 2.36	0.92 0.92 - 0.98	5,100 5,100 - 2,500	sition	Indicated M+I Inferred Measured Indicated M+I	(t) 1,583,000 1,583,000 9,000 - 670,000 670,000	2.15 2.15 1.50 - 2.43 2.43	1.43 1.43 0.70 - 1.39	72,600 72,600 200 - 30,000 30,000
Transition	M+I Inferred Measured Indicated M+I Inferred	1,755,000 9,000 - 750,000 750,000 192,000	2.12 1.50 - 2.42 2.42 2.40	1.38 1.38 0.70 - 1.35 1.35	77,700 77,700 200 - 32,600 32,600 4,500	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred	172,000 172,000 - - 79,000 79,000	1.87 1.87 - - 2.36 2.36	0.92 0.92 - 0.98 0.98	5,100 5,100 2,500 2,500 -	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	(t) 1,583,000 1,583,000 9,000 - 670,000 670,000 192,000	2.15 2.15 1.50 2.43 2.43 2.40	1.43 1.43 0.70 - 1.39 0.73	72,600 72,600 200 - 30,000 30,000
sition	M+I Inferred Measured Indicated M+I Inferred Measured	1,755,000 9,000 - 750,000 750,000 192,000	2.12 1.50 - 2.42 2.42 2.40	1.38 1.38 0.70 - 1.35 1.35	77,700 77,700 200 - 32,600 32,600 4,500	sition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	172,000 172,000 - - 79,000 79,000	1.87 1.87 - - 2.36 2.36	0.92 0.92 - 0.98 0.98	5,100 5,100 - 2,500 2,500 - 137,700	sition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	(t) 1,583,000 1,583,000 9,000 - 670,000 670,000 192,000	2.15 2.15 1.50 - 2.43 2.43	1.43 1.43 0.70 - 1.39 1.39	72,600 72,600 200 - 30,000 30,000 4,500
Transition	M+I Inferred Measured Indicated M+I Inferred Measured Indicated	1,755,000 9,000 - 750,000 750,000 192,000 - 25,027,000	2.12 1.50 - 2.42 2.42 2.40 - 2.84	1.38 1.38 0.70 - 1.35 1.35 0.73 - 1.18	77,700 77,700 200 - 32,600 32,600 4,500 - 947,900	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	172,000 172,000 - - 79,000 79,000 - - 3,517,000	1.87 1.87 - 2.36 2.36 - 2.88	0.92 0.92 - 0.98 0.98 - 1.221	5,100 5,100 - 2,500 2,500 - 137,700	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	(t) 1,583,000 1,583,000 9,000 - 670,000 670,000 192,000 - 21,510,000	2.15 2.15 1.50 - 2.43 2.43 2.40 - 2.84	1.43 1.43 0.70 - 1.39 0.73 - 1.17	72,600 72,600 200 - 30,000 4,500 - 810,200
Transition	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	1,755,000 9,000 - 750,000 750,000 192,000 - 25,027,000 25,027,000	2.12 1.50 - 2.42 2.42 2.40 - 2.84 2.84	1.38 1.38 0.70 - 1.35 1.35 0.73 - 1.18 1.18	77,700 77,700 200 - 32,600 32,600 4,500 - 947,900 947,900	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	172,000 172,000 - 79,000 79,000 - 3,517,000 3,517,000	1.87 1.87 - 2.36 2.36 - 2.88 2.88	0.92 0.92 - 0.98 0.98 - 1.221 1.22	5,100 5,100 - 2,500 2,500 - 137,700	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	(t) 1,583,000 1,583,000 9,000 - 670,000 192,000 - 21,510,000 21,510,000	2.15 2.15 1.50 - 2.43 2.43 2.40 - 2.84 2.84	1.43 1.43 0.70 - 1.39 1.39 0.73 - 1.17	72,600 72,600 200 - 30,000 30,000 4,500 - 810,200 810,200
Fresh Transition	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	1,755,000 9,000 - 750,000 192,000 - 25,027,000 25,027,000 5,706,000	2.12 1.50 - 2.42 2.42 2.40 - 2.84 2.84 2.71	1.38 1.38 0.70 - 1.35 1.35 0.73 - 1.18 1.00	77,700 77,700 200 - 32,600 32,600 4,500 - 947,900 947,900 176,100	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated M+I Inferred	172,000 172,000 79,000 79,000 - 3,517,000 3,517,000	1.87 1.87 - 2.36 2.36 - 2.88 2.88	0.92 0.92 - 0.98 0.98 - 1.221 1.22	(ozs) 5,100 5,100 - 2,500 2,500 - 137,700 137,700	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Measured M+I Inferred	(t) 1,583,000 1,583,000 9,000 - 670,000 192,000 - 21,510,000 21,510,000 5,493,000	2.15 2.15 1.50 2.43 2.43 2.40 - 2.84 2.84	1.43 1.43 0.70 - 1.39 1.39 0.73 - 1.17 1.10	72,600 72,600 200 - 30,000 30,000 4,500 - 810,200 810,200
Transition	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured MHI Inferred Measured	1,755,000 9,000 - 750,000 750,000 192,000 - 25,027,000 25,027,000 5,706,000	2.12 1.50 - 2.42 2.42 2.40 - 2.84 2.84	1.38 1.38 0.70 - 1.35 1.35 0.73 - 1.18 1.18 1.20	77,700 77,700 200 32,600 32,600 4,500 - 947,900 947,900 176,100	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	172,000 172,000 79,000 79,000 - 3,517,000 3,517,000	1.87 1.87 - 2.36 2.36 - 2.88 2.88	0.92 0.92 - 0.98 0.98 - 1.221 1.22 - 1.20	(ozs) 5,100 5,100 - 2,500 2,500 - 137,700 137,700	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Measured M+I Inferred	(t) 1,583,000 1,583,000 9,000 - 670,000 192,000 - 21,510,000 21,510,000 5,493,000 -	2.15 2.15 1.50 - 2.43 2.43 2.40 - 2.84 2.84	1.43 1.43 0.70 - 1.39 1.39 0.73 - 1.17 1.00	72,600 72,600 200 - 30,000 4,500 - 810,200 810,200 - 176,100

A substantial amount of work has been completed on the Cobra Creek prospect where various targets have also been identified.

1.14 Recommendations

Based on the results of the 2024 Mineral Resource Estimate (MRE) and exploration results on early-stage targets across the project area, Pivot and TECT recommend a two-phase exploration program to further advance the Kibi Gold Project.

Phase 1 is aimed of advancing the Kibi Gold Project by further delineation of existing mineral resources and identification of additional resource bodies within the MRE footprint area, continued advancement of early-stage targets across the Apapam concession, and property-scale target generation exploration work.

Phase 2, designed to support the continued advancement of the project, includes additional drilling to further define mineral resources, an updated mineral resource estimate, completion of a Preliminary Economic Assessment (PEA), metallurgical test work, and collection of additional data to support future scoping studies. With the implementation of Phase 2 being contingent upon the success of Phase 1.

A cost estimate for the recommended two-phase work program serves as a guideline. The estimated drilling expenditures are based on all-inclusive drilling costs utilizing Xtra-Gold's inhouse operated diamond core drill rigs. Total expenditures are estimated at USD 5,185,000, including: USD 3,585,000 for Phase 1; and USD 1,600,000 for Phase 2. With the implementation of Phase 2 being contingent upon the success of Phase 1.

2 INTRODUCTION

Xtra-Gold Resources Corp. (Xtra-Gold) commissioned TECT Geological Consulting Ltd (TECT) and Pivot Mining Consultants Ltd. (Pivot) to prepare an updated Mineral Resource Estimate for the Apapam Concession (LVB 5191/09). This Independent Technical Report (ITR) has been prepared in accordance with disclosure and reporting guidelines set forth in National Instrument 43-101 (NI 43-101) and companion Form 43-101F1 of the Canadian Securities Administrators' Standards of Disclosure for Mineral Projects. The Apapam Concession forms part of Xtra-Gold's Kibi Gold Project in southern Ghana.

2.1 Scope of the Report

The purpose of this report is to publish an Independent Technical Report summarizing the geology, past exploration activities and Mineral Resource Estimate on the Kibi Gold Project. With the updated mineral resource estimate based on drilling completed from July 2021 to January 2024. Xtra-Gold Resources Corp.

Xtra-Gold's corporate offices are located at Village Road Shopping Plaza, Suite 2150, P.O. Box AP 59217, Nassau, Bahamas. Xtra-Gold is a public company listed on the Toronto Stock Exchange (TSX: XTG), with the company also trading on the over-the-counter equity market in the United States (OTCQB: XTGRF). Xtra-Gold is a junior exploration company focused on the advancement and development of its Kibi Gold Project in southern Ghana.

2.2 Principal Sources of Information

The data for the areas of gold mineralisation on the Kibi Gold Project and the geology of southern Ghana was obtained from Xtra-Gold. TECT and Pivot reviewed all of the available historical and current exploration work data and consider the data to be reliable.

The close-out date of the mineral resource database is January 29, 2024.

2.3 Participants, Qualifications, and Experience

The participants in the team consist of technical experts brought together by TECT and Pivot to estimate the Mineral Resources who are Competent Persons as defined in the Canadian Institute of Mining, Metallurgy and Petroleum (CIM) guidelines. The participants in the review and their individual areas of responsibility are listed as follows:-

Dr Corné Koegelenberg - Structural-Economic Geologist

Ph.D. Structural Geology and Tectonics

M.Sc. Magmatic Sulphide Petrology and Ni-Cu-PGE mineralization

MGSSA, MSEG, Pr.Sci.Nat.

Dr Koegelenberg has 12 years exploration and consulting experience and has worked on more than 68 projects with experience in mapping, deciphering and 3D modelling of structurally complex, low- and high-grade, fold-and-thrust belt metamorphic terranes. He specializes in dynamic and kinematic structural analysis of Au-ferrous quartz-carbonate vein systems, high-

grade metamorphic shear-hosted Cu-Co mineralization systems, whilst also having an academic background in magmatic sulphide and Ni-Cu-PGE mineralization processes in mafic to ultramafic complexes. He has extensive experience in advanced Micromine and Leapfrog Geo 3D modelling of structural networks, lithotypes, alteration and mineralization, combined with lithological and structural interpretation of geophysical and LandSat/ASTER data sets. He has published 8 peer-reviewed papers to date.

Dr Ian Basson - Structural-Economic Geologist

Ph.D. Structural Geology FGSSA, MSEG, AMSAIEG, Pr.Sci.Nat.

Dr Basson has over 25 years of experience spanning across a 150 projects, 18 commodities and a plethora of deposit types, throughout 23 African countries and the Middle East. He has evaluated several tens of shear zone-hosted, vein-hosted, greenstone belt and Birimian gold mineralization in variably metamorphosed terranes. This has included the interpretation and integration of geophysical surveys, field mapping, core logging, 3D geological modelling and reporting. He has published 38 peer-reviewed papers to date, excluding conference abstracts and presentations. In 2002, he founded TECT Geological Consulting, which specializes in the application of structural geology to exploration and mining projects and deposits at all stages of development. The focus of the company is to resolve complex structural environments and the way that they affect deposition, evaluation, mining and geohydrological modelling.

Ken Lomberg, Director (Geology and Resources), Pivot

B.Sc. (Hons) Geology, B.Com., M.Eng., FGSSA, Pr.Sci.Nat. Project Management, Mineral Resources, geological interpretations, site visits, report preparation.

Mr Lomberg has some 38 years' experience in the minerals industry (especially platinum and gold). He has been involved in exploration and mine geology and has experience in the technical development of mining projects from inception to full production. He is a respected professional with advanced capability, particularly in project management and Ore Reserve and Resource estimation as a result of his exposure to a wide range of mineral sector consulting assignments. Mr Lomberg has undertaken Mineral Resource and Reserve estimations and reviews for platinum, chromite, gold, copper, uranium and fluorite projects. He has assisted with the reviews or estimation of diamond and coal projects. He has assisted with or compiled Competent Persons Reports/NI 43-101 for various companies that have been listed on the TSX, JSE and AIM.

2.4 Independence

Neither TECT nor Pivot, or the key personnel contributing to the completed and reviewed work, has any interest (present or contingent) in Xtra-Gold Resource Corp. (Xtra-Gold) or its subsidiaries, its directors, senior management, advisers or the mineral properties reported on in this report. The proposed work, and any other work done by TECT and Pivot for Xtra-Gold, is strictly in return for professional fees. Payment for the work is not in any way dependent on the outcome of the work, nor on the success or otherwise of Xtra-Gold's own business dealings.

There is no conflict of interest in TECT or Pivot in undertaking the assignment as contained in this document.

2.5 Site and Technical Visits

Dr Koegelenberg has undertaken two visits to the Kibi Project Area. Firstly, over the period 1-12 September 2019, and secondly, 29 November - 10 December 2020. The first visit was solely to conduct structural field mapping, review of selected drill core and 3D structural modelling relevant to Zone 5 (Cobra Creek). The second visit focussed on structural drill core reviews, structural field mapping and 3D structural and mineralization modelling of targets and prospects situated in Zones 1-4, which contain the mineral resources described in this report.

Mr Lomberg accompanied TECT geoscientists on the second visit, when several trenches, road cuts, and drill hole collars were visited and evaluated in the field for mineralization style and appropriate trench and drillhole sampling methods. At site, the storage sheds and sampling areas were examined, as well as the data room and recent drill core.

Mr Lomberg undertook a second visit over the period 20 – 25 June 2024. A visit to the field to was undertaken in Zone 4 and Zone 3 to confirm drill collar and trench locations as well as be oriented to the targets areas. At the core yard, the storage sheds and sampling areas were examined, and the process of core management followed to confirm that the data collection was undertaken in the appropriate manner. The examination of the core allowed the evaluation of the style of mineralisation and confirmed the important aspects of the gold paragenesis.

3 RELIANCE ON OTHER EXPERTS

This report was prepared as a National Instrument 43-101 Technical Report, in accordance with Form 43-101F1, for Xtra-Gold, the issuer, by TECT Geological Consulting Pty Ltd and Pivot Mining Consultants Pty Ltd. The quality of information and conclusions contained herein is consistent with the level of effort involved in Pivot's services and is based on:

- Information available at the time of preparation from Xtra-Gold,
- Third party technical reports prepared by Government agencies and previous tenement holders.
- Other relevant published and unpublished third-party information.

This report is intended to be used by Xtra-Gold, subject to the terms and conditions of its contract with TECT and Pivot. This contract permits Xtra-Gold to file this report as a Technical Report with Canadian Securities Regulatory Authorities pursuant to National Instrument 43-101, Standards of Disclosure for Mineral Projects. Any other use of this report by any third party is at that party's sole risk.

A final draft of this report was provided to Xtra-Gold, along with a written request to identify any material errors or omissions, prior to lodgement.

Neither TECT nor Pivot, nor the authors of this report, are qualified to provide extensive comment on legal facets associated with ownership and other rights pertaining to Xtra-Gold's mineral properties described in Section 4. Pivot did not see or carry out any legal due diligence confirming the legal title of Xtra-Gold to the properties.

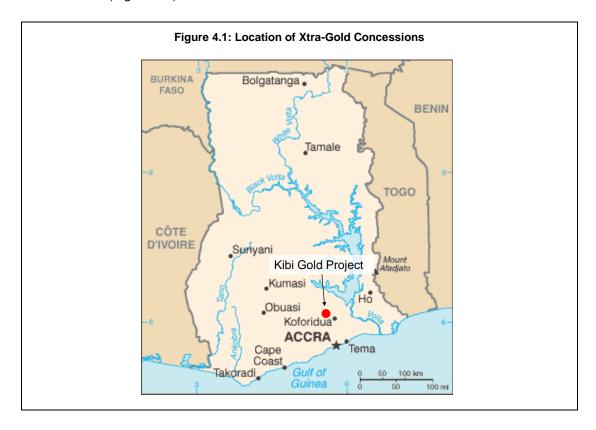
Similarly, neither TECT nor Pivot nor the authors of this report are qualified to provide extensive comment on environmental issues associated with Xtra-Gold's mineral properties, as discussed in Section 4.

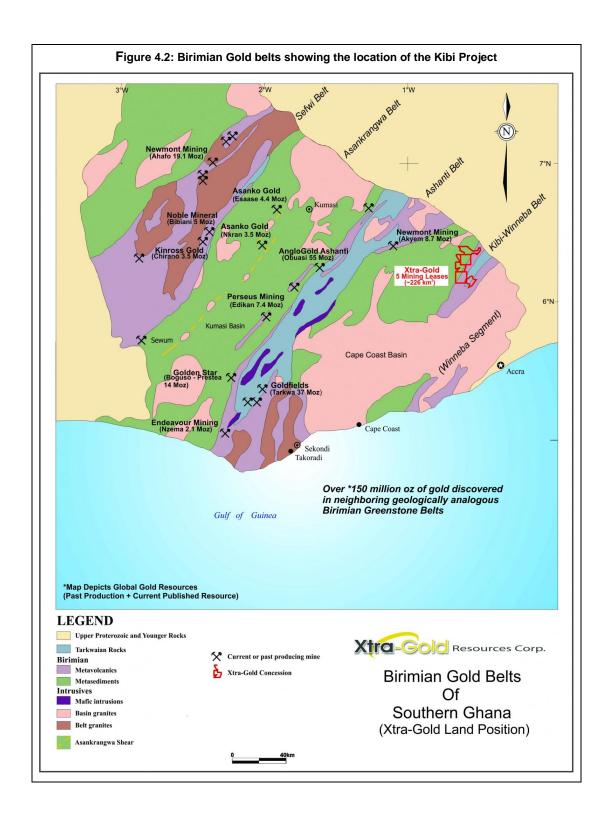
TECT and Pivot relied on Xtra-Gold for the information in respect of the Prospecting Permits and Environmental Permits.

4 PROPERTY DESCRIPTION AND LOCATION

4.1 Property Description and Location

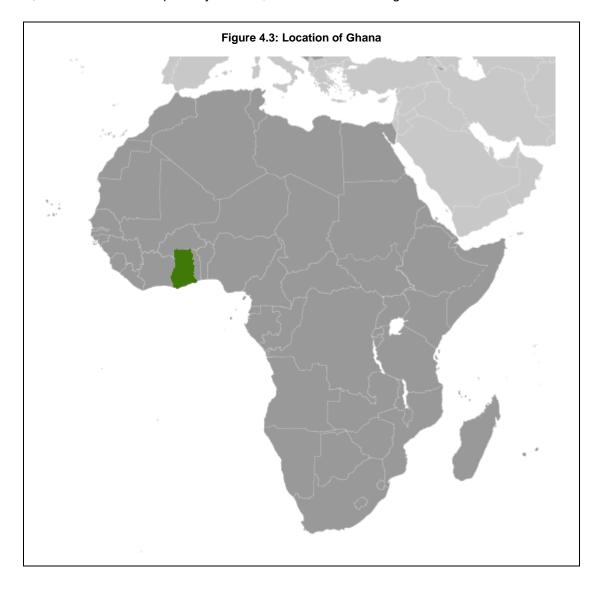
The Apapam Concession is located approximately 75 km north-northwest of Accra, in the East Akim District of the Eastern Region of Ghana, on the eastern flank of the Atewa Range near the headwaters of the Birim River (Figure 4.1). The centre of the concession is situated at approximately 6° 09' 30" West Longitude and 0° 34' 15" North Latitude (WGS 84). The Kibi Project area spans 3,365 ha and is located at the northern extremity of the Kibi Winneba Greenstone Belt (Figure 4.2).





4.2 Country Profile: Ghana

The Republic of Ghana (Ghana), formerly known as the Gold Coast, is located in West Africa on the Gulf of Guinea (Figure 4.1) and shares borders with Côte d'Ivoire (Ivory Coast) to the west, Togo to the east and Burkina Faso (formerly Upper Volta) to the north. To the south are the Gulf of Guinea and the Atlantic Ocean. Ghana has a total land area of approximately 239,533 km². Ghana's capital city is Accra, which is located along the south-eastern coast.



The Republic of Ghana was formed from the merger of the British colony of the Gold Coast and the Togoland trust territory in 1957. Ghana endured a series of coups before Lt. Jerry Rawlings took power in 1981 and banned political parties. After approving a new constitution and restoring multiparty politics in 1992, Rawlings won presidential elections in 1992 and 1996 but was constitutionally prevented from running for a third term in 2000. John Kufuor of the opposition New Patriotic Party (NPP) succeeded him and was re-elected in 2004. John Atta Mills of the National Democratic Congress won the 2008 presidential election and took over as Head of State. Mills died in July 2012 and was constitutionally succeeded by his vice president,

John Dramani Mahama, who subsequently won the December 2012 presidential election. In 2016, Nana Addo Dankwa Akufo-Addo of the NPP defeated Mahama, marking the third time that Ghana's presidency has changed parties since the return to democracy. He was re-elected after a tightly contested election in 2020.

After several peaceful transitions to power, the generally smooth elections and peaceful transfer of power in January 2017 has confirmed Ghana as one of Africa's most stable democratic states (The World Bank 2017).

4.2.1 Economy

Ghana has a market-based economy with relatively few policy barriers to trade and investment in comparison with other countries in the region. The country is endowed with economically viable mineral and petroleum natural resources. Ghana's economy was strengthened by a quarter century of relatively sound management, a competitive business environment, and sustained reductions in poverty levels, but in recent years has suffered the consequences of loose fiscal policy, high budget and current account deficits, and a depreciating currency.

Agriculture accounts for about 20% of GDP and employs more than half of the workforce, mainly small landholders. Gold, oil, and cocoa exports, and individual remittances, are major sources of foreign exchange. Expansion of Ghana's nascent oil industry has boosted economic growth, but the fall in oil prices since 2015 reduced by half Ghana's oil revenue. Production at Jubilee, Ghana's first commercial offshore oilfield, began in mid-December 2010. Production from two more fields, Ten and Sankofa, started in 2016 and 2017 respectively. The country's first gas processing plant at Atuabo is also producing natural gas from the Jubilee field, providing power to several of Ghana's thermal power plants.

As of 2018, key economic concerns facing the government include the lack of affordable electricity, lack of a solid domestic revenue base, and the high debt burden. The Akufo-Addo administration has made some progress by committing to fiscal consolidation, but much work is still to be done. Ghana signed a \$920 million extended credit facility with the IMF in April 2015 to help it address its growing economic crisis. The IMF fiscal deposits require Ghana to reduce the deficit by cutting subsidies, decreasing the bloated public sector wage bill, strengthening revenue administration, boosting tax revenues, and improving the fiscal health of Ghana's banking sector. Priorities for the new administration include rescheduling some of Ghana's \$31 billion debt, stimulating economic growth, reducing inflation, and stabilizing the currency. Prospects for new oil and gas production and follow through on tighter fiscal management are likely to help Ghana's economy in 2018.

According to the African Development Bank (https://www.afdb.org/en/countries/west-africa/ghana/ghana-economic-outlook), the COVID–19 pandemic has significantly curtailed Ghana's economic growth momentum. Real GDP growth was estimated to decelerate from 6.5% in 2019 to 1.7% in 2020 due to the slump in oil prices and weakened global economic activity.

Real GDP growth slowed to 3.3% in 2022 from 5.4% in 2021 due to macroeconomic instability, global financial tightening, and spillover effects of Russia's invasion of Ukraine. Inflation was an estimated 31.5% in 2022, up from 10% in 2021, driven by food and energy prices and depreciating local currency. The Bank of Ghana tightened monetary policy; the policy rate was hiked to 27% in 2022 from 14.5% in 2021. The fiscal deficit widened slightly, to 9.3% of GDP from 9.2% in 2021, due to higher spending. Public debt hit 93.5% of GDP in 2022, up from 82.0% in 2021, driven by primary fiscal deficits and exchange rate depreciation. (https://www.afdb.org/en/countries/west-africa/ghana/ghana-economic-outlook)

4.2.2 Infrastructure

Local infrastructure consists of 110,000 km of roads (14,000km tarred and 96,000 km untarred), 947km railways, seven paved airport runways.

4.2.3 Population

Ghana has a population of about 33.8 million (2023) consisting of a number of ethnic groups (Akan 45.7%, Mole-Dagbon 18.5%, Ewe 12.8%, Ga-Dangme 7.1%, Gurma 6.4%, Guan 3.2%, Grusi 2.7%, Mande 2.0%, other 1.6%)(2021 est) and speaking various local languages (Asante 16%, Ewe 14%, Fante 11.6%, Boron (Brong) 4.9%, Dagomba 4.4%, Dangme 4.2%, Dagarte (Dagaba) 3.9%, Kokomba 3.5%, Akyem 3.2%, Ga 3.1%, other 31.2%)(2010 est). English is the official language. Various religions are practiced:- Christian 71.3% (Pentecostal/Charismatic 31.6%, Protestant 17.3%, Catholic 10.0%, other 12.3%), Muslim 19.9%, traditional 3.2%, other 4.5%, none 1.1% (2021 est.). Some 80.38% of the population are considered literate (2021 est).

4.2.4 Gold in Ghana

(Quoted from_THE WORLD BANK June 2017 shifting Ghana's competitiveness into a higher gear Ghana economic update 40pp.)

For centuries, gold and Ghana have been synonymous. Hundreds of years before the Portuguese sailors first arrived (late 1400s) along the coast of West Africa in search of gold, large quantities of this treasured metal had been mined and transported across the Sahara to North Africa. This trade helped to establish and support major trading centres in the Sahel along the course of the Niger River and contributed much to the main trading ports along the southern coast of the Mediterranean Sea.

The arrival of the European traders led to increased gold production within the forest areas of southern Ghana and the area became known as the Gold Coast. It soon became one of the most important gold-producing areas in the world and eclipsed other major producing areas in present day western Mali (Bambuk) and northern Guinea (Boure), which had been the earliest sources in West Africa that had fostered the development of several successive inland states/empires in the Sahel over a period of many centuries.

For a period of almost 400 years (1490s to late 1800s), European traders competed vigorously for gold brought to the many forts and trading posts strung out along the Gold Coast. Most

unfortunately, the gold business was much diminished for almost 3 centuries by the diabolical slave trade and the coastal forts soon became way-stations for West Africans carted off to the New World where they were enslaved as plantation workers. This period was very disruptive to the gold trade, which recovered only in the early to mid-1800s, after slave trading was abolished in most countries and curtailed by the British navy who intercepted slave ships along the coast of West Africa.

In the late 1800s, when the emerging European powers were carving up their respective interests in Africa, Great Britain emerged the dominant European power along the Gold Coast and claimed it as a British colony in the mid-1870s. This marked the beginning of a new era in gold mining in the region as foreign companies were able to acquire gold concessions from local chiefs in very prospective areas in the interior of the country. By the beginning of the 20th century, modern gold operations began to emerge in many districts of southern Ghana.

During the colonial period, gold exploration and production waxed and waned according to economic conditions in the world economy. There was a huge but brief gold rush at the very beginning of the 20th century, which coincided with the Boer War in South Africa, and a much more sustained rush throughout the 1930s when gold production reached historical highs and world-class operations developed at the famous Ashanti mine in Obuasi and in the Tarkwa, Prestea, and Bibiani districts

After World War II and leading up to Ghana's independence in 1957, the gold production remained substantial and the mines were critical to the economy of the new nation. However, the rising costs of production at a time of a fixed gold price made many of the existing operations marginally profitable, except for the Ashanti mine whose traditionally very high grades kept it amongst the premier gold producers in the world. In the early 1960s, the Government of Ghana bought out several of the marginal producers and formed the State Gold Mining Corporation. This parastatal company achieved very good results in the 1960s and early 1970s but a sustained lack of capital investment in the existing operations led to a downward spiral in production throughout the 1970s and early 1980s. Furthermore, there had been little if any new exploration since the 1930s, so no new producers were on the horizon.

It was not until the mid-1980s that the downward trend was reversed as a result of the implementation of a broad Economic Recovery Programme, which included a significant focus on the mining sector. This resulted in updated laws and regulations as well as fiscal incentives to attract foreign capital to carry out exploration and the development of new producers or the upgrading of existing gold operations. These new policies were extremely successful and created the necessary 'enabling environment' that culminated in huge capital expenditures on exploration and development projects.

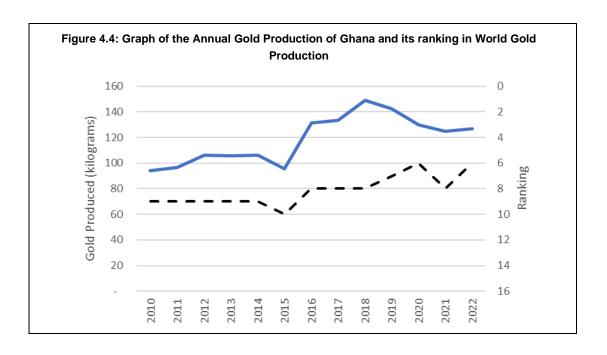
Dramatic increases in production were achieved and have been sustained for the past 20 years during which Ghana has been the largest gold producer in Africa and over the past few years the country has broken into the top ten world producers. Ghana's success has become a case history for many developing nations, especially with respect to stimulating exploration and mining.

During the latest exploration boom, much of the early interest naturally focused on previous producers and the implications of new treatment schemes to more effectively mine lower grade deposits at mines that had closed down. In addition, many of the older mines now had more reason to re-evaluate known resources on their extensive concessions and to bring in new ideas and ways to increase production. This was especially true at Obuasi where a new and enlightened management team undertook huge development schemes that greatly increased gold production from about 250,000 oz in the mid-1980s to almost 1 million ounces in the mid-1990s.

In addition, the Government's efforts to disassemble and privatize the exploration and mining assets of the State Gold Mining Corporation were finally successful in the early 1990s and subsequent developments have seen very impressive production increases. For example, in Tarkwa, the old underground operations were producing a very modest 20,000-40,000 oz/yr in the late 1980s through the mid-1990s but Goldfields started a very large low-grade open-pit operation in the late 1990s and saw production shoot up to over 300,000 oz by the end of the decade and for the past several years it has been the #1 producer in the country at over 600,000 oz/yr.

In addition, the Newmont group, which had inherited several prospects in the Ahafo area on the north side of the Sefwi Belt, discovered a world-class district that has very quickly become the third biggest producer in the country and keeps expanding production. Newmont is also now developing the very large Akyem deposit on the NE margin of the Ashanti Belt. The annual production and associated ranking of Ghana as a gold producer is presented in Figure 4.4. Ghana has regained its position as the number one producer of gold in Africa with the production of 3.74 Moz.

Ghana plans to maintain an 'enabling environment' for investment in the mining sector. This starts with a full commitment to good governance and the strengthening of democratic institutions that has gained the nation much favourable attention in recent times. It also includes modifying laws and regulations to improve the overseeing and administering of exploration and mining activities as well as establishing a fair and equitable distribution of the profits from mining in the country.



4.2.5 Overview of the Mineral Laws of Ghana

The Minerals and Mining Act, 2006 (Act 703) (the Mining Act) was enacted in 2006.

According to the Mining Act, all minerals are the property of the Republic of Ghana and are vested in the President in trust for the people of Ghana. Granting of the various mineral titles is done by the Minister responsible for mines on behalf of the President and on the recommendation of the Minerals Commission. Ghana is now using a cadastral system for new tenement applications where the country is divided into blocks that are 15 seconds of longitude by 15 seconds of latitude (approximately 21 hectares or 0.21 km² in area).

Table 4.1 summarises the characteristics and conditions for the various types of mineral titles as described in the Mining Act.

- A Reconnaissance Licence allows a holder exclusive right to conduct exploration activities not including drilling or excavation.
- A Prospecting Licence allows a holder to explore for minerals exclusive right to conduct exploration activities including drilling or excavation.
- A Mining Lease allows a holder to extract and process ore. Similarly, a small-scale mining lease also allows a holder to extract and process ore, including the use of mercury, but may only use explosives with the written permission of the Minister.

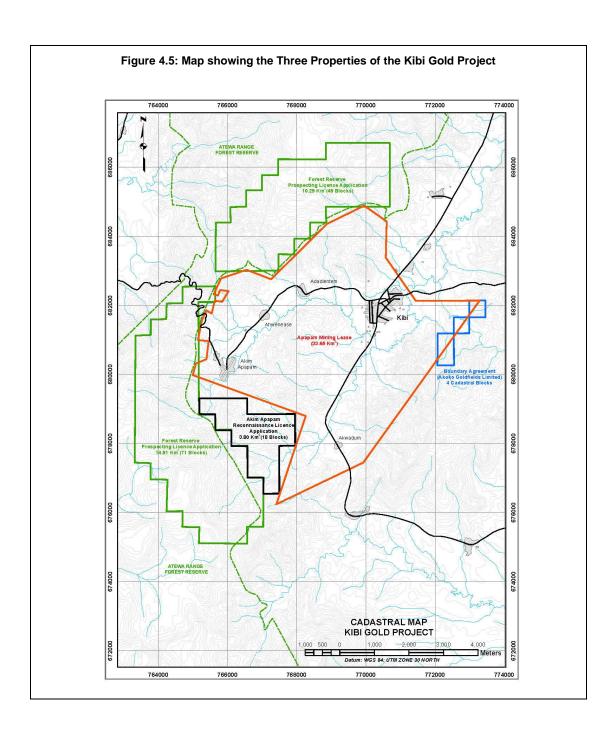
With mining leases, holders can hold up to 90% interest. The remaining 10% interest is held by the Government of Ghana. Pursuant to the Mining Act, the Government of Ghana acquires a 10% free carried interest in all mining leases by way of 10% share ownership in all Ghanaian corporations who hold mining leases.

Table 4.1: Characteristics of the Mineral Titles - Minerals and Mining Act, 2006 (Act 703).					
Type of mineral title	Reconnaissance Licence	Prospecting Licence	Mining Lease		
Maximum area allowed (blocks)	5,000	750	300		
Minimum area to be relinquished after initial term	-	50%	-		
Initial term of mineral title (years)	1	2	30		
Extendable for a further period (years) - 100% retained	1	1	30		
Renewable for a further period (years) - 50% relinquishment		2			
Application forms (US\$)	250	250			
Processing fee - applications & renewals (US\$)	500	500			
Consideration fee - applications & renewals (US\$)	15,000	20,000	100,000		
Consideration fee - extension (US\$)		15,000			
Ministerial consent to agreements (US\$)	20,000	40,000	80,000		

4.3 Mining Tenure

Through its subsidiary companies, Xtra-Gold has three titles in the Kibi Gold Project including the Apapam Mining Lease (Figure 4.6). The Apapam Concession is a granted mining lease, whereas the other two titles are currently applications that have been submitted to the Minerals Commission (Table 4.2).

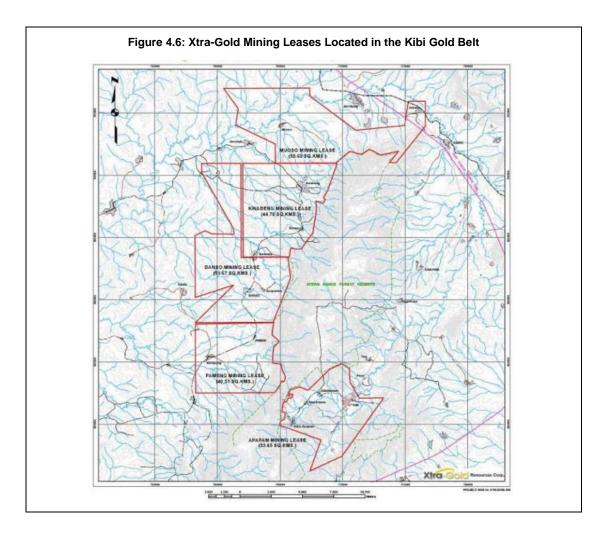
Table 4.2: Mineral Tenements of the Kibi Gold Project						
Title Number	Name	Type of Mineral Title	Area (km²)	Date of Application	Date Granted	Expiry Date
LVB 5191/09	Apapam	Mining Lease	33.65		18/12/2008	17/12/2015
RL5/44	Akim Apapam	Reconnaissance Licence Application	3.80	15/1/2008	Pending	N/A
PL5/260	Forest Reserve Prospecting Licence Application	Forest Reserve Prospecting Licence Application	25.49 (2 Blocks)	08/08/2018	Pending	N/A



4.4 Licence Status

Xtra-Gold holds five (5) concessions in the Kibi Gold Belt (Figure 4.6), including the Apapam mining lease located on the eastern flank of the Atewa Range, and four (4) contiguous mining leases situated on the western side of the Atewa Range, for a total land position of approximately 226 km² (22,600 ha) (Table 4.3).

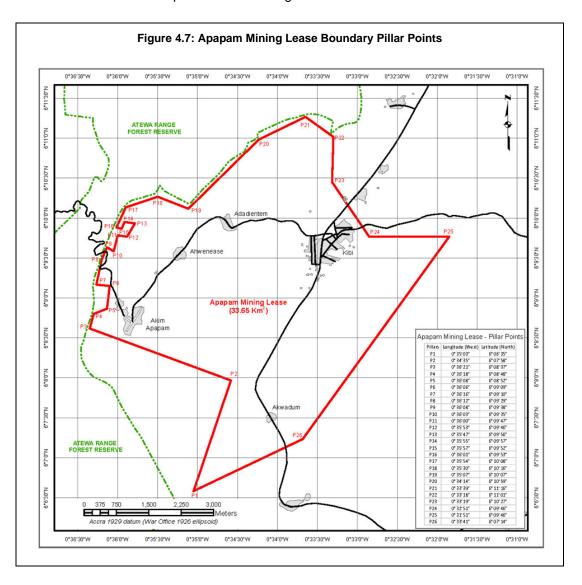
Table 4.3: Summary of Xtra-Golds Mining Leases						
Title	Name Title	Area (km²)	Date of Grant	Expiry Date	Date of Application for Renewal	
	Mining Leases					
LVB 5191/09	Apapam	33.65	Dec.18, 2008	Dec.17, 2015	June 17, 2015	
LVD 1896A/2011	Muoso	55.65	Jan. 6, 2011	Jan. 5, 2024	June 12, 2023	
LVD 1896/2011	Banso	51.67	Jan. 6, 2011	Jan. 5, 2025	May 31, 2024	
LVB 2911/89	Kwabeng	44.76	July 26, 1989	July 26, 2019	Dec. 13, 2018	
LVB 4815/89	Pameng	40.51	July 26, 1989	July 26, 2019	Dec. 13, 2018	
	•	226.64		•	•	



4.4.1 Apapam Mining Lease

The Apapam Mining Lease is stamped as No. LVB 5191/09 by the Lands Commission, registered as No. 24/2009 at the Land Registry, and has the File No. PL.5/142 at the Minerals Commission of Ghana.

The concession boundaries have not been legally surveyed but are described by Latitude and Longitude coordinates via decree. With the mining lease boundaries defined by a series of pillar points in Latitude and Longitude coordinates utilizing the local Accra 1929 datum based on the British War Office (1926) ellipsoid. The concession pillar points as described by the Minerals Commission are depicted / listed on Figure 4.7.



Xtra-Gold Mining Limited, which is 90% owned and controlled by Xtra-Gold with the remaining 10% interest being held by the Government of Ghana, is the registered holder of the Apapam Mining Lease. Pursuant to the Mining Act, the Government of Ghana acquires a 10% free carried interest in all mining leases by way of 10% share ownership in all Ghanaian corporations who hold mining leases.

While the mining lease expired in 2015 it can be renewed for a further 30-year term in accordance with the Mining Act by Xtra-Gold Mining making application not less than six months prior to the expiry date.

An application to extend the lease for a further 15 years was duly submitted to the Ghana Minerals Commission on June 17, 2015. The lease extension application also includes the area between the existing lease boundary up to the Forest Reserve boundary, which is termed the "Buffer Zone". The Ghanaian Minerals Commission's Technical Committee has certified the expansion of the lease area, which includes the Buffer Zone, to be in conformity with the official cadastral system under Ghana Mining Laws and, as such, is a proper extension of the current lease. Refer to the Apapam mining lease Status Opinion prepared by REM Law Consultancy (Appendix B) for further details on the lease renewal / extension, including the Buffer Zone area.

Xtra-Gold is currently awaiting formal approval of the lease extension. Although the Company has taken all legal steps to extend the lease with the addition of the Buffer Zone, there is no assurance that such approval will be granted. In the interim, under Ghanaian mining law, the existing lease continues to remain in full force and effect.

Pursuant to the terms and conditions of the Apapam Mining Lease, Xtra-Gold was granted surface and mining rights by the Government of Ghana to work, develop and produce gold in the mining lease area (including the processing, storing and transportation of ore and materials).

4.4.2 Apapam Mining Lease Boundary Agreement

After the grant of the Apapam Mining Lease, the Minerals Commission implemented the cadastral system as required under the Minerals and Mining (Licensing) Regulations 2012 (L.I. 2176). As a result of the implementation of the cadastral system it became necessary for the common boundary shared by the Apapam mining lease and the Akoko mining lease to be readjusted / modified to conform with the cadastral blocks created by the new cadastral system.

Xtra-Gold Mining Limited and Akoko Goldfields Limited subsequently entered into a mutual consent agreement on September 26, 2016, giving the authority to the Mineral Titles Division of the Minerals Commission to readjust / modify the common boundary between the Apappam and Akoko mining leases to conform to the cadastral system with relation to the four (4) cadastral blocks depicted on the Kibi Gold Project Cadastral Plan (Figure 4.4). The boundary agreement was filed with the Minerals Commission on November 1, 2016.

4.4.3 Akim Apapam Reconnaissance Licence Application

Xtra-Gold's land position in the Kibi Gold Project area also includes the Akim Apapam reconnaissance licence application, contiguous to the southwest extremity of the Apapam mining lease (Figure 4.5). The reconnaissance license application for this 3.80 km² ground parcel was submitted to the Minerals Commission on January 15, 2008, in the name of Xtra-Gold Exploration Limited. The application area was originally 7.0 km² but was reduced to its current 3.80 km² (18 cadastral blocks) by the Minerals Commission in 2015 to conform with the new cadastral system.

The application was approved by the Minerals Commission on January 18, 2019, but as at the date of this Report, the reconnaissance licence is still being processed by the Minerals

Commission, and Xtra-Gold has yet to receive legal title to this ground. Xtra-Gold has conducted limited exploration work on the Akim Apapam application ground.

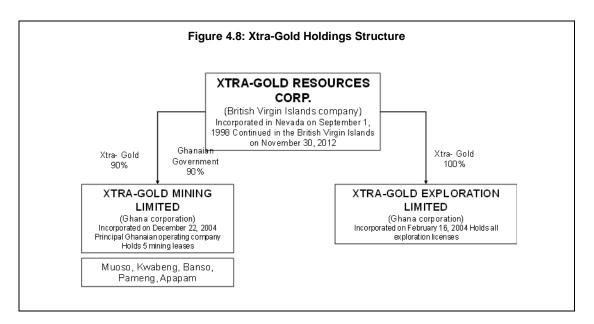
4.4.4 Forest Reserve Prospecting Licence Application

The Kibi Gold Project land position includes a Prospecting Licence Application to conduct mineral exploration activities within the fringes of the Atewa Range Forest Reserve bordering the Apapam mining lease and the Akim Apapam reconnaissance licence application (Forest Reserve Application) (Figure 4.5). The Forest Reserve Application, covering a total area of approximately 25.49 km², encompasses two blocks: a northern block of approximately 10.29 km² (49 cadastral units); and a western block of approximately 14.91 km² (71 cadastral units). With the approximately 450 m gap between the two blocks occupied by the Company's Pameng mining lease (Figure 4.6). The current Forest Reserve Application supersedes the November 13, 2009, Apapam mining lease extension application.

The application process for mineral rights within a Forest Reserve requires two separate permits: a Forest Entry Permit from the Forestry Commission; and a Prospecting Permit from the Minerals Commission. The application for the Forest Entry Permit was submitted on June 20, 2017, with the Forest Entry Permit granted by the Forestry Commission on October 17, 2018. The Prospecting Licence Application was submitted on August 8, 2018, and as at the date of this Report, the application is still being processed by the Minerals Commission.

4.5 Holdings Structure

The corporate holdings structure is summarised in Figure 4.8.



4.6 Royalties and Agreements

With respect to the Apapam mining lease, Xtra-Gold are:

- Required to obtain a yearly Mining Exploration Operating Permit from the Inspectorate Division of the Ghana Minerals Commission in the amount of GH¢5,000 (approximately USD 400) to conduct prospecting activities on the concession. Xtra-Gold has an Exploration Operating Permit valid until December 31, 2024.
- Required to pay applicable taxes and annual rental (ground rents) fees to the Government of Ghana in the amount of approximately USD 21,172 (approximately GH¢275,000), when the renewal extension is granted; and
- Committed to pay a royalty in each quarter to the Government of Ghana, through the Commissioner of Internal Revenue, based on the production for that quarter within 30 days from the quarter end as well as a royalty on all timber felled in accordance with existing legislation;
- Required to:
 - Commence commercial production of gold within two years from the issue date of the mining lease;
 - Conduct operations with due diligence, efficiency, safety and economy, in accordance
 with good commercial mining practices and in a proper and workmanlike manner,
 observing sound technical and engineering principles using appropriate modern and
 effective equipment, machinery, materials and methods and paying particular regard
 to the conservation of resources, reclamation of land and environmental protection
 generally; and
 - Mine and extract ore in accordance with the preceding paragraph, utilizing methods which include dredging, quarrying, pitting, trenching, stoping and shaft sinking in the Apapam lease area.

Xtra-Gold are required to furnish to the government authorities of Ghana, comprising the Minister of Lands, Forestry and Mines, the Head of the Inspectorate Division of the Minerals Commission, the Chief Executive of the Minerals Commission and the Director of Ghana Geological Survey (government authorities), with technical records which include:

- a report in each quarter not later than 30 days after the quarter end to the government authorities in connection with quantities of gold won in that quarter, quantities sold, revenue received and royalties payable;
- A report half-yearly not later than 40 days after the half year end to the government authorities summarizing the results of operations during the half year and technical records, which report shall also contain a description of any geological or geophysical work carried out by the company in that half year and a plan upon a scale approved by the head of the inspectorate division of the minerals commission showing dredging areas and mine workings;
- A report in each financial year not later than 60 days after the end of the financial year summarizing the results of the operations in the lease area during that financial year and the technical records, which report shall further contain a description of the proposed operations for the following year with an estimate of the production and revenue to be obtained:
- A report not later than three months after the expiration or termination of the Apapam mining lease, to the government authorities giving an account of the geology of the lease area including the stratigraphic and structural conditions and a geological map on scale prescribed in the mining regulations;

- A report not less than 21 days in advance of the proposed alteration, issuance or borrowing
 to the government authorities (except for the head of the inspectorate division of the
 minerals Commission and the Director of Ghana Geological Survey) of any proposed
 alteration to the applicable regulations,
- A report not less than 21 days in advance of the proposed alteration, issuance or borrowing to the government authorities (except for Head of the Inspectorate Division of the Minerals Commission and the Director of Ghana Geological Survey) on the particulars of any fresh share issuance or borrowings in excess of an amount equal to the stated capital of Xtra-Gold Mining;
- A copy of Xtra-Gold Mining's annual financial reports to the government authorities (except for the Head of the Inspectorate Division of the Minerals Commission and the Director of Ghana Geological Survey) including a balance sheet, profit and loss account and notes thereto certified by a qualified accountant, who is a member of the Ghana Institute of Chartered Accountants, not later than 180 days after the financial year end; and
- Such other reports and information in connection with the operations to the government authorities as they may reasonably require.

All gold production will be subject to a production royalty of the net smelter returns (NSR) payable to the Government of Ghana.

The Kibi Gold Project is not subject to any back-in rights, payments or other agreements and encumbrances.

4.7 Environmental Liabilities

All exploration activities in Ghana are subject to regulation by governmental agencies under various environmental laws. These laws address emissions into the air, discharges into water, management of waste, management of hazardous substances, protection of natural resources, antiquities and endangered species, and reclamation of lands disturbed by mining operations. Compliance with environmental laws and regulations may require significant capital outlays and may cause material changes or delays in intended activities.

An Environmental Permit for Mineral Exploration is required from the Environmental Protection Agency (EPA) to carry out prospecting activities on any type of mining licence in Ghana. Xtra-Gold was granted a two-year mineral exploration permit to conduct its exploration work on the Apapam Mining Lease which expired on May 8, 2024. An application for the renewal of the environmental permit was submitted to the EPA on January 16, 2024, and as at the date of this Report, the EPA has yet to issue a new environmental permit to cover the ongoing exploration activities on the concession. It is accepted practice in Ghana for a mining company to carry on with its exploration activities while awaiting for the EPA to process an application for an environmental permit. The granting of environmental permits by the EPA can take several months or longer.

The project is not subject to any known environmental liabilities except as set forth below:

In accordance with the rules and regulations of the Environmental Protection Agency (EPA) of Ghana, the open trenches excavated by Xtra-Gold must be backfilled after mapping and

sampling has been completed. Xtra-Gold has adopted a program of backfilling all excavations once mapped and sampled, however, some trenches have been preserved for ongoing exploration purposes and comparison with drilling. Drilling requires the construction of access roads and clearing of land for drill pads to accommodate the drill during operation. Xtra-Gold has adopted a policy of keeping the width of access roads and size of drill pads to a minimum to mitigate the impact on the vegetation. Also, drill cuttings are collected in sumps with the sumps backfilled after the completion of the drill hole.

In areas where there is a lawful surface holder or occupier, Xtra-Gold is required under the Mining Act to pay compensation when land is disturbed, in most cases this is related to the disturbance of crops during access road construction and trenching / drilling activities. Reasonable / fair crop compensation terms are always negotiated with the farm owners prior to the start of exploration work and Xtra-Gold has a good working relationship with the local communities.

Xtra-Gold is not responsible for small-scale artisanal and alluvial mining that has taken place across the project area.

5 ACCESSIBILITY, CLIMATE, LOCAL RESOURCES, INFRASTRUCTURE AND PHYSIOGRAPHY

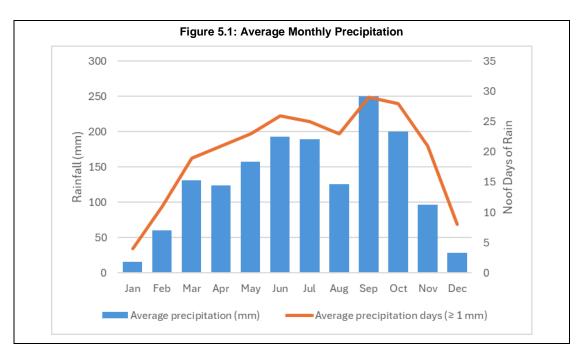
5.1 Access

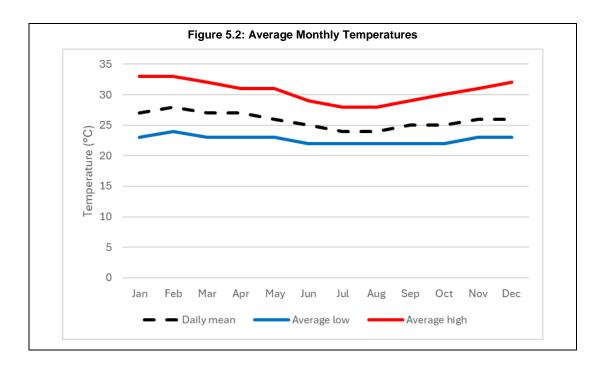
The concession is located approximately 75 km north-northwest of Accra and can be accessed by two asphalted secondary highways. Access to the Kibi Gold Project can be achieved by driving approximately 75 km northwest from Accra on the paved Accra- Kumasi Trunk Road which is the main national highway. A tarred road emanating from the Accra-Kumasi Trunk Road approximately 15 km northeast of Kibi dissects the north-central and south-eastern portions of the Kibi Gold Project, while the tarred road servicing the town of Apapam provides access to the south-western extremity of the project.

A network of foot paths and tracks link most of the communities within the concession areas and provide access to the areas where exploration and drilling have been taking place. Xtra-Gold constructed a number of roads and 4WD tracks to provide access to the drill sites.

5.2 Climate

The climate within the area is equatorial with relatively high humidity throughout the year. Rainfall is generally characterised by two periods, with high and unpredictable rain especially during the peak period which falls in May/June with a second peak in September/October. The rainfall per annum is some 1600 mm (Figure 5.1). The dry season is generally from January to February. Temperatures ranges between 22°C and 33°C (Figure 5.2: Average Monthly TemperaturesFigure 5.2). Operations can be undertaken throughout the year.





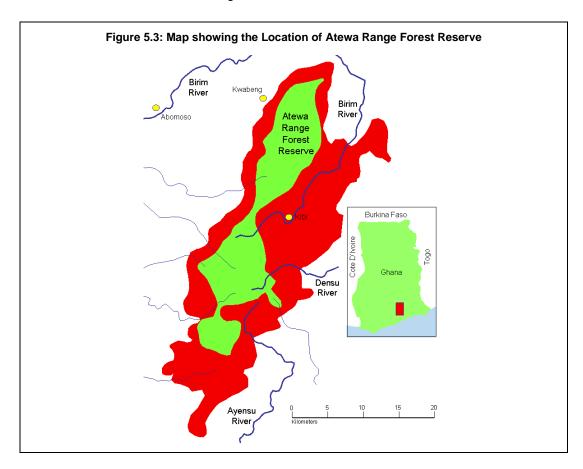
5.3 Physiography

The topography of the Apapam Concession is characterized by steep sloping ridges and undulating mountain side hills due to the prominent, NNE trending Atewa Range that is about 50 km long and 10-15 km wide and dominates the area. The range consists of steep-sided hills with fairly flat summits as the last remains of the Cenozoic peneplain that once covered southern Ghana, and contains ancient bauxitic soils. The steep flanks feature a wide variety of high canopy tropical hardwoods typical of south-western Ghana whereas the summit has a diverse flora, including extensive hanging vines. The range is the site of an important forest reserve, and the source of the Birim River and its tributaries drain the area. The Birim River makes a long detour north and southwest around the Atewa range before joining the Pra River.

Relief in most parts of the Apapam Concession is quite modest (10-30 m) but changes abruptly at the base of the steep-sided flanks of the Atewa Range. The maximum elevation on the Range is about 780 m above mean sea level and stands well above the surrounding lowlands, which are at approximately 180-200 m above mean sea level.

A large area of the Atiwa range has been declared a Forest Reserve, including about 17,400 hectares of upland evergreen forest, rare for Ghana (Figure 5.3). The reserve is managed by the Forestry Commission of Ghana in collaboration with other stakeholders, key among them is the Okyeman Environment Foundation, which has restricted people from farming in the area and instead is trying to encourage eco-tourism. However, the reserve is under pressure from logging and hunting for bushmeat. It is also vulnerable to mining exploration activities since the reserve contains gold deposits as well as low-grade bauxite. The Apapam concession lies at the base and the lower flanks of the Atewa Range immediately adjacent to the boundary of the Forest Reserve.

The main vegetation of the area is basically moist deciduous forest which is characterized by thick canopy tall trees with a layer of shorter trees and evergreen shrubs in the undergrowth. The area is dominated by tropical hard wood species such as Odum, Wawa, Ofram, Asamfra, Mahogany, Teak, Bamboo and Okyenkyen. Tree crops such as cocoa, oil palm, coffee, mango and citrus are also found to be thriving well in the area.



5.4 Local Resources and Infrastructure

Ghana has a fairly good network of paved highways and roads. Within the Apapam Concession, numerous tracks and paths are available for easy access to most points.

Power is available in larger towns and cities. The electrical grid follows the main secondary roads and most of the major villages in the Kibi District have electrical power. When the national power grid is not available, generators are used for backup power.

The district has quite a large population that is well spread out in many towns (1,000-5,000 population) and villages; small farming hamlets occur throughout the area. The district capital is Kibi (also Kyebi), which is also the seat of the Paramount Chief, or Okyehene, of the Akim people. Kibi is a small city with considerable infrastructure (schools, hospital, police headquarters, etc.). The major towns (Asaaman, Apapam, Kibi and Anyinam) have limited centralized pipe-born water supplies with most of the towns depending on wells and drill holes as well as nearby streams.

Telephone communications are fairly stable and mobile cellular phones are typically used outside of centralized areas of Ghana. Communication are fairly easy due to the availability of the mobile communication network companies such as One Touch, Areeba and Tigo.

The infrastructure in the Kibi District is fairly well developed. The town of Kibi is a major regional centre with a population of some 12,000 (2013). Kibi is connected to the national electricity supply network, and hospital, postal and other community facilities are available. Extensive mining infrastructure is in place in all of the major gold producing areas of Ghana.

5.5 Land Use

The main land uses include secondary forest, subsistence and cash crop farming, and artisanal gold mining. Agriculture is the dominant economic activity of the area with about 60% of the labour population engaged in it (Atiwa District Assembly, 2012). The Birimian series forest ochrosol soil coupled with the moist deciduous forest vegetation enables the cultivation of crops and the rearing of livestock. Crops such as, plantain, cassava, maize, yam, cocoyam, oil palm cocoa and vegetables are cultivated in the districts.

6 HISTORY

6.1 Project History

The Kibi District is one of the oldest gold-producing districts in Ghana. Virtually all of the past gold mining activity has focused on alluvial gold deposits in the many river valleys throughout the Kibi area.

6.1.1 Historic Alluvial Mining

This section largely cites Rae et al. 2006.

Past gold mining activity mainly focused on alluvial gold occurrences in many river valleys throughout the area. Long before Europeans arrived, the local villagers mined the area for generations using the traditional pitting methods, to penetrate through 2 - 3 m of barren overburden into the underlying gravels, which are often quite rich in coarse gold.

The area was mentioned in early historical accounts as being dangerous to wander from well-established trails because of the myriad of pits throughout the district. The Akim district was very much coveted by the Ashantis in their rise to become a regional power over 200 years ago, because of its known wealth in gold, and unsurprisingly it was one of their earliest conquests.

Direct European interest in the area started mainly in the frenzied but short-lived gold rush that started in 1898. The 1902 concession map of the Gold Coast shows many small to quite large concessions covering virtually the entire Atewa Range and adjacent areas with most of the area under the control of Goldfields of Eastern Akim. Many of the concessions were concentrated along major streams coming off the range where extensive artisanal mining was evident.

The most famous of these areas was known as Pusu, a small village at the base of the northeast flank of the range, approximately 5km due north of the village of Asiakwa. Junner (1935) reported that Europeans started alluvial mining operations in this area in 1903 and continued intermittently until 1930. The area was known for coarse nuggets of gold and recorded production from the companies during the 1920s was over 8600 oz from about 298,000 m³ with a recovered grade about 0.5 g/m³ in the nearby Birim River dredging was attempted in 1904-1905 but it was unsuccessful.

6.1.2 Artisanal Mining

In February 1990 Kibi Goldfields International Ltd which shares a boundary on the eastern side of the Saaman concession carried out a pitting program involving ten reconnaissance pits, under the instructions of Minproc Engineers, a consulting company. From October 2006, vigorous illegal small-scale gold mining 'galamsey' activity for hardrock mining was ongoing at Abompe, also near the area. Extensive workings, including very numerous pits and underground stopes extend over their working area. This revealed the presence of mineralized sheared quartz veins hosted in phyllite.

The artisanal miners had identified some mineralized quartz veins and lodes in the Kibi district. As a result there is abundant evidence of small-scale mining activity i.e., the presence of shafts and pits.

6.1.3 Exploration

A London-based junior, Akim (1928), Ltd carried out some exploration and development work on the quartz vein at the Kibi Mine on the outskirts of the town of Kibi. It is likely that some of the other known vein occurrences in the area were also explored but apparently yielded little success.

When the exploration interest in Ghana picked up in the mid-1980s, properties in the area were applied for and granted to small Ghanaian companies, in some cases in partnership with foreign backers. Their interest was almost exclusively on alluvial gold. Two groups, Sun Gold International and Kibi Goldfields, acquired prospecting concessions on the north western and north eastern margins of the Atewa range.

The Sun Gold group of Chicago, USA was in partnership with the Akyem-Abuakwa Development Corporation and was granted two mining concessions (totalling 85 km²) in 1987 at Kwabeng and Pameng on the NW margin of the Atewa hills. Sun Gold was unable to finance the project and Shefford Resources of Toronto entered the picture in late 1988.

Shefford did an initial evaluation, which proved encouraging, and then went immediately ahead with a full test-pitting program to assess the resource potential. This was followed by Minproc Engineers of Perth, Australia who completed a Bankable Feasibility Study in 1989. The project was then completely restructured whereby the mining leases were put into a new company, Goldenrae Mining Company, and Sikaman Gold Resources of Toronto amalgamated with Shefford. Sikaman then brought in a senior partner, the London-based ITM Group, who had a great variety of business interest throughout Africa, including management of some alluvial diamond operations in Angola.

The ITM Group provided additional equity funding and assisted in arranging debt financing with Dutch and German banks. The project went into production on substantial alluvial resources at Kwabeng in late 1990 and, although the resource base was well-confirmed by subsequent mining, the project encountered a variety of technical and financial difficulties and it eventually had to be closed down in late-1993.

In the early 1990s, the EQ Resources group of Toronto also picked up a large concession (Apapam) on the eastern flank of the Atewa range, covering the drainage of the upper Birim River in the vicinity of the town of Kibi. EQ carried out a successful pitting program in cooperation with Goldenrae with the intention of setting up a satellite production unit under Goldenrae management, however the demise of Goldenrae left this project in limbo. The successors to Goldenrae were eventually able to secure this prospecting concession after it had lapsed.

In the late 1990s, a private Australian company (Sword Construction) briefly operated a small, skid-mounted alluvial plant close to Osino on the Kibi Goldfields concession. They carried out some bulk sampling, which apparently confirmed earlier grade estimates, but equitable financial arrangements could not be worked out with Kibi Goldfields and the work ceased.

Meanwhile, also in the early 1990s, the Kibi Goldfields group had brought in private financing by Canadian investors and commissioned a Pre-Feasibility Study by Minproc Engineers, which confirmed substantial alluvial resources close to Osino along a major tributary of the Birim River on the NE flank of the Atewa Range. Eventually a decision was made to purchase a dredge from Malaysia, which was transported to site, but funds ran short and the equipment remains unassembled on site.

There was another small alluvial operation (Narawa), which started up in the late 1990s, along the Birim River valley close to Osino. This also involved a small, skid-mounted plant but there are no details as to production levels, grades, etc. In any event, this was at a time when the price of gold was very low (less than 300 USD/oz) and the operation was short-lived.

Yet another group tried their luck, this time on the small concession held by the Asikam cooperative just north of Kibi. This project was taken over by Ashanti Goldfields who had inherited an alluvial gold plant through their takeover of the Midras Mining group who originally had alluvial prospects on the Ofin River, west of Obuasi. However, the Asikam concession had higher-grade resources, so a plant was set up close to the Birim River and mining started in the late 1990s. Midras produced 7,510 oz in 1998 but production dropped to 1,066 oz in 1999 and the operations ceased in that year. Again, it would seem that the low gold price was a major factor in the closing of this operation. It is also likely that gold production was too low to support an operation with the invariably high overheads associated with foreign management.

A local group, Bugudon Company Ltd, acquired a prospecting license in the early 1990s at the southern tip of the Atewa Range, in the general vicinity of Asamankese. This company formed a partnership with a Russian group with considerable experience in alluvial gold mining; they imported equipment from Russia that had apparently started some gold production but this was not sustained and the project did not proceed. There seems to have been an acrimonious parting of the ways between the local and foreign partner.

In 2006, there was a small alluvial operation owned by Med Mining immediately south of the Pameng lease. This was a dry mining operation that has been operating for about two years. The project is said to be financed by a Turkish group and managed by Ghanaian personnel.

6.2 Ownership

Xtra-Gold Mining's interest in the Kibi Gold Project was previously by way of a prospecting license granted by the Government of Ghana on March 29, 2004 covering a licensed area of 33.65 km². In May 2008, Xtra-Gold Mining applied to the Government of Ghana to convert the Kibi prospecting license to a mining lease (the "Apapam mining lease").

The Government of Ghana granted and registered the Apapam mining lease to Xtra-Gold Mining on the following terms and conditions.

- The Apapam mining lease is dated December 18, 2008 and is owned and controlled by Xtra-Gold Mining, as to a 90% interest; and is a registered subsidiary of Xtra-Gold Resources Corp.
- The remaining 10% free carried interest in Xtra-Gold Mining is held by the Government of Ghana
- The Apapam mining lease had a seven year term which expired on 17 December 2015.

An application to extend the lease for a further 15 years was duly submitted to the Mining Commission. The lease extension application also includes the area between the existing lease boundary up to the Forest Reserve boundary, which is termed the "Buffer Zone". The Ghanaian Minerals Commission's Technical Committee has certified the expansion of the lease area, which includes the Buffer Zone, to be in conformity with the official cadastral system under Ghana Mining Laws and, as such, is a proper extension of the current lease.

Xtra-Gold is currently awaiting formal approval of the lease extension. In the interim, under Ghanaian mining law, the existing lease continues to remain in full force and effect.

Xtra-Gold has been granted surface and mining rights by the Government of Ghana to work, develop and produce gold in the Apapam lease area (including the processing, storing and transportation of ore and materials)..

6.3 Exploration

6.3.1 Alluvial Gold Deposits

The early exploration work in this area by the Shefford/Sikaman team focused on carrying out extensive pitting to establish alluvial gold resources. These 1x1m pits on an initial 800 m x 100 m grid. Where potential was found a 200 m x 25 m grid was sampled. The pits were dug through 20-30 cm of the underlying weathered bedrock; some pits were up to 8 m depth but most were in the range 4-6 m with the overburden and gravel each being 2-3 m thick on average. Early testing indicated that the overburden was consistently barren so only the gravels were processed at the field site using mainly small portable units, which consist mainly of a shaking screen with strong water spray and underlying sluice with an astro-turf bedding that captures most of the gold. Samples representing 50 cm intervals in the pits were processed and the volumes of each interval were carefully measured prior to processing.

The concentrates of heavy minerals from the astro-turf are taken to a central laboratory at the Kwabeng camp and subsequently panned carefully by an expert who can usually extract virtually all of the gold by hand. The gold is then weighed and placed in a glass vial for storage and future reference. In some cases, mercury has been introduced into concentrates with a substantial amount of fine-grained gold; the mercury is then burned off under a well-ventilated hood and the small 'sponge' of gold (± traces of Ag) left from the amalgam is then weighed and stored. In the vast majority of cases, the gold in the Kwabeng, Pameng, and Kibi areas is sufficiently coarse and the technicians are sufficiently adept at recovering the gold that mercury is rarely needed.

6.3.2 Bedrock Gold Deposits

Before the exploration work conducted by Xtra-Gold, very little systematic exploration work for bedrock gold deposits has been conducted in the Kibi area since the 1930s.

Although the Atewa Range is best known for the extensive alluvial occurrences, the area also hosts quite a few quartz vein prospects, which have attracted some attention in the past. Numerous gold reefs (i.e., veins) were reportedly discovered during the course of this early alluvial mining with the most noteworthy of these lode gold prospects being located on and/or in close proximity to Xtra-Gold land positions, including the Clearing Reef (Kibi Mine) and Hill Reef (Gold Mountain) lying at the north-central extremity of the Apapam Concession. Although these lode gold prospects were reportedly worked or subjected to underground development by London-based mining syndicates in the early 1900s, it is unclear if they ever reached actual commercial production as there is no known gold production data available.

Cogill (1904) refers to minor prospecting work carried out on a narrow NW trending vein at Kibi and a shallow shaft that was sunk close to the town of Kwabeng on another vein. Geological traverses in the early 1900s revealed minor gold in coarse clastics, which were identified are probably being equivalent to the Kawere Conglomerate of the Tarkwa district (Junner, 1935) and gold was also reported to be in some of the laterite capping the range.

Very little systematic work has been done in the district on identifying bedrock sources of gold. In the mid- 1990s Ashanti Goldfields had a reconnaissance license covering much of the area. They completed an airborne geophysical survey (magnetics and radiometrics) but apparently did very little follow-up ground work. In view of the widespread occurrences of alluvial gold, along with generally favourable geological structures, the systematic evaluation of the lode gold potential is justified.

On the northern outskirts of the town of Kibi is a small quartz vein occurrence (Kibi mine), which was extensively explored in the 1920s (Annual Report of the Department of Mines, 1924, 1925). The work included a main shaft sunk to a depth of 48 m with levels driven at depths of 20 m and 46 m (Junner, 1935); the vein appears to contain fairly high but erratic values in gold. The gold is associated with pyrite, and the vein is generally quite narrow (usually from 10 to 100 cm wide) but has been traced for about 300m along strike.

Cogill (1904) reports work done on a large quartz vein in the immediate vicinity of the town of Kwabeng on the north western flank of the Atewa range at the beginning of the 20th century; the exact location is not known but it is believed to be just north of town. At surface, the vein is quite massive and wide (up to about 7-8m) and several shafts were sunk to test continuity and grade at depth. The vein strikes approximately ENE and dips 30-40° to the SE (Mining Yearbook, 1902-1903 for Kwaben Mines). The depth of the exploratory work was apparently at least 37 m but specific details are lacking; the vein is apparently quite patchy in gold values with some sections assaying at better than 1oz/tonne but overall grades were considerably lower. There apparently were plans to develop this into a small mine in 1903-04 but these plans were never realised and development work ceased a few years later. While carrying out alluvial prospecting in this area in the late 1980s, Goldenrae Mining stumbled across what is almost

certainly the same vein but all of the old shafts had collapsed. The nearby alluvial deposits show a slight increase in grade in this vicinity, no doubt because of the proximity to this vein.

6.3.3 Historical Exploration of the Apapam Concession

The general Apapam Concession area was first systematically explored in the late 1980s by the West Africa Resource Development and Investment Group Plc (WARDIG) who held a large tract of land extending from Pawtroasi in the south-west to Sajumasi in the north-east, encompassing the present Apapam ML and the Akim Apapam reconnaissance license application areas. In 1987 to 1988, RTZ Consultants Limited (RTZ Consultants) undertook preliminary exploration activities primarily designed to evaluate the alluvial gold potential. The work undertaken included sampling of quartz veining in alluvial test pits and along roads, and a compilation of the historical data on the Kibi Mine (Clearing Reef) lode gold prospect for WARDIG.

A zone of quartz veining characterized by isolated quartz stringers, lenses, and a 1m wide vein system was also located along the Kibi-Apapam road (Birim Valley) during the 1987-1988 reconnaissance exploration program. A 32 m long trench was excavated to further expose the vein system and 11 samples collected from the more prominent quartz veins/lenses. This sampling yielded less than detection limit and/or just above detection limit gold values in the <0.01 g/t to 0.025 g/t range. This veining occurrence was located and sampled by Xtra-Gold in 2006; with seven (7) samples yielding similar detection limit gold values. Sampling of four (4) greywacke-hosted quartz vein occurrences intersected by test pits along the Birim Valley yielded gold grades of between <0.01 g/t to 0.52 g/t.

In the late 1980s, most of the major valleys extending to the summit of the Atewa Range were subjected to geological mapping and stream geochemistry as part of a lateritic gold reconnaissance program conducted jointly by Sikaman Gold Resources and BHP Minerals Ghana Inc. This work was undertaken under a special permit issued by Mincom.

In the mid-1990s, Ashanti Goldfields Company Limited (now AngloGold Ashanti Ltd.) held a reconnaissance license covering much of the area. They completed an airborne geophysical survey (magnetics and radiometrics) but apparently did very little follow-up work. In the late 1990s, Ashanti Goldfields set up an alluvial processing plant on the banks of the Birim River on the Midras Mining concession located immediately north of the Apapam Concession.

6.4 Mineral Resource

6.4.1 Mineral Resource Estimate 2012

During July to October 2012 SEMS Exploration undertook a Mineral Resource estimate for Zones 2 and 3 of the Kibi Gold Project. The estimate based on all exploration drilling completed as of June 2012 is presented in Table 6.1**Error! Reference source not found.**. The Mineral Resource estimate was prepared in compliance with the Definitions and Guidelines of the Canadian Institute of Mining, Metallurgy and Petroleum (CIM Standards). All work was carried

out using Datamine[™] software. Data were verified in accordance with standard QA/QC procedures.

The database consisted of 256 drill holes covering sampling Zones 1 to 4 of the Apapam Concession. Ninety percent (90%) of these drill holes occur in Zones 2 and 3. Within Zones 2 and 3, five separate mineralised bodies (Big Bend, Double 19, East Dyke, Mushroom, South Ridge) have been defined by 190 drill holes of which 88% are diamond-drill holes.

The five targets cover a combined strike length of 1.6 km, with separations between the mineral resource areas varying from almost contiguous to 200m for four of the targets occurring in Zone 2 and the fifth target of Double 19 lying 500m to the southwest in Zone 3.

The 2018 – 2019 South Ridge drilling data was outsourced to Goldspot Discoveries Inc. (Goldspot) of Montreal, Canada for integration into an updated 3D geological model of the Zone 2 – Zone 3 Mineral Resource footprint area. The new 3D geological model was completed by Goldspot in late July 2019 with the detailed modelling geared towards the identification of prospective litho-structural gold settings to help guide upcoming resource expansion drilling efforts.

Table 6.1: Mineral Resource 2012 Declared in terms of the guidelines of the CIM Standards					
	Category	Tonnage (Mt)	Gold Grade (g/t)	Gold oz	
Dia Daniel	Indicated	2.73	2.44	213,000	
Big Bend	Inferred	0.51	1.60	27,000	
Double 19	Indicated				
	Inferred	0.61	2.43	48,000	
5 (5)	Indicated	0.65	3.03	64,000	
East Dyke	Inferred	0.08	4.37	11,000	
Mushroom	Indicated				
	Inferred	0.25	2.32	18,000	
South Ridge	Indicated				
	Inferred	0.90	1.48	43,000	
Total	Indicated	3.38	2.56	277,000	
	Inferred	2.35	1.94	147,000	

6.4.2 Mineral Resource Estimate 2021

Pivot and TECT collaborated to update the mineral resource estimate in 2021. A detailed review and investigation of structural readings from drill core, trench and outcrops was also undertaken by TECT. The body of work was aided by the use of digital mapping devices for field mapping and a Reflex IQ-Logger to obtain more precise and dense structural datasets. These datasets where integrated with 3D Geological/Target Modelling, exploration drillhole targeting and downstream resource estimation.

The geological work supporting this mineral resource estimate included additional trenching and drilling in Zones 2 and 3. Trenching was undertaken from October 2018 – June 2021 on Zone 1 – Zone 2 – Zone 3 to supplement the drilling of the targets already recognised. Fifty-six (56) trenches (1,577.7 sample metres) were excavated on the majority of the targets that form part of the mineral resource estimate.

A total of 212 diamond core drill holes totalling 25,198.55 m was drilled during the 2012-2201 reporting period. With most of the drilling,158 drill holes totalling 21,321.45 m (85%), completed from February 2018 – June 2021 on targets within the Zone 1 – Zone 2 – Zone 3 Mineral Resource estimate footprint area.

The Mineral Resource Estimate is reported in Table 6.2.

Table 6.2: Mineral Resource Declaration - 2021 Declared in terms of the CIM Standards Cut-off: Au 0.5 g/t						
Big Bend						
	Tonnage (t)	Density (t/m³)	Grade - Au (g/t)	Au (oz)		
Measured	-	-	-	-		
Indicated	6,472,000	2.78	1.48	307,400		
M+I	6,472,000	2.78	1.48	307,400		
Inferred	1,257,000	2.82	1.03	41,400		
		Double 19				
	Tonnage (t)	Density (t/m³)	Grade – Au (g/t)	Au (oz)		
Measured	-	-	-	-		
Indicated	1,584,000	2.62	1.38	70,400		
M+I	1,584,000	2.62	1.38	70,400		
Inferred	-	-	-	-		
		East Dyke				
	Tonnage (t)	Density (t/m³)	Grade - Au (g/t)	Au (oz)		
Measured	-	-	-	-		
Indicated	3,102,000	2.72	1.49	148,800		
M+I	3,102,000	2.72	1.49	148,800		
Inferred	1,128,000	2.84	1.19	43,300		
	Gate	House and Gold Mour	ntain			
	Tonnage (t)	Density (t/m³)	Grade – Au (g/t)	Au (oz)		
Measured	-	-	-	-		
Indicated	-	-	-	-		
M+I	-	-	-	-		
Inferred	2,366,000	2.76	0.79	65,200		
Mushroom						
	Tonnage (t)	Density (t/m³)	Grade – Au (g/t)	Au (oz)		
Measured	-	-	-	-		
Indicated	505,000	2.64	1.37	22,300		
M+I	505,000	2.64	1.37	22,300		
Inferred	-	-	-	-		

Road Cut						
	Tonnage (t)	Density (t/m³)	Grade - Au (g/t)	Au (oz)		
Measured	=	=	-	=		
Indicated	225,000	2.80	0.85	6,100		
M+I	225,000	2.80	0.85	6,100		
Inferred	-	-	-	-		
		South Ridge				
	Tonnage (t)	Density (t/m3)	Grade - Au (g/t)	Au (oz)		
Measured	-	-	-	-		
Indicated	2,005,000	2.70	1.07	68,700		
M+I	2,005,000	2.70	1.07	68,700		
Inferred	943,000	2.82	1.02	30,800		
	Total					
	Tonnage (t)	Density (t/m³)	Grade - Au (g/t)	Au (oz)		
Measured	-	-	-	-		
Indicated	13,893,000	2.73	1.40	623,700		
M+I	13,893,000	2.73	1.40	623,700		
Inferred	5,694,000	2.80	0.99	180,700		

6.5 Mineral Reserve

No Mineral Reserves have been declared.

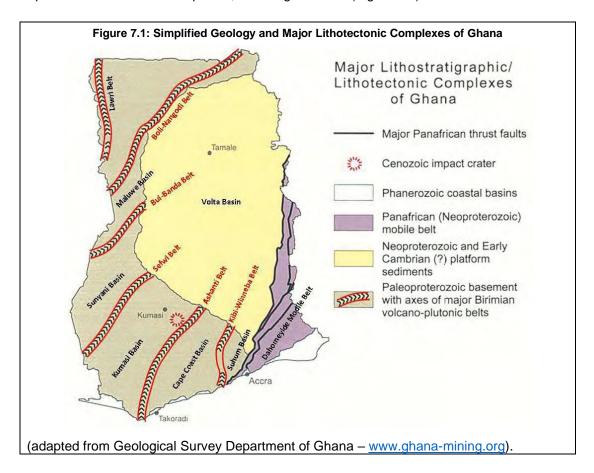
6.6 Historical Production

There has been no historical production.

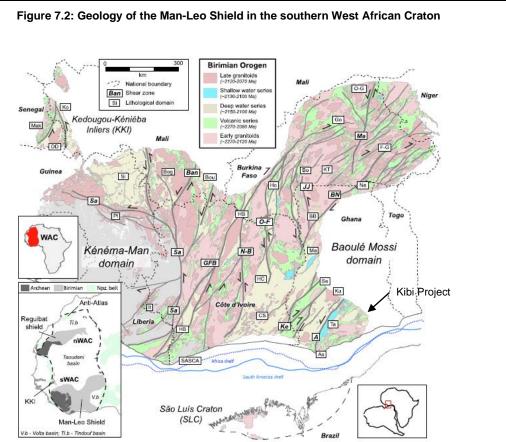
7 GEOLOGICAL SETTING AND MINERALIZATION

7.1 Regional Geological Setting

Xtra-Gold's Kibi Project is hosted within the Kibi Belt, which forms the northern continuation of the Paleoproterozoic Greenstone Kibi-Winneba Belt (KWB), located in south-western Ghana. Ghana's geology is generally described as part of the Guinea Shield of West Africa. It is broadly divided into four major lithotectonic terranes or settings, *viz.* Phanerozoic Costal Basins, Pan-African Mobile Belts (Neoproterozoic), Neoproterozoic to Cambrian platform sediments and Paleoproterozoic basement complexes, including the KWB (Figure 7.1).

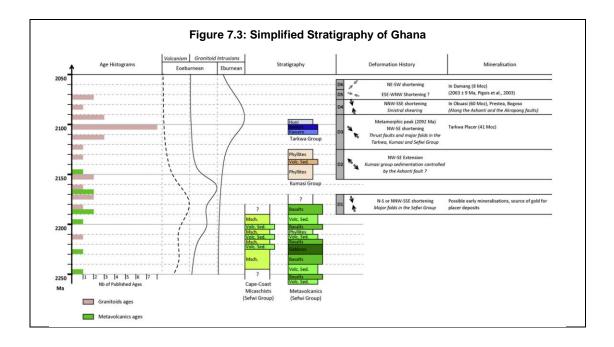


Paleoproterozoic Basement Complexes, on a regional-tectonic-scale, form part of the West African Craton (WAC) that is exposed as a series of inliers and shields located in the northern and southern sections of the craton, while the central portion is covered by largely Mesoproterozoic to Palaeozoic sedimentary cover sequences (Figure 7.2). The southern section of the West African Craton (sWAC) is assigned to the Man-Leo Shield, which comprises regional sequences of volcanic, volcanoclastic and siliclastic successions of the Birimian Supergroup, which were deposited in (sub-) basins immediately prior to and during the regional Eburnean Orogeny between ca. 2160 and 1960 Ma (Grenholm *et al.*, 2019, and references therein) – also referred to as the Birimian Orogen of West Africa.



Geological map of the Man-Leo Shield in the southern West African Craton (sWAC), including Birimian crust in the Paleoproterozoic Baoule Mossi domain Kedougou-Kenieba Inliers (KKI), Sao Luis Craton (SLC), and Archean crust in the Kenema-Man domain. Location of SLC after Klein and Moura (2008). Inset show location of the Man-Leo Shield in the West African Craton (WAC). Modified after Milesi *et al.* (2004). As – Ashanti series; Ta – Tarkwaian series; Ku – Kumasi series; Se – Sefwi series; CS – Comoe-Sunyani series; Ma – Maluwe series; Na – Nangodi series; BB – Bole-Bulenga domain; La – Lawra series; Bo – Boromo series; KT – Kedougou-Tumu domain; Go – Goren series; OG – Oudalan-Gouroul domain; OF – Ouango-Fitini Shear zone; HC – Haute Comoe series; HB – Hana Lobo–Bandama–Banfora series; It – Ity-Toulepleu inlier; BB – Boundiali–Bagoe; Bo – Bogouni; Si – Siguiri series; Ko – Kofi series; DD – Diale-Dalema series; Ma – Mako series. After Grenholm, *et al.* (2019)

In Ghana, the KWB forms part of a series of north- to north-east trending greenschist facies volcano-clastic to volcano-plutonic belts and sub-greenschist facies supracrustal basins, which are relics of the Birimian Orogen. These comprise stratigraphically upwards of the Sefwi Group (ca. > 2174 Ma) and Kumasi Group (ca. 2154 - 2125 Ma), both assigned to the Birimian Supergroup, and representing an early-tectonic pulse of volcanism and sedimentation. The Birimian Supergroup is, in turn, unconformably overlain by the Tarkwa Group (ca. 2107 – 2097 Ma) (also referred to as the Tarkwanian Supergroup), which represents a later tectonic pulse between ca. 2115 - 2080 Ma. Notably, this entire sequence is host to contemporaneous voluminous early- (ca. 2270 - 2120 Ma) to late-tectonic granitoids (ca. 2120 - 2070 Ma) (Figure 7.3).



The evolution of the regional Eburnean Orogeny (Perrouty *et al.*, 2012, and references therein; Figure 7.3) is described as transitional between two phases: firstly, an accretionary or converging plate-subduction phase (ca. > 2.16 Ga) (D₁) and secondly, an eventual plate-collision phase (D₃ – D₆) that peaked from ca. 2.16 to 2.07 Ga. The accretionary phase is largely underpinned by the deposition and deformation of the Sefwi Group, which describes early supracrustal mafic to felsic volcanism and the deposition of coeval volcaniclastic and siliclastic sequences in an intra volcanic arc setting, while lateral facies variations are attributed to the proximity of eruptive vents and distal marginal basins. In contrast, the overlying Kumasi Group comprises relatively deep water (low-energy deposition) phyllites and subordinate volcaniclastic sediments that reflects basinal extension or subsidence (D₂), localized within an overall converging plate-subduction setting (deep frontal- or back-arc basin setting?). The unconformably overlying Tarkwa Group, in turn, comprises predominantly shallow water sediments, which represents final basin inversion, towards eventual plate collision (viz. D₃ – D₆).

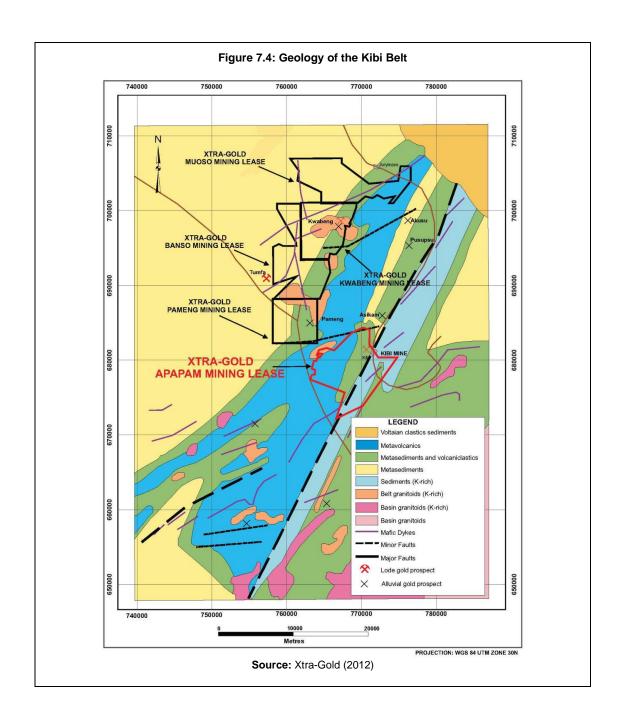
The margins of the KWB, along with the Ashanti-, Asankrangwa- and Sefwi Belts to the west, are bound by regional-scale major faults, which have throughout the Eburnean Orogeny, been subject to repeated reactivation. These major faults are inferred to have acted as thrusts during plate margin convergence, accretion and collision $(D_1 - D_3)$, but were in turn, re-activated to accommodate increasing lateral escape, during eventual plate margin collision and peak burial metamorphism $(D_3 - D_4)$. This gave rise to increased oblique-strike-slip kinematics (sinistral-reverse) along major faults (*viz.* D_4). A subordinate, late-tectonic sub-horizonal crenulation cleavage, cross-cutting earlier D_{1-4} structural features, is tentatively attributed to late tectonic loading (D_5) , while reverse faults associated with late-stage mineralization from ca. 2092 to 2063 Ma, cross-cutting all earlier features (D_{1-5}) , are assigned to D_6 .

7.2 Geology of the Kibi Belt

The KWB is the easternmost of the Birimian Greenstone Belts in southern Ghana. It is located east of, and parallel to, the well-endowed and world-renowned Ashanti Gold Belt, which hosts many of Ghana's active producing gold mines. The north-east trending Kibi Belt (KB), *viz.* the northern continuation of the KWB, is approximately 60 km long and 20 km wide. Its southern limit is truncated by a large granitoid batholith, whereas its northern extent is overlain by younger, flat lying sediments of the Pan-African Voltain Supergroup. Locally, the geology of the KB is poorly established, compared to the Ashanti Gold Belt, largely due to the poor exposures, limited government survey mapping and the lack of formal exploration activities. The general Kibi Project area geology is summarized from Griffis (1998) and Griffis *et al* (2002), and a Kibi Gold Belt Geological Map derived from regional geological survey traverses and airborne aeromagnetic and radiometric data (Figure 7.4).

The Kibi Project area is topographically dominated by the steep-sided Atewa Mountain Range exhibiting a relief of approximately 500 m with the surrounding valleys; showing its flat summits attaining an elevation of approximately 780 m above sea level. The Atewa Mountain Range is underlain by north-east trending Birimian Supergroup sequences (*viz.* Sefwi Group), including largely altered basalts and andesites (greenstones), interleaved with phyllites, meta-tuffs, epidiorite, meta-greywacke and chert. The broad valleys are underlain by thicker sequences of metasediments (greywacke, argillite, and phyllite), that are more susceptible to weathering and erosion.

The north-western extremity of the Atewa Mountain Range is the type-locality for the Birimian metasediments and metavolcanics. Regional traverses and airborne geophysical data indicate the presence of extensive volcaniclastics with narrower bands of mafic flows and mafic sills. Numerous, small, radiometrically inferred plutons appear to be emplaced within the belt, as well as several northeast-elongated bodies within the metasediments along the western margin of the belt. In general, granitoids associated with fold belts in southern Ghana are of dioritic to granodioritic (intermediate) composition, while granitoids in low-amplitude basins are more felsic and of granodioritic to granitic composition. Fold belt associated granitoids are suggested to have been emplaced as early sub-volcanic plutonism between 2179 and 2136 Ma (Hirdes *et al*, 1992) (D_1); while more felsic granitoids are associated with peak burial metamorphism and plate margin collision ($D_3 - D_6$). The belt exhibits several north-northeast to north-east trending major and secondary structures/faults that are conspicuous from airborne geophysical data and topographic patterns.



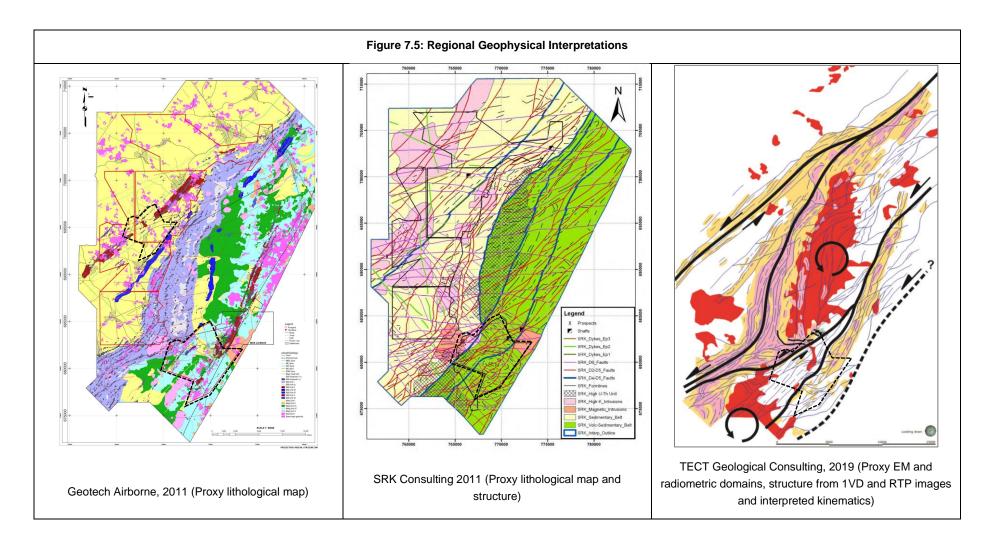
7.3 Geology of the Kibi Project

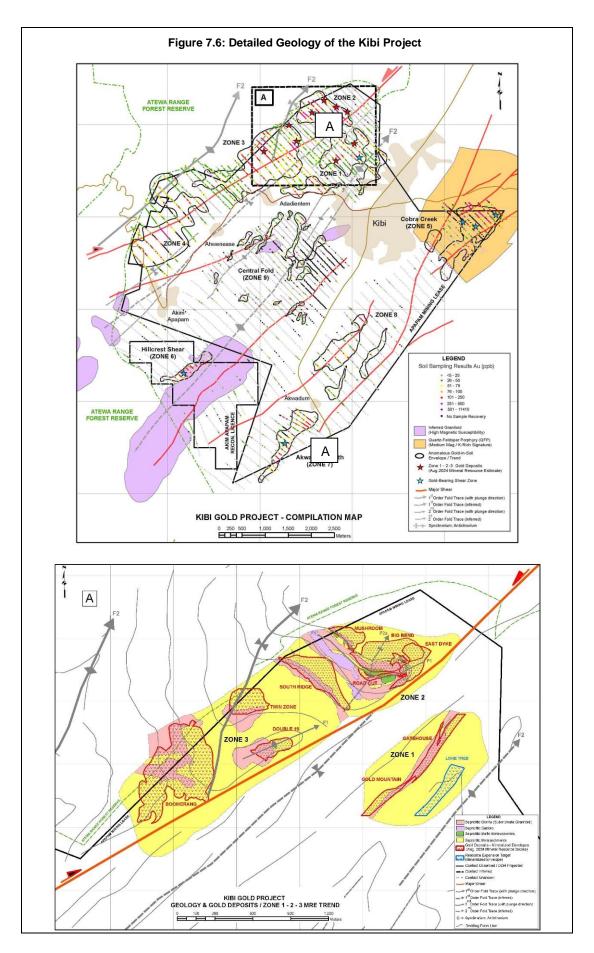
The knowledge and understanding of the geology of the Kibi Project area has been advanced significantly by Xtra-Gold's on-going exploration since 2006 which provided significant increases in mapping, trenching, drilling and soil geochemistry data in an otherwise poorly-explored region. Notable advances include, but are not limited to, interpretations of licence-scale (covering all of Xtra-Gold's concessions) airborne VTEM, aeromagnetic and radiometric surveys (Geotech Airborne, 2011; SRK Consulting, 2011 and TECT Geological Consulting, 2020 - 2024) spanning across all of Xtra-Gold's concessions (Figure 7.5). Moreover, field-based structural and lithological excursions and high-resolution 3D implicit geological and structural modelling on a target-scale on the Apapam Concession, targeting specifically Kibi Project Zones 1 – 5 (SRK Consulting 2010 and 2011; TECT Geological Consulting, 2019, 2020a, 2020b, 2023 and 2024), have been largely instrumental in ground truthing geophysical interpretations and deciphering licence- to target-scale geological settings and structural geometries.

Detailed geological mapping, constrained to Zones 1-5 (Error! Reference source not found.), located along the north-western (Zones 1-4) and north-eastern margin (Zones 5, viz. Cobra Creek) of the Apapam Concession, which includes all of Xtra-Gold's modelled targets (this report), largely comprises observations made from lateritic gravel cover, bedrock geology mapping, trench and road-cut exposures, in addition to drillhole observations. However, due to dense vegetation and the relative scarcity of surface exposures, coupled with relatively broad drillhole spacing, correlation of lithologies remains challenging across the concession.

In general, lithological mapping, trenching, and drilling indicates that Zones 1-4 is characterized by predominantly metasedimentary rocks interlayered with (saprolitic) granitoid to granodiorite and (saprolitic) mafic metavolcanic horizons (Error! Reference source not found.). This metasedimentary dominant sequence is at variance with historical geological maps (Figure 7.4), which depict a largely mafic metavolcanic sequence. Notably, apart from regional correlations, these rocks are largely undifferentiated with respect to the Berriman Supergroup and overlaying Tarkwa Group, as well as associated intrusives.

Dominant metasediments are characterized by thinly-bedded, medium- to coarse-grained greywackes with siltstone intercalations, and phyllites that are often graphitic. Interlayered mafic to intermediate metavolcanic units (basalt/andesite) and/or mafic sills (dolerite/diorite?) are characterized by massive, medium-grained textures, and typically range from 5 m - 30 m in thickness. Granitoid (granodiorite, quartz-diorite and tonalite) interlayers, previously sills and/or dykes (offshoots from larger pluton's?), are relatively thick, often > 40 m, and typically medium to coarse-grained.





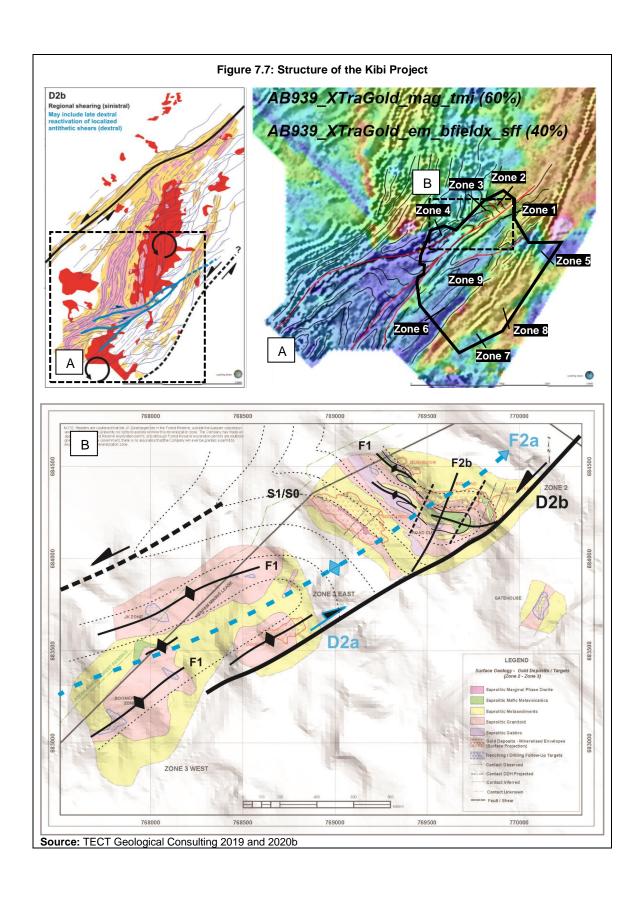
7.4 Structure - Overview

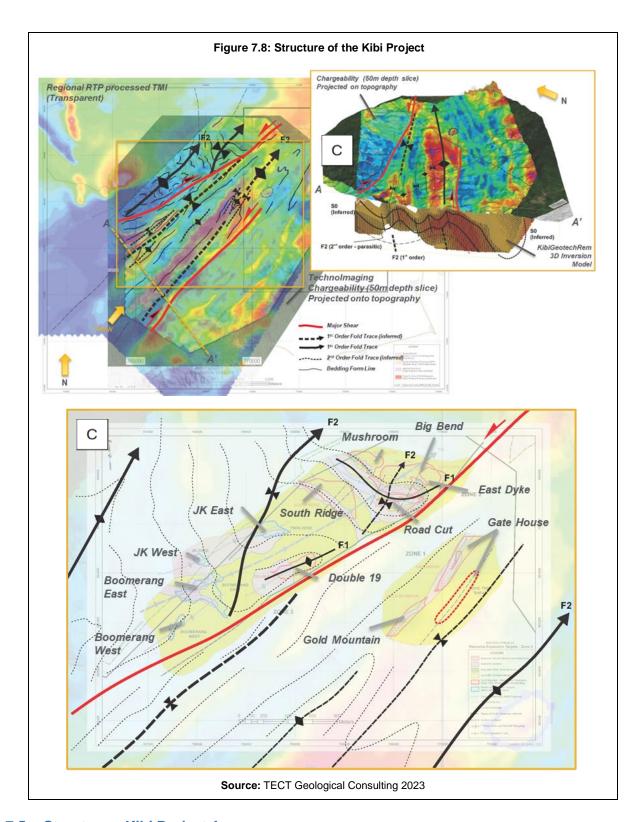
Undifferentiated metasedimentary and interlayered metavolcanics and granitoids are generally intensely deformed. A distinct $D_1(S_1)$ and $D_2(S_2)$ deformation event has been distinguished from aeromagnetic and radiometric datasets (e.g., SRK Consulting, 2012, TECT Geological Consulting 2019, 2020a and 2020b) and structural field observations. Notably, D_1 - and D_2 -associated feature recognition and the resolution of from the regional-scale geophysical interpretations, down to a target- or outcrop-scales is often difficult although general trends are evident. Indeed, the complexity of folds (tight to isoclinal) and shears that define the target-scale structural settings of the Kibi Project area, often result in regional-scale and geophysically-interpreted shears/contacts/lineaments being misconstrued. Regardless, a thus far consistent structural inventory, with interpreted kinematics, has been established (Figure 7.7 and Figure 7.8):

 D_1 (S_1) - A bedding (S_0) parallel foliation and associated F_1 isoclinal folds. Regional to licence-scale S_1 foliations are sub-parallel to the lithological contacts, which otherwise provide sufficient contrast on geophysical surveys. Associated features, including F_1 isoclinal folds (viz. shearfolds) that show fold widths varying from centimetre-scale up to 200 m, are only observed on a deposit-scale. Here, D_1 is consistent and readily correlated with the regionally defined D_1 , as described by Perrouty *et al.*, 2012 (Section 7.1, Figure 7.3).

 D_2 (S_2) - Regional to licence-scale NNE- to NE trending S_2 foliations are consistently observed and characterised as a pervasive and penetrative axial-planar cleavage to tight- to isoclinal F_2 folds. S_2 cleavages are particularly well-defined in metasediments, that are prone to preferential cleavage development, compared to more rheologically competent, interlayered metavolcanics and intrusions. Importantly, S_2 and associated F_2 folds, locally transpose and refold S_1 and S_2 foliations and folds, respectively. Moreover, regional to licence-scale shears, that otherwise constrain, internally dissect and segment and relict S_2 domains defining the Kibi Belt, are assigned to S_2 . In this context, S_2 , is tentatively compared/associated with regionally defined S_3 -reverse-sinistral kinematics, as described by Perrouty *et al.*, 2012 (Section 7.1, Figure 7.3).

Regionally defined D₅₋₆ structural features have not been observed (Section 7.1, Figure 7.3).





7.5 Structure – Kibi Project Area

The structural location of the Kibi Project area, i.e., the Apapam Concession, within the Kibi Belt is depicted in Figure 7.7. The north-western portion of the Kibi project (Zones 1-4) occupies a local antithetic dextrally sheared domain (D_{2a}), within an overall sinistrally sheared and

transposed Kibi Belt (D_{2b}). The central to south-eastern portion (Zones 5-9) occupies a largely sinistral domain, approximating the eastern margin of the Kibi Belt. The local D_{2a} antithetic kinematic behaviour is attributed to belt-scale asymmetrical boudinage of a relatively more competent metavolcanic sequence rimmed by less competent metasediments. Inter-boudin (antithetic) zones facilitate clockwise rotation of boudins in an overall belt wide sinistral shear setting or zone. Notably, however, D_{2a} local features are subject to gradual D_{2b} transposition and the reactivation of shears to eventual sinistral kinematics.

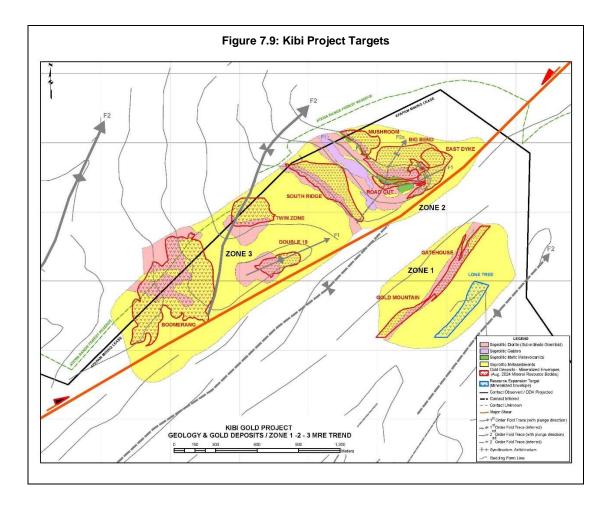
On a licence to deposit scale, Au mineralization in Zones 1-4, that are the subject of this Mineral Resource declaration and arguably Zones 6 and 9, is related to progressive tight to isoclinal F₁ and open F₂ fold development, which have widths ranging from of approximately 0.1 km (F₁) to 0.5 km (F₂) and amplitudes exceeding fold widths by 3-4 times (Figure 7.7). F₂ fold plunges are generally moderate to steep towards the NE, but may also be kinematically rotated against major D₂ shear planes/zones (e.g., Big Bend, JK East). Major D₂ shear zones are largely delineated by the interpretation of geophysical data and limited drillhole intersections, as well as major lithological or structural trend discontinuities (for e.g., the southeastern margins of Zones 2-4).

Relevant deposits (that are the subject of this Mineral Resource declaration) are predominantly hosted in diorite, or mafic to intermediate metavolcanic layers or sills/dykes, which are juxtaposed against a metasediment contact. From these characteristics several scales of structural control for auriferous mineralization may be ascertained for the Kibi Project Area:

- 1st-Order: Licence- to regional-scale F_{2a} and F_{2b} tight to isoclinal fold hinges.
- 2nd-Order: F_1 isoclinal fold anticlinal hinges that are parallel with D_1/S_1 foliation generally depicted by S_0 bedding contacts.
- Lithological contacts with a relatively high degree of competency contrast, e.g., dioritemetasediment or volcanic-metasediment contacts.
- Geochemically favourable hosts, such as intermediate to mafic intrusive units.

7.6 Mineralization and Deposits

The Kibi project is located on the Apapam concession and is thus far Xtra-Gold's only material project, which hosts the maiden Zone 2 – Zone 3 Mineral Resource estimate (October 26, 2012). This includes the Big Bend, East Dyke, South Ridge and Mushroom targets in Zone 2, and the Double 19 target in Zone 3 (Figure 7.9). Collectively, these five gold targets lying within approximately 1.6 km of each other have been estimated to encompass an indicated mineral resource of 3.38 million tonnes grading 2.56 g/t gold for 278,000 ounces of contained gold and an additional Inferred Mineral Resource of 2.35 million tonnes grading 1.94 g/t gold for 147,000 ounces of contained gold (@ base case 0.5 g/t cut-off). The Zone 2 – Zone 3 maiden Mineral Resource represented the first mineral resource generated on a lode gold project within the Kibi Gold Belt. It was filed in accordance with National Instrument 43-101 (NI 43-101) requirements. The Technical Report is entitled "Independent Technical Report, Apapam Concession, Kibi Project, Eastern Region, Ghana", prepared by SEMS Explorations and dated October 31, 2012, filed under the Company's profile on SEDAR at www.sedar.com.



Relevant gold mineralization is characterized by auriferous quartz-carbonate tensional vein arrays or stockworks hosted in either granodiorite, diorite and mafic- to intermediate metavolcanic layers or undifferentiated sills/dykes. This is geologically analogous to other intrusive- or volcanic-hosted gold deposits of Ghana, including Golden Star Wassa Mine in the Ashanti Belt, Kinross Gold's Chirano and Newmont Mining's Subika deposits in the Sefwi gold belt. These tensional vein arrays are generated where the abovementioned competent host lithologies unable to absorb preferential strain, leading to fracturing and auriferous mineralization. Such sites are typically localized along host-metasediment contacts that define F₁ and F₂ limbs, fold hinges or apparent flexures. Auriferous vein arrays are penetrative into metasediments, but with a very limited extent.

Hydrothermal alteration adjacent to the quartz-carbonate veins is highly variable, but in heavily veined (stockworks) granitoid the assemblage is characterized by moderate to strong quartz, carbonate, chlorite and sericite alteration. This is also associated with patchy to pervasive sulphidisation in the form of disseminated pyrrhotite, pyrite, and arsenopyrite (+/- sphalerite). Variations in alteration intensity in granitoid hosts is reflected in gold grade, with higher intensity alteration leading to higher Au grades, particularly encompassed in pervasive stockworks.

After the 2012 maiden Mineral Resource estimate, ongoing exploration has further delineated the Big Bend, East Dyke, South Ridge, Mushroom and Double 19 targets, and has also

identified new targets including Road Cut (Zone 2), Gate House (Zone 1) and Gold Mountain (Zone 1) – which were at that time only subject to early-stage exploration (Figure 7.9). Currently, alongside the abovementioned targets, early-stage prospects include Boomerang West and East (Zone 3), JK West and East (Zone 3), Cobra Creek (Zone 5), Hillcrest Shear (Zone 6) and Akwadum (Zone 7).

Since the 2021 Mineral Resource declaration, the exploration work and drilling has been focussed on the advancement of the Boomerang and Twin Zone targets in Zone 3. Boomerang has since been combined into a larger, structurally-related, gold complex, and now includes prospects previously assigned to Boomerang East and West, and JK West. JK East is now referred to as Twin Zone and is structurally unrelated to JK West (now assigned under Boomerang). Trenching was primarily undertaken in Zone 3, to delineate the near surface lithostructural and mineralization nature of the new Boomerang and Twin Zone mineral resource estimate bodies. The field exploration was combined with 3D geophysical modelling and lithostructural modelling, to further define the structural controls of the gold mineralization, and to generate high-priority exploration targets to help guide ongoing resource expansion drilling.

The respective structural inventory of each target is summarized as below. The 3D structural and geological modelling of each target was undertaken in Leapfrog GeoTM. The underlying reports are TECT Geological Consulting 2019 and 2020b.

7.6.1 Big Bend

- Big Bend target occupies an open F_{2a} anticlinal hinge zone, that plunges steeply 53° towards 078° (078°/53°) (estimated from 3D model and triangle mesh orientations, in turn informed by structural analyses and drillhole intersections). Big Bend's F_{2a} fold geometry was partially transposed by an apparent D_{2b} shear zone towards the east, thus kinematically rotating the plunge direction more towards the NW (anticlockwise). Notably, the South Ridge target is considered a direct analogue, with a plunge of 033° 48, albeit less transposed by D_{2b} (see below).
- Mineralization consists of tensional arrays of auriferous quartz-carbonate veins. Vein geometries may be divided into three principal sets. The dominant vein set ranges between 259°/34°, 306°/49 and 289°/28° (contoured maxima from Reflex IQ-Logger, kenometer and field measurements, respectively), and largely occupies extensional or shear extensional geometries with respect to a pervasive S₂ axial planar cleavage (133°/68°), confirmed by field mapping.
- Secondly, an equally intense vein set (040°/38°) is orientated sub-parallel to a S₁/S₀ bedding parallel foliation (previously recorded on downhole logs as cleavage) of 046°/60°, confirmed largely by IQL measurements. Lastly, and perhaps with the most varied in orientation, is a vein set ranging between 236°/34° and 162°/26° that striking sup-parallel to S₁/S₀ bedding parallel foliations, albeit dipping at opposite and high angles. Both orientations are consistent with D₁/S₁ flexural slip during folding, either occupying extensional to extensional-shear, or S1 (shear-) foliation-parallel geometries.

- It remains unclear which particular set of veins are predominantly mineralized, if not all of them. However, the context of the target-scale structural setting the spatial distribution of mineralization along the F_{2a} fold hinge, reasonably suggests that veins with a tensional relationship with S₂ axial planar cleavage, are preferentially mineralized. This rationale is supported by the smaller Road Cut target, a direct analogue to the immediate south, where similar vein geometries are recorded.
- Veining is restricted to the diorite-volcanoclastic sediment contact and internally within diorite. Due to its relatively high competency, diorite is prone to fracture to a greater degree, compared to volcaniclastic units.

7.6.2 Double 19

- Double 19 target occupies a tight to isoclinal F₁ anticlinal hinge zone. The fold plunge ranges between 067° 52° (derived from a fit to poles to measured bedding) and 084° 50° (modelled bedding). This is broadly consistent with axial-planar cleavages of 118°/57° (field mapping) and 221°/48° (IQL).
- Mineralization consists of tensional arrays of auriferous quartz-carbonate veins. Vein geometries from field mapping and Reflex IQ-Logger measurements are conspicuous and show a dominant set with contoured maxima varying between 279°/42° and 309°/29°. This vein set largely occupies extensional or shear extensional geometries with respect to a pervasive S₂ axial planar cleavage and shearing.
- Veining is restricted to the diorite-volcanoclastic sediment contact and internally within diorite. Due to its relatively high competency, diorite is more prone to fracture much more, compared to volcaniclastic units;

7.6.3 East Dyke

- East Dyke target occupies a closed to isoclinal F₁ anticlinal hinge zone, that is partly defined by an undifferentiated mafic sill or metavolcanic layer with variable lateral thickness (boudinage/pinch-and-swell). Mineralization is constrained to diorite-volcanoclastic and mafic sill-volcaniclastic contacts, which manifest in 3 sub-targets in the inner and outer arc of the F₁ anticline. The F₁ anticline plunges steeply towards 078° 63° (estimated from 3D model and triangle mesh orientations, in turn guided by the structural/kinematic analysis and drillhole intersections). Similarly to Big Bend, the initial F₁ fold geometry was partially transposed by a D_{2b} shear zone to the east, thus kinematically rotating the plunge direction towards the NW (anticlockwise);
- Mineralization consists of tensional arrays of auriferous quartz-carbonate veins. Vein geometries are difficult to gauge, due to limited data. The most reliable vein set, from mapping data, shows a contoured maxima of 281°/46°, and corresponds with extensional or shear extensional vein geometries with respect to a pervasive S₂ axial planar cleavage (133°/68°), as derived from Big Bend mapping data. Other apparent vein sets or contoured

maxima from drill core logging range between 345°/27° and 230°/32°, which determined to be partial representations/populations of the high-confidence mapping data.

 High-confidence mapping, viz. quartz vein geometries, should primarily be considered for estimation

7.6.4 Gold Mountain and Gate House

- Gold Mountain and Gate house are the least studied targets. Gate House comprises two
 inferred parallel shears hoisting auriferous quartz-carbonate veins, while Gold Mountain
 along strike to the south, comprises a single, similarly mineralized shear. Although with
 limited data, these apparent shears are interpreted to represent a single continuous shear
 zone.
- The respective shear zone strikes NE and records overall subvertical to steep SE dips (ca. 124°/84°). It is structurally located to south-east of a major D_{2b} shear zone and a juxtaposed F₂ isoclinal fold with extremely attenuated and possibly sheared limbs. This suggests that the mineralized shear facilitates F₂ limb-parallel shearing.
- Auriferous mineralization is predominantly hosted by granitoids (granite) and subordinate
 diorite, undifferentiated mafic units and metasediments. It is inferred that shearing or
 shears, straddle and dissect more competent above-mentioned units that are otherwise
 more prone to fracturing, followed by fluid ingress and Au mineralization.
- More detail structural analysis is required to confirm shear zone kinematics, mineralized vein orientations and the overall depositional setting.

7.6.5 Mushroom

- Mushroom target occupies a tight F₁ anticlinal hinge zone that plunges steeply at 063° 58° (estimated from 3D model and triangle mesh orientations).
- Mineralization consists of tensional arrays of auriferous quartz-carbonate veins. Vein measurements are limited, but contoured maxima are 162°/26° and 302°/43°. These orientations are consistent with pure extensional to extensional-shear fracture/veins geometries that dissect So/S1 contacts at sub-normal angles.
- Veins are restricted to the diorite-volcanoclastic sediment contacts and internally within diorite. Here, due to its relatively high competency, diorite is prone to more fracturing, compared to volcaniclastic/metasediments units. Diorite is also a geochemically favourable host due to its relatively high Fe-Mg content, which may cause precipitation of Au and pyrite/pyrrhotite (depending on metamorphic conditions).

7.6.6 Road Cut

- Road Cut target occupies an open F_{2a} anticlinal hinge zone, that plunges parallel to Big Bend, steeply towards 078° - 53°. Downhole structural data is unfortunately limited for Road Cut, but it is structurally identical to Big Bend.
- Mineralization consists of tensional arrays of auriferous quartz-carbonate veins. Vein geometries from field mapping are conspicuous and show a dominant set at 306°/47°. This vein set largely occupies extensional or shear extensional geometries with respect to a pervasive S₂ axial planar cleavage (111°/65°), confirmed by field mapping.
- Veining is restricted to the diorite-volcanoclastic sediment contact and internally within diorite. Due to its relatively high competency, diorite is prone to fracture much more, compared to volcaniclastic units.

7.6.7 South Ridge

- South Ridge occupies a more pristine open F_{2a} anticlinal hinge zone, with no apparent transposition by the D_{2b} shear zone that has affected Big Bend and East Dyke. The fold plunges at 033° 48°, which is broadly consistent with an S₂ (axial planar cleavage) contoured maxima of between 119°/61° (field mapping) and 095°/55° (IQL) . High resolution Reflex IQ-Logger readings are limited to only one drillhole, whilst field mapping data is spatially much more representative.
- Mineralization consists of tensional arrays of auriferous quartz-carbonate veins. Vein geometries from field mapping are conspicuous and show a dominant set at 282º/33º. This vein set largely occupies extensional or shear extensional geometries with respect to a pervasive S₂ axial planar cleavage (119º/61º), confirmed by field mapping;
- Veining is restricted to the diorite-volcanoclastic sediment contact and internally within diorite. Due to its relatively high competency, diorite is prone to fracture much more, compared to volcaniclastic units;

7.6.8 Twin Zone (previously JK East)

• The Twin Zone, or JK East, deposit occupies an open to tight F₂ synclinal hinge zone. The fold plunge is estimated from modelled bedding contacts, viz. 073° - 46°, which is broadly consistent with a 1st order F₂ syncline trace across Zones 2-3. Au mineralization largely comprises extensional (brittle) vein-arrays within granodiorite, that was unable to accommodate extensional and compressional stress.

7.6.9 Boomerang (Previously Boomerang East and West, JK West)

 With ongoing exploration Boomerang East and West, and JK West developed into a structurally coherent gold complex. Several sub-deposits, structurally parallel to each other, have been distinguished (JK1-11, and Boomerang West, which have been deemed economic).

- JK1-11(sub-deposits) occupy the north-western limb of an apparent open (meso-scale) F2 synclinal hinge, plunging approximately 053° 33°. Within JK orebodies the available downhole structural data suggest that quartz veins and Au mineralization are hosted within subhorizontal tension fractures within the broader auriferous structures, similar to Big Bend and South Ridge. Similarly, the hosting competent granodiorite was unable to accommodate compressional stresses (folding) and strain, and consequently formed extensional (brittle) vein-arrays.
- The JK West auriferous structures are largely concentrated within the more competent diorite
 and granite bodies, however some of broader auriferous structures do transgress through into
 the volcano-sedimentary units.
- JK1-11 Au grade shells seem to be roughly parallel the overall foliation / schistosity of the F2
 mesoscale fold (Figure 16), with the structures likely occupying the axial planar foliation of the
 fold.
- deposit parallels the axial pane of an inferred NNW-SSE-trending F1 hinge zone. It is contained within the south-eastern limb of a 1st-order F2 anticlinorium (slides 17 and 29) that is partially transposed by a major D2 shear zone that is juxtaposed to the SE.

Boomerang West (sub-deposit) deposit parallels the axial pane of an inferred NNW-SSE-trending F1 hinge zone. It is contained within the south-eastern limb of a 1st-order F2 anticlinorium (slides 17 and 29) that is partially transposed by a major D2 shear zone that is juxtaposed to the SE.

8 DEPOSIT TYPES

8.1 Hydrothermal Gold Deposits of Ghana

The deposit types being targeted at the Apapam Concession consist of mesothermal or orogenic gold mineralization of the granitoid-hosted type and classic Ashanti-style sediment-hosted shear zones; that are likely to be linked with a major northeast-trending D₃₋₄ (after Perrouty *et al.*, 2012, and references therein; Figure 7.3) major fault along the eastern flank of the Atewa Range. At the present, the largely diorite-hosted type accounts for the majority of the identified gold occurrences of potentially economic significance on the concession and is consequently the current focus of Xtra-Gold's exploration efforts. However soil geochemistry, prospecting and geophysical data interpretation, and historical auriferous quartz vein showings, indicate that the concession is also prospective for Ashanti-style shear zone gold mineralization.

8.1.1 Shear Hosted Gold Deposits of Ghana

Characteristics of the Ashanti-style shear zone hosted gold deposits are described as follows by Naas (2008). Mineralization associated with major D₃₋₄ faults (after Perrouty *et al.*, 2012, and references therein) or major belt bounding faults was the target for both local prospectors and foreign exploration companies moreover due to the presence of coarse-grained visible gold. Deposits of this type in Ghana include Obuasi, Prestea, Bogosu, Konongo and Bibiani. There are several commonly observed associations with this mineralization environment, which include:

- Located on, or close, to the lithological contact between greenstones and metasediments.
- Spatially related to deep-seated, high-angle wrench faults, which have a strike extent
 exceeding 100 km. Cross-cutting northwest-southeast trending faults have also exerted
 an influence on the location of gold remobilized from the main zones.
- Native gold is hosted by quartz veins, which may possess an en-echelon character.
 Grade-width characteristics persist virtually unchanged to depths exceeding one (1) km.
 The veins broadly parallel the regional foliation but in detail are seen to cross-cut this foliation.
- Disseminated sulphides in the wall rock are common.
- Several generations of quartz veining are common, and gold is seemingly associated with the final phase.
- Mineralization is spatially associated with graphitic phyllites and manganiferous sediments
- Mineralogy is simple with a strong positive correlation between gold and arsenopyrite. Accessory minerals include pyrite, chalcopyrite, pyrrhotite, and bornite.
- Strong silicification is common, accompanied by sericite and carbonate alteration.
 Tourmaline may also be present.
- Granitoids may or may not be spatially associated with mineralization.

8.1.2 Granitoid-hosted Gold Deposits of Ghana

Over 20 significant gold occurrences hosted by Belt and Basin-type granitoids are known in Ghana, with a number of these constituting significant deposits. The structural setting and mineralization style for Belt and Basin granitoid-hosted gold deposits are very similar in nature to the Ashanti-style, and probably just represents a different structural and host rock setting, accordingly. These deposits represent a subtype of the orogenic gold deposits of the Ghanaian Birimian terrane. Belt-type intrusion hosted gold deposits include Newmont Mining's Subika deposit, the largest deposit of the Ahafo mine project, and Kinross Chirano deposits, both in the Sefwi Belt; and Golden Star Resources' Hwini-Butre deposit at the southern extremity of the Ashanti Belt. Basin type granitoid hosted gold deposits include Perseus Mining's cluster of deposits at the Central Ashanti Gold Project, and AngloGold-Ashanti's Anyankyerim and Nhyiaso deposits to the west of Obuasi, along the western flank of the Ashanti Belt.

As opposed to the typical lode gold deposits of the Ashanti, Prestea and Bibiani districts, which were (re-)discovered by Europeans during the gold rush of the late 1800s, all of the aforementioned granitoid-hosted gold deposits have been discovered since 1990. Tectonically, the host intrusive bodies lie within or proximate to reactivated regional structures and have deformed in a brittle >>> ductile fashion. In terms of lithology the Belt-type granitoids are most commonly of diorite to granodiorite composition, and the Basin-type granitoids of granodiorite to granite composition. Enhanced brittle deformation of granitoids, internally and along contacts, of granitoids, appear to have served as preferential conduits for fluid flow (due to their relatively high competency compared to their metasedimentary hosts). The emplacement of granitoidhosted mineralization is considered contemporaneous with the main gold mineralizing episode that resulted in the more prevalent Ashanti-type Birimian metasediment/metavolcanic shear hosted deposits of Ghana (Figure 7.2). The mineralization typically consists of quartz vein stockworks and pervasive alteration zones developed in brittle structures in the granitoids. The ore mineral assemblage is mainly composed of pyrite, pyrrhotite and arsenopyrite, with minor chalcopyrite, sphalerite and rutile. Hydrothermal alteration minerals are dominated by quartz, sericite (muscovite), sulphides (mainly pyrite, pyrrhotite, and arsenopyrite) and carbonates. Gold tends to be closely associated with the sulphides in both quartz veining and alteration zones.

8.2 Gold Deposits of the Kibi District

Primary gold mineralization of potentially economic significance discovered to date on the Apapam Concession by Xtra-Gold (Sections 7.5 and 7.6.) consists predominantly of mesothermal/orogenic gold mineralization of the granitoid-hosted type. The gold is associated with quartz albite-carbonate-sulphide stockwork or tensional veining developed in or especially near the margins of sills, dykes, and possibly small plutons (stocks) of granodiorite, quartz diorite, and tonalite bodies. Possible primary shear zone hosted gold mineralization (e.g., Zones 1, 6-8 and 9) does occur within the concession, but this is still subject to the relatively early stages of exploration.

9 EXPLORATION

9.1 2006 – 2007 Exploration Program

Two (2) separate work programs were conducted on the Apapam Concession during 2006-2007. The first work program was undertaken and managed by CME Consultants Inc. (CME); a Canada-based geological consultancy with over 15 years of project management experience in Ghana. The second program was undertaken and managed by Xtra-Gold personnel.

The Phase I exploration program was designed to test the Apapam Concession on a regional scale. The field work was implemented by CME from August 12 to September 23, 2006 and included:

- Concession-wide stream sediment sampling (88 samples collected from 44 sites);
- Survey grid establishment (33.78 line-km);
- Soil sampling (1,306 samples);
- GPS surveying (33.78 line-km);
- Rock sampling (89 samples); and
- Historical adit and bulldozer cut sampling (100 samples).

The Phase II exploration program consisted of a reconnaissance trenching program intermittently implemented by the Xtra-Gold exploration staff from February 2007 to December 2007. The trenching was carried out to test the geochemical signature at depth of the gold-insoil anomalies detected within the north-western portion of the concession during the Phase I work program. A total of 542 channel samples were collected from 21 trenches totalling 1,090 linear metres. In order to obtain an independent assessment of the 2007 Xtra-Gold trenching results, a NI 43-101 compliant data verification program was undertaken by CME in December 2007. The program involved the re-sampling of selected trenches which yielded exploration-significant gold mineralization intervals.

9.1.1 Phase I Exploration Program (2006)

Stream Sediment Sampling

A total of 88 samples were collected at 44 samples sites from two (2) major streams and their respective tributaries. These included 44 silt samples for geochemical analysis (Bulk Leach Extractable Gold - BLEG) and 44 pan concentrate samples for visual gold grain counts.

Stream sampling returned gold-in-silt values of up to 710 ppb located at the western extremity of the concession. Values greater than the threshold value (mean + 2 standard deviations) of 144 ppb are considered to be anomalous. Five (5) samples yielded values greater than the threshold value. Gold grain counts of the pan concentrates showed visible gold grains in 36 of the 44 samples ranging from two (2) small grains per sample up to 16 flakes. Grain sizes varied from flour to over 3 mm.

The stream sediment anomalies are divided into two (2) zones as follows:

- Zone A (Adansu Anomaly): consists of a 1.5 km stretch from Line 158+00N westwards at an average with of 1.0 km along the north western boundary of the concession; and
- Zone B (Kokorabo Anomaly) is 3.0 km long by 2.0 km in width sector covering the area between the south-western boundary of the concession and the floodplains of the Birim and Krensen rivers.

Soil Geochemical Sampling

A total of 1,306 soil samples were collected from 30.58 line-kilometres of cross lines established within the Kibi North and Kibi South survey grids. Grid location was based on testing historical mineral occurrences located on and around Kibi Mountain (Kibi North Grid) and promising silt samples results from creeks southeast of Kibi (Kibi South Grid). Line spacing was at 100 m intervals within the Kibi Mountain area of the North Grid and 400 m elsewhere within the gridded areas. Soil samples were collected at a depth of 60 cm at 25 m intervals along the SE-NW trending grid lines. Soil samples were by analyzed by Fire Assay and reported in parts per billion (ppb).

The geochemical soil survey conducted on the Apapam Concession produced several interesting gold-in-soil anomalies. A geochemical trend of 050° to 060° (NE-SW) is in conforms to the regional geological trend. The highest gold value from the 1,306 samples was 1,413 ppb, located at L166+00N/60+25E. A total of 105 samples produced gold-in-soil values greater than a threshold value (mean plus two (2) standard deviations), of 98 ppb gold.

Rock and Historical Adit Sampling

A total of 89 rock samples were collected during stream sediment and soil sampling traversing. In addition, three (3) historical adits ranging from 7 m to 85 m in length and a bedrock face exposed along a bulldozer cut, located on and in the vicinity of Kibi Mountain, were also sampled during the Apapam Phase I exploration program. A total of 77 samples were collected from the three adits and 23 samples from the bulldozer cut face.

Seven (7) out of the 89 rock samples returned values greater than the threshold value of 82 ppb gold. The highest gold value recorded is 1.01 g/t, with the remaining anomalous values falling in the 140 ppb to 970 ppb gold range. No significant gold values were returned from the Adit 1 and Adit 3 sampling, but the six (6) samples collected from Adit 2 yielded economically significant values between 710 ppb and 6.36 g/t gold from chip and channel samples.

Rock (float) and adit sampling has also confirmed that a potential for lode gold mineralization exists on the Apapam Concession, especially in the vicinity of Kibi Mountain. Three (3) of the anomalous rock samples are located at the base of Kibi Mountain (1.01 g/t Au, 255 ppb Au and 385 ppb Au) and also float sample (510 ppb Au) is located within the Area 4 gold-in-soil anomaly, thus confirming mineralization within the vicinity. Results returned from rock samples indicated that float of fractured and limonitic rock can be a useful tool for future prospecting programs.

9.1.2 Phase II Exploration Program (2007)

A total of 21 reconnaissance trenches ranging from 2 m to 224 m in length were excavated by Xtra-Gold personnel during the 2007 Phase II exploration program. A total of 542 channel samples (2 m) were collected from the 21 trenches, totalling 1,090 linear metres. Trenches were manually excavated by pickaxes and shovels to a typical width of 1 m and an average depth of 3 m, with some sections of the trenches reaching 4 m in depth. Trenching typically extended down to the saprolite horizon but locally, the saprolite could not be reached due to safety concerns. Sampling consisted of a continuous channel sample collected from a channel excavated along the trench floor. Prior to sampling, the floor of the trench was cleaned of any loose material and an approximately 10 cm wide by 2-3 cm deep channel excavated along the centreline of the base of the trench.

The bulk of the trenching efforts, including eight (8) trenches totalling 834 linear metres (approx. 75%), focused on testing the Area 1, 2 and 3 gold-in-soil anomalies detected during the 2006 Phase I work program. Eight (8) trenches totalling 144 m were excavated to test the subsurface in an area of extensive Ashanti-style pits discovered by prospecting in what is now the north-central portion of the Zone 3 gold-in-soil anomaly. An additional five (5) trenches totalling 112 m were excavated to test the subsurface in areas of mineralized rock floats. Four (4) out of the 21 trenches yielded length-weighted average grade intervals greater than the arbitrarily set exploration-significant threshold of 1.0 g/t gold.

In order to obtain an independent assessment of the 2007 Xtra-Gold trenching results, a NI 43-101 compliant data verification program was undertaken by CME in December 2007. The program involved the re-sampling of selected trenches which yielded exploration-significant gold mineralization intervals. The trenching program results noted hereunder correspond to the results returned by the independent CME data verification program. The CME re-sampling included 116 channel samples totalling 115.41 linear metres. Sampling consisted of a horizontal channel cut along the sidewall of the trench, approximately 0.2 m above the trench floor. Sampling was typically established at one (1) metre intervals, with sample lengths locally adjusted to accommodate geological features. Forty-six (46) out of the 116 channel samples collected by CME returned values greater than 1.0 g/t gold. The reported mineralized intercepts represent trench lengths and are not necessarily indicative of the true width of the mineralization.

Gold mineralization on the Apapam Concession was found to occur in several different geological setting, including steeply-dipping and flat-lying quartz veins and alteration haloes proximate to the quartz veining. The presence of shallow dipping (*viz.* flat lying) veins may produce an exaggeration in both the width and grade of the mineralization. This is estimated to represent a true width of 3 to 4 m due to the flat lying nature of the quartz veins. Determination of true widths within trenches can be difficult, as not all of the geological features are properly exposed.

Trenching was found to be an effective way to test gold-in-soil anomalies on the Apapam Concession. The following best practices sampling techniques were recommended for future trenching programs:

- (1) Channel samples should be taken from the side wall of the trench and not from the floor of the trench in order to mitigate contamination issues and eliminate sampling bias when sampling exposed, shallow dipping quartz veins.
- (2) Sampling must be constrained by alteration, structure and lithology.
- (3) If two (2) metre sampling widths are to be used for budget reasons, detailed sampling of anomalous areas must be undertaken as follow-up.

9.2 **2008 – 2010 Exploration Program**

Exploration work on the Apapam Concession during the 2008-2010 reporting period was aimed at advancing the Kibi Project which consists of over 5.5 km long mineralized trend delineated from gold-in-soil anomalies, trenching, drilling, and geophysical interpretations along the northwest margin of the Apapam Concession; and characterized by widespread gold occurrences of the granitoid hosted-type.

An extensive soil geochemistry survey, covering approximately 47 line-kilometres (1,827 samples), was implemented in early 2008 to further define the extensive gold-in-soil trend. The entire K grid was also covered by IP/Resistivity (~ 64 km) and ground magnetometer (~79 km) surveys to help define the lithological and structural pattern of the mineralized trend, and prioritized trench/drill targets.

Exploration activities in 2008 also included a manual trenching program encompassing 18 trenches totalling approximately 1,217 linear-metres, including: 4 trenches (302 m) on Zone 2; and 14 trenches (915 m) on Zone 3 of the 5.5 km long gold-in-soil trend. In addition, 67 excavator dug trenches totalling approximately 2,223 m were also excavated in conjunction with the 2008 and 2009 drilling programs.

As part of the ongoing exploration efforts Xtra-Gold commissioned SRK Consulting (Canada) Inc (SRK) to conduct a structural study of the Apapam Concession. The goal of the study was to investigate key exposures and available drill core to document and understand the structural controls on gold mineralization at the Kibi Project. SRK reviewed 14 diamond drill holes (Zone 1 and 2) as well as available trench exposures (Zone 2 and 3) on the Apapam concession from March 16 to 27, 2010. Due to diamond drilling density and accessible trenches, SRK's structural study focused largely on Zone 2 of the Kibi Project. SRK also reviewed Xtra-Gold's geological and structural mapping to date for zones 1, 2 and 3 of the Kibi Project.

A petrographic study was also implemented in March 2010 to characterize the lithological units and ore mineralogy of the Kibi Project. A total of 36 thin sections and nine (9) polished sections were studied by Professor K. Dzigbodi-Adjimah of the University of Mines and Technology, Tarkwa, Ghana. The findings of the structural and petrographic studies are incorporated in the

property structure (Section 7.4) and mineralization (Section 7.6) of the technical report, respectively.

9.2.1 Soil Geochemistry

In early 2008, the Kibi Project grid was expanded to provide control for follow-up soil sampling and geophysical surveys. A total of 54.45 line-kilometres of crosslines (sample lines) and 2.1 km of baselines were established. The expanded grid now covers the entire north-western portion of the concession with a total of 78.8 line-kilometres of SE-trending crosslines extending along a 6.1 km baseline.

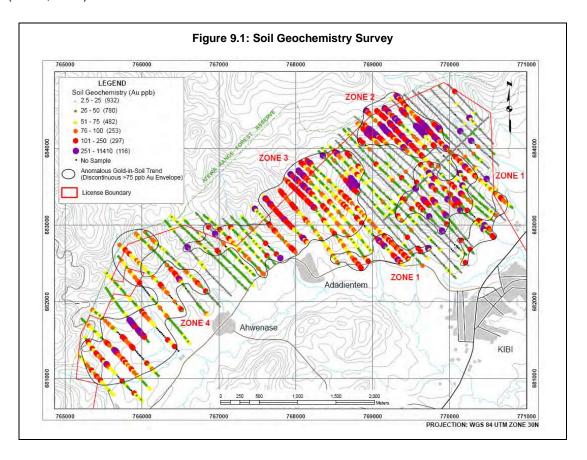
An extensive soil geochemistry survey was undertaken to provide detailed (100 m) soil sampling coverage of the gold-in-soil anomalies yielded by the Phase I (2006) work program and the bedrock gold occurrences identified in the 2007 trenching, and for reconnaissance (200 m) soil sampling to follow-up on anomalous gold-in-silt samples identified in streams at the south-western extremity of property during the 2006 regional exploration program. A total of 1,827 soil samples were collected at 25 m station spacing along 46.975 line-kilometres of cross lines. Including the Phase I (2006) work program, a total of 2,859 soil samples have been collected on the Kibi Project.

Regolith development in most of the Kibi Project area is favourable for soil sampling. The generally steep topography along the flank of the Atewa Range has resulted in relatively thin colluvial (lateritic gravel) cover in the project area. As a result, gold-in-soil anomalies on steeper slopes and ridges probably reflect a good, although not exactly quantitative, measure of gold distribution in the underlying saprolite. Similarly soil sampling has been primarily completed across areas of stronger, positive, topographic relief where alluvial gold deposits are unlikely to have developed. On very steep slopes, the anomalies show some asymmetry due to downslope dispersion, however the core of these anomalies are not significantly displaced downhill from the source. This close reflection of saprolitic bedrock gold distribution associated with gold-in-soil anomalies developed on steep slopes is demonstrated by trench TAD019, located at the south-eastern extremity of the Zone 3 gold-in-soil anomaly; which returned a channel sample intercept of 4.93 g/t gold over a 45 m trench-length, including 10.12 g/t gold over 12 m, from a 75 m long, 620 ppb to 2,280 ppb deposit gold-in-soil anomaly. In areas exhibiting less relief and more extensive development of laterite, the resulting geochemical patterns tend to be characterized by much broader dispersion haloes that produced gold-in-soil anomalies reaching 200 m or more in width.

The anomalous threshold for the soil sample results was arbitrarily set at 75 ppb gold based on past exploration experience by Xtra-Gold in the Kibi Greenstone Belt. A total of 666 (23%) out of the 2,859 soil samples returned gold values greater than the 75 ppb anomalous threshold, including: 253 (9%) samples from 76 ppb to 100 ppb gold; 297 (10%) samples from 101 ppb to 250 ppb gold; and 116 (4%) samples above 251 ppb gold (11,410 ppb Au maximum). The expanded soil survey outlined an approximately 5.5 km long, NE-trending, anomalous gold-in-soil trend (Figure 9.1). The typically NE-trending clusters are defined by discontinuous/patchy,

> 75 ppb gold, anomalous gold-in-soil envelopes ranging from 50 m to 250 m by 900 m to 250 m - 1,200 m by 2,500 m in area.

The gold-in-soil anomalies are considered significant based on the fact that in Ghana soil geochemistry values greater than 50 ppb, gold can normally be considered anomalous (Griffis, 2002). For instance, in the Obuasi gold camp, AngloGold-Ashanti reportedly follows up all soil anomalies greater than 50 ppb gold with trenching or drilling. At the Ahafo mine project, gold-in-soil anomalies in the 100 ppb to 200 ppb range led to multimillion ounce gold discoveries (Griffis, 2002).



The best indication that the anomalous gold-in-soil trend may be considered to be significant is that the trenching and/or drilling of soil anomalies has yielded notable, saprolitic bedrock, gold intercepts in three (3) of the main four (4) anomalous gold-in-soil clusters/envelopes (i.e., Zone 2, Zone 3, Zone 1); with five (5) zones of granitoid-hosted gold mineralization having been discovered to date within the approximately 1,200 m by 500 m to 800 m, Zone 2 gold-in-soil anomaly. The Zone 4 gold-in-soil anomaly has yet to be tested by trenching/drilling but its spatial association with a prominent, linear, high chargeability/low resistivity IP anomaly, and the eluvial/colluvial gravel-characterized, northern portion of the historical, alluvial gold resource Block B, renders the area prospective for shear hosted gold mineralization.

9.2.2 Ground Geophysics

The entire 5.5 km length of the Kibi Project anomalous gold-in-soil trend was covered by pole-dipole IP/Resistivity and Magnetometer surveys to help define the lithological and structural pattern of the mineralized trend and prioritized trench/drill targets. The geophysical surveys were implemented in August to September 2008 by Sagax Afrique of Ouagadougou, Burkina Faso.

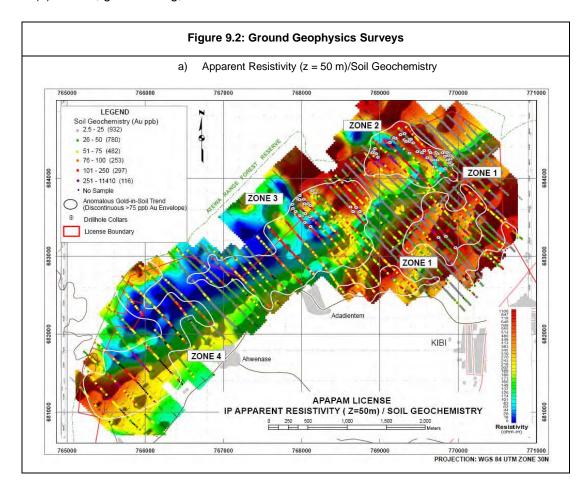
The approximately 64 line-kilometre IP/Resistivity (Time Domain) survey, covering the entire extent of the Kibi gold-in-soil trend at 200 m spacing, with some 100 m detailed sections centred on known gold showings (38 survey lines), was conducted using a Pole- Dipole Array with a dipole length of 50 m and dipole separations of n = 1 to 6. This survey design was selected to yield an approximate depth of investigation of about 175 to 200 m at n = 6. The ground magnetometer survey covered the entire Kibi Project soil geochemical grid totalling approximately 79 line-kilometres at 12.5 m station readings.

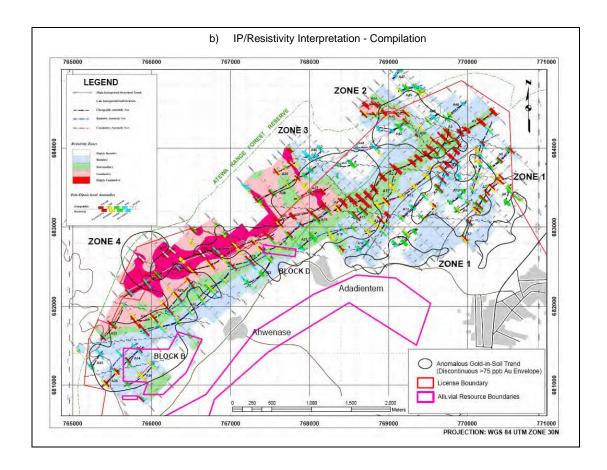
The IP/Resistivity survey identified 2 main resistive domains (Figure 9.2a) exhibiting a close spatial relationship with the main four (4) gold-in-soil anomalies of the Kibi Project (Zone 1 to 4); the resistive terrain is interpreted to reflect the widespread granitoids and/or carbonate-silica alteration associated with the known gold mineralization. A total of 36 chargeable anomaly axes have been identified (A1 to A36 on Figure 9.2b), including 24 interpreted as priority anomalies; with the majority of these anomalies yet to be field tested. The survey outlined a prominent NE-trending, high chargeability, typically with high conductivity (i.e., low resistivity), central corridor that possibly reflects a regional graphite-bearing shear zone. The continuity of the central anomalous chargeability corridor is intermittently dissected by inferred, NW-trending structures; with some exhibiting anomalous geophysical and soil geochemical signatures. The northeast extremity of the anomalous chargeability corridor is associated with a nonmagnetic domain (Mag Low), possibly reflecting alteration-related magnetite destruction of the host-rock.

Although the widespread granitoid-hosted gold mineralization in Zone 2 is typically characterized by 1-3% (locally up to 5%), disseminated sulphides (pyrrhotite, pyrite, arsenopyrite), the known mineralization occurrences failed to produce a chargeability response. The entire extent of Zone 2 is characterized by a resistive to strongly resistive signature; but individual granitoid bodies are not discernible on the resistivity map. A narrow shear zone exhibiting low grade gold values associated with quartz-carbonate veining and developed within graphitic phyllites in the eastern portion of Zone 2, is spatially associated with a NE-trending, moderate chargeability/moderate resistivity anomaly (i.e., A48). Trenching and drilling observations indicate that the broad, highly chargeable/highly conductive, NE-trending anomaly lying at the southeast extremity of Zone 2 reflects a graphitic metasedimentary rock sequence (i.e., A1).

The Trench TAD001 to TAD004 granitoid-hosted gold zone located within the northcentral portion of the Zone 3 gold-in-soil anomaly is spatially associated with an approximately 800 m long IP chargeability anomaly exhibiting a spatial relationship with a geophysically inferred, NE-trending, regional structural trend. Limited scout RC drilling along the north-eastern, moderate

chargeability/very high resistivity portion (200 m) of the IP anomaly outlined an approximately 135 m wide, NE-trending, granitoid hosted, structural corridor appearing to encompass at least five (5) distinct, gold-bearing, sheeted to stock worked vein zones.





Trench TKB003 excavated on the Zone 1 gold-in-soil anomaly exposed a system of foliation parallel quartz-carbonate-pyrite (+/- arsenopyrite) veins hosted by shear zones developed within a tightly folded metasedimentary rock sequence; spatially associated with a high chargeability/low resistivity IP anomaly lying along a geophysically inferred, NE-trending, regional structural trend. Scout drilling of this zone yielded intermittent, exploration significant, anomalous gold values over a 60 m core length, including individual intercepts of 1.43 g/t gold over 13.5 m, 1.04 g/t gold over 6 m and 1.02 g/t gold over 8 m. An excavation on a gold-in-soil anomaly spatially associated with the southern margin of the same chargeability anomaly, returned a channel sample intercept of 2.51 g/t gold over a 4 m trench length.

The south-western portion of the gold-in-soil trend is characterized by an approximately 3.5 km long, NE-trending, generally moderate to high chargeability/ moderate to high resistivity anomaly lying along or proximate to the contact between the southern resistive domain and the central conductive corridor; and exhibits a spatial relationship with a geophysically inferred, NE-trending, regional structural trend.

The south-western segment of the chargeable/resistive anomaly exhibits a spatial association with the Zone 4 gold-in-soil anomaly and with the northern portion of alluvial gold resource Block B; which according to historical alluvial exploration work consists of eluvial/colluvial gravels rather than alluvial gravels (sensu stricto). Eluvial/colluvial gravels consisting of residual material derived by in situ rock weathering or weathering plus gravitational/slump movements, and loose rock/soil material deposited by gravity at the base of a steep slope. The north-

eastern extremity of the chargeable/resistive response exhibits a spatial association with the south-western portion of the Zone 3 gold-in-soil anomaly. The central portion of the anomaly is not, for the most part, characterized by an anomalous gold-in-soil signature, which could reflect extensive alluvium/colluvium cover at the base of the Atewa Range, but prospecting efforts have identified an extensive gold anomalous float train spatially associated with this section of the IP/Resistivity anomaly.

9.2.3 Prospecting

A total of 109 grab samples have been collected by Xtra-Gold during prospecting and reconnaissance geology traverses primarily designed to follow-up on gold-in-soil anomalies. Sampling included 44 rock floats and 65 in situ samples consisting of saprolitic or saprock (weakly oxidized) bedrock material; with bedrock samples typically collected along road cuts and from historic workings. Eighteen (18) out of the 109 samples returned below detection limit gold values (<0.01 g/t); 63 samples yielded between 0.01 - 0.10 g/t gold; 20 samples yielded between 0.1-1.0 g/t gold; and eight (8) samples returned between 1.0 - 7.5 g/t gold. A fair portion of the 28 anomalous grab samples (> 0.1 g/t Au) were collected from areas that were subsequently tested by trenching and/or drilling.

Prospecting identified an extensive gold anomalous float train, exhibiting minimum 550 m x 325 m dimensions, along the south-western margin of the Kibi Project grid area. The float train is spatially associated with the northern flank, of the central portion, of an approximately 3.5 km long, NE-trending, moderate to high chargeability/moderate to high resistivity anomaly associated with a geophysically inferred, NE-trending, regional structural trend. A total of 14 rock samples were collected from the float train, with seven (7) samples yielding anomalous values ranging from 0.16 g/t to 3.49 g/t gold; but including a 7.5 g/t gold value. Mineralized floats are sub-angular to sub-rounded, average 0.15 to 0.75 m in diameter (2.5 m max.), and generally consist of strongly silicified metasedimentary rock (possibly siltstone) crosscut by sheeted to stock worked quartz stringers; with disseminated limonitic boxworks appearing to be after pyrite.

9.2.4 Trenching

Reconnaissance trenching designed to test the geochemical signature at depth of the approximately 5.5 km long gold-in-soil trend continued in 2008 with the implementation of a manual (i.e., hand dug) trenching program encompassing 18 trenches totalling approximately 1,217 linear-metres, including 4 trenches (302 m) on Zone 2 (TKB006-009) and 14 trenches (915 m) on Zone 3 (TAD008-021). In addition, 67 mechanical (i.e., excavator) trenches totalling approximately 2,223 linear metres were also excavated in conjunction with the 2008 and 2009 drilling programs, including 58 trenches (1,931 m) on Zone 2; 7 trenches (193 m) on Zone 3 and 2 trenches (99 m) on Zone 1 of the Kibi Project; with this trenching primarily designed to help map/trace the granitoid bodies hosting the gold mineralization.

The reconnaissance trenches designed to test the subsurface signature of the gold-in-soil anomalies were sampled in their entirety, with a total of 629 channel samples collected. Only a small percentage (12%) of the mechanical trenches were sampled based on the fact that this

trenching was predominantly designed to guide the drilling by mapping the contacts of the host granitoid bodies and, to a lesser degree, tracing the known mineralized vein zones. Only eight (8) out of the 67 granitoid mapping trenches were subjected to sampling, either in their entirety or partially, for a total of 132 channel samples (205 m). Xtra-Gold then adopted the practice of completely sampling all trenches regardless of trenching purpose or target.

Manual (pickaxe and shovel) and mechanical (excavator) trenches are typically excavated to widths of 1 m and 1.5 m, respectively, and an average depth of 3 m, with some sections of trenches reaching 4 to 5 m in depth. Trenching typically extends down to the saprolite horizon, or locally to saprock, but often the saprolite cannot be reached due to safety concerns. The entire length of the trench is subjected to systematic geological mapping and channel sampling; with wooden pegs stuck to the side of the trench at 2 m intervals. Prior to sampling the wall of the trench is cleaned of any loose material to avoid contamination.

Samples consist of continuous, horizontal channels excavated along the bottom sidewall of the trench (~ 0.10 m above floor) with emphasis on constant sample volume over the length of the sample interval. Saprolite/rock chips are collected on a clean plastic sheet laid on the trench floor and immediately placed into a labelled plastic sample bag containing a unique sample ticket stapled to the inside lip of the bag, and securely sealed by staples.

Figure 9.3 depicts channel sampling of a trench on the Apapam Concession. Samples are typically 2 m in length; with 1 m, to locally 0.5 m, samples being utilized in areas of geological interest and/or to delineate specific lithological/structural features. Any economically significant mineralized intersection yielded by 2 m sampling is re-sampled at 1 m intervals (this procedure implemented in 2010 sampling programs). The sample intervals (i.e., sample numbers) are marked on aluminium tags stapled to wooden pegs stuck to the sidewall of the trench. Samples are collected by a trained field assistant under the supervision of a company geologist. Road cut and drill pad face samples are collected using the same general methodology as the trench channel samples.

First pass exploration trenches are located by tying-in the trench start and end points to the DGPS surveyed grid stations, or by handheld GPS readings if no grid is present, and azimuth and slope information collected by compass/inclinometer. Typically, any trenching in hilly terrain (> 10° slope) is excavated utilizing step-like benches in order to maintain horizontal sampling intervals. Since January 2010, trenches yielding significant mineralization and/or forming part of a detailed trenching program have been surveyed utilizing combined DGPS and Total Station defined control points. The survey crew also systematically record azimuth and slope measurements.



Figure 9.3: Channel Sampling in Trench at Apapam Concession

Zone 2 Trenching

A total of 62 trenches (7 m to 136 m) totalling 2,233 linear-metres were excavated on the approximately 1,200 m by 500 m to 800 m, SE-trending, Zone 2 gold-in-soil anomaly during 2008-2009, including four (4) hand dug, reconnaissance trenches (302 m) and 58 mechanized, granitoid mapping/tracing trenches (1,931 m). For these reasons, only five (5) out of the 58 mechanized trenches were sampled.

Out of the four (4) reconnaissance trenches, only trench TKB006, located in the north-central portion of the Zone 2 gold-in-soil anomaly, yielded a significant mineralized intercept. Trench TKB006, targeting a 120 ppb to 385 ppb gold-in-soil anomaly, returned a mineralized intercept of 1.46 g/t gold over 36 m, including 2.20 g/t gold over 17 m, from an extensive, granitoid-hosted, quartz vein system. Four (4) out of the five (5) mechanized trenches subjected to sampling returned significant mineralized intercepts. Trench TKB010, designed to trace the mineralization intersected in scout drill hole KBD08008, at the north-western extremity of the Zone 2 gold-in-soil anomaly, returned a mineralized intercept of 1.29 g/t gold over 42 m, including 2.26 g/t gold over 13m, from an extensive system of NE to NW-trending quartz-carbonate veining developed along the margin of a tonalitic intrusive body. For comparison purposes, RC drill hole KBRC09068, designed to undercut trench TKB010, yielded a mineralized intercept of 76 m grading 1.62 g/t gold, including 20 m grading 3.36 g/t gold.

Trenches TKB014E and TKB014F, targeting a gold-in-soil anomaly spatially associated with a quartz float train, lying approximately 225 m west-northwest of the trench TKB006 gold occurrence, exposed an auriferous, granitoid-hosted, quartz-carbonate vein network. The two (2) trenches positioned end to end on the same soil geochemical anomaly line, both returned significant channel sample intercepts separated by an approximately 20.5 m distance, including

8.49 g/t gold over a 5 m trench-length, including 2 m grading 14.85 g/t gold, in trench TKB014E and 6.86 g/t gold over an 8 m trench-length, including 1 m grading 22.4 g/t gold, in trench TKB014F.

Zone 3 Trenching

A total of 14 manually excavated reconnaissance trenches (15 m to 154 m) totalling 915 linear-metres (TAD008 to TAD021) were completed on the approximately 2,500 m by 250 m to 1,200 m, NE-trending, Zone 3 gold-in-soil anomaly in 2008. Mineralized intercepts reported below are trench-lengths; true width of mineralization is unknown at this time (see Table 2 for significant Zone 3 trench results).

Five (5) out of the 14 trenches, designed to test the geochemical signature at depth of gold-in-soil anomalies, exposed altered granitoid bodies exhibiting variable amounts of quartz-carbonate veining. The Zone 3 reconnaissance trenching yielded two (2) significant granitoid-hosted mineralization intercepts (TAD019, TAD016); with the remaining three (3) altered/veined granitoid occurrences yielded lower grade but exploration significant, anomalous gold values.

Extensive, strongly indurated, lateritic clays and gravels prevented the proper testing of some gold-in-soil anomalies due to the saprolite horizon was not reachable at many localities in the hand dug trenches. Mechanized trenching and/or RAB drilling is recommended to further test the geochemical signature of the Zone 3 gold-in-soil anomalies at depth within the saprolite horizon.

9.2.5 Structural Study

SRK Consulting (Canada) Inc (SRK) to conduct a structural study to understand the structural controls on gold mineralization at the Kibi Gold Project. SRK reviewed 14 diamond drill holes (Zone 1 and 2) as well as available trench exposures (Zone 2 and 3).

The following observations were made:

- Overall N- to NE- trending shear zones occurs throughout the Apapam concession that are preferentially developed along the margins of quartz diorite intrusions;
- Gold mainly occurs in gently to moderately NW dipping quartz-albite-carbonate vein stockworks bounded by NE trending, steeply dipping faults within quartz diorite intrusions;
- Low-grade gold mineralization is associated with quartz-carbonate veins in narrow (graphitic) shear zones that occur in tightly folded metasedimentary sequences;
- Hydrothermal alteration typically consists of quartz, carbonate, chlorite and sericite where auriferous quartz veins occur within quartz diorite intrusions;
- Vein geometries and rare kinematic indicators suggest a reverse sense of shear associated with vein and shear zone development;
- The 3D geometry of quartz diorite intrusions and the auriferous portions were poorly constrained at that date.

9.2.6 Petrographic Study

A petrographic study was also implemented to characterize the lithological units and ore mineralogy and alteration of the Kibi Gold Project. A total of 36 thin sections and nine (9) polished sections were studied by Professor K. Dzigbodi-Adjimah of the University of Mines and Technology, Tarkwa, Ghana.

The observations made include:

- Identification of the altered granitoid as quartz dioritic to granodioritic in composition.
- Identifying three generations of pyrite with the gold being associated with the last sulphide forming event.
- Observing that the gold often occurs as inclusions within pyrite, pyrrhotite and arsenopyrite which suggests that the sulphides are not refractory.

9.3 2010 -2012 Exploration Program

9.3.1 Geophysics

VTEM Survey

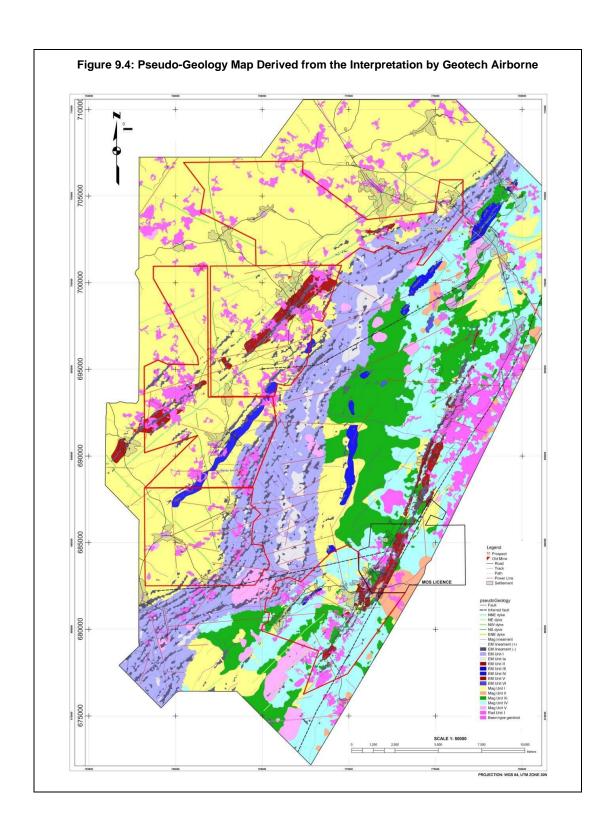
A VTEM survey was flown over the Kibi Gold Project by Geotech Airborne from December 2010 to February 2011. The survey measured ground elevation, radiometrics, magnetic field and electromagnetism (resistivity). Interpretation of the data resulted in an interpretive pseudogeology map of the area (Figure 9.4). The different geophysical units can be correlated with various geological units.

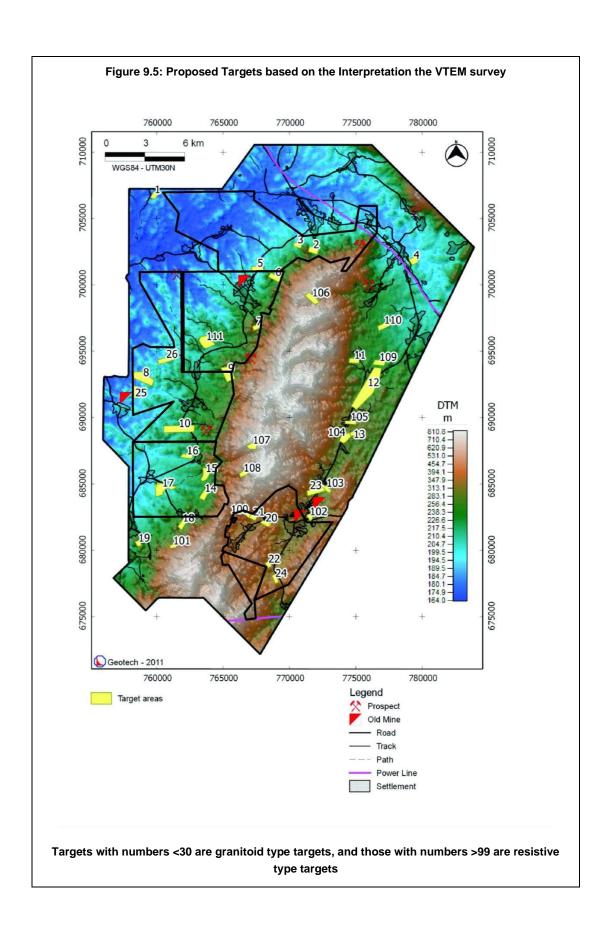
Target areas were defined in the report for further ground exploration (Figure 9.5). Two kinds of targets were defined:

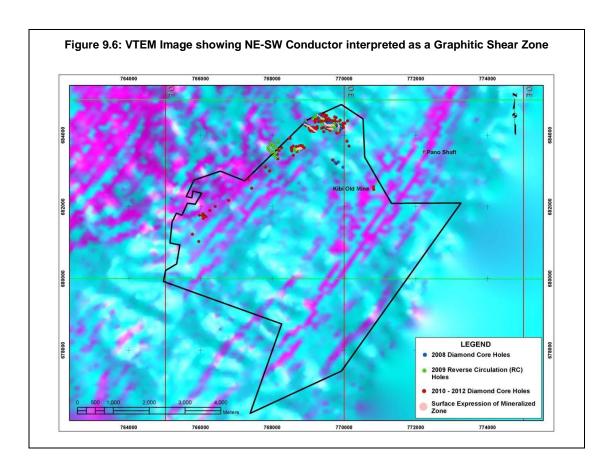
- Resistive type: following the silicification model for gold mineralisation, areas of low conductance/susceptibility occur within the interpreted conductive graphitic shear zones and the interpreted conductive graphitic sediment units.
- Granitoids-type: shear/ fracture zones in basin-type granitoids are known to exhibit subeconomic to economic gold mineralisation in the survey area. As such targets have been defined where interpreted fractures intersect the interpreted basin-type granitoids.

In the Apapam Concession 1 resistive-type (100) and 4 granitoid-type (20, 21, 22, and 24) targets were proposed (Figure 9.5). According to Geotech Airborne, targets 100 and 21 are priority 1 targets, target 20 is a priority 2 target, and targets 22 and 24 are priority 3 targets.

Following on from the work by Geotech Airborne, Xtra-Gold geologists have performed more detailed processing to highlight the various faults around the project area in the VTEM data (Figure 9.6).







9.3.2 Geochemistry

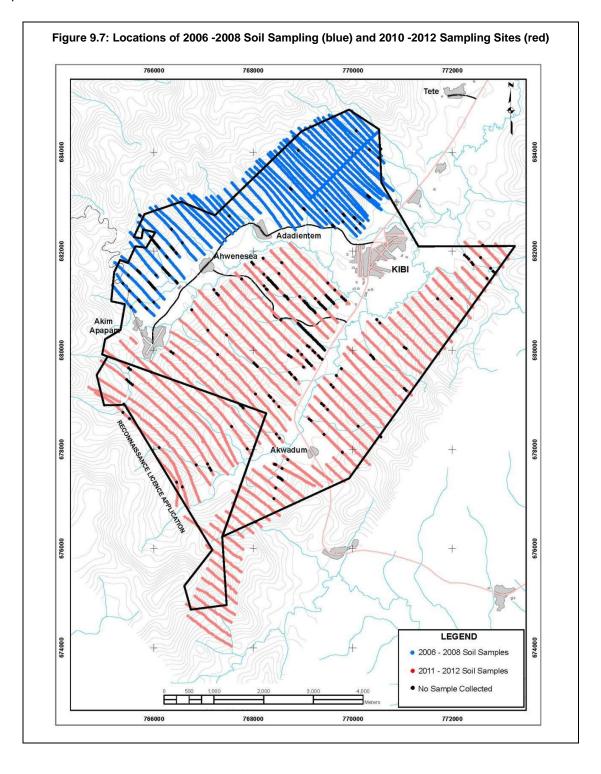
Soil Sampling

From June 2011 to April 2012, 3,833 soil samples were collected by Xtra-Gold to complete the soils sampling program over the Apapam Concession (Figure 9.7). Previously only the north-western part of the concession had been covered by soil sampling. The samples were collected every 25 m along lines 200 m apart. Of the 3,833 samples collected, 2,747 were submitted to the laboratory for analysis.

As per program design every second sample (50m stations) was initially submitted for gold analysis (1,859 samples); with the held – back samples to be subsequently analysed where required to delineate/bracket anomalous gold-in-soil anomalies. Based on gold results an additional 888 infill (25 m station) samples were selected for analysis to further define the anomalous gold-in-soil trends.

Soil samples are collected from 20 cm to 30 cm diameter, hand-dug pits at a nominal depth of 75 cm using the local digging tool called soso. Approximately 2.5 kg of material is collected into labelled plastic bags with unique sample tickets stapled to inside lip of the bag, and securely sealed by staples. To avoid any contamination only dry samples are collected. Field logging includes sample depth, landscape, slope direction, land use, soil type/characteristics, residual/erosional/depositional environment, and regolith type.

Float and outcrop grab samples (115 samples) were collected to assist with delineation of potential mineralisation.



Rock Chip Sampling

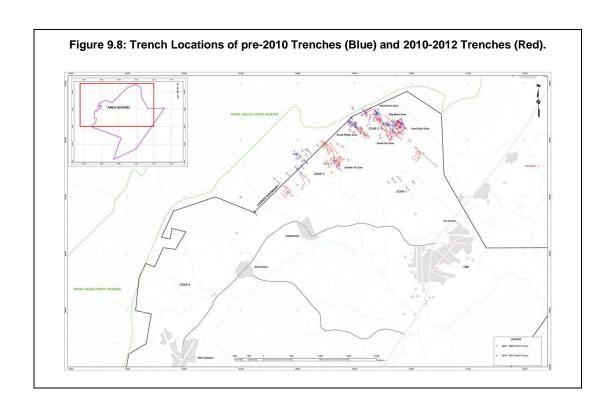
One hundred and fifteen (115) rock grab samples have also been collected from outcrops and from floats and analysed for gold. The rock sampling was conducted in conjunction with the 2011 – 2012 soil geochemical survey and follow up prospecting of gold-in-soil anomalies. Out of the 115 grab samples collected six (6) returned gold grades between 0.19 g/t and 12.35 g/t; and the remaining 109 yielded below or slightly above detection limit gold values. Sampling of felsic intrusive outcroppings returned highs of 1.5 g/t and 12.35 g/t gold, and a quartz float yielded 11.0 g/t gold.

9.3.3 Trenching

Trenching was undertaken from July 2010 – August 2012 to supplement the drilling of the targets already recognised. Trenches were excavated on the majority of the targets that form part of the mineral resource estimate (Table 9.1).

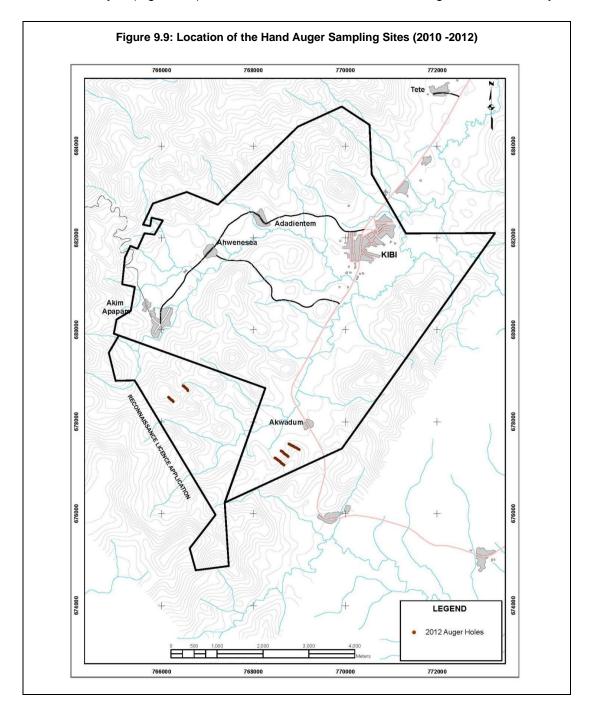
Table 9.1: Summary of Trenching (2010 - 2012)								
July 2010 to August 2012	No of Trenches	Metres	No of Channel Samples	Sample Metres	Vertical Sample Sections	Metres	No of Vertical Channel Samples	Sample Metres
Big Bend	24	863	673	783	28	68	68	68
East Dyke	12	436	336	398	38	85	84	85
Mushroom	5	94	41	41	0	0	0	0
South Ridge	26	1,106	847	977	33	81	81	81
Double 19	26	1,173	786	1,036	78	162	154	162
Other Targets	107	6,021	4,346	4,971	202	514	509	514
Total	200	9,693	7,029	8,206	379	910	896	910

Trenching was undertaken in an effort to help define the extents and geological context of gold mineralisation (Figure 9.8). Due to the high relief in the area, some of the trenches are actually cleaned road cuttings on the sides of the hills. Of the 200 trenches that were dug, 198 of them were dug using an excavator, while the remaining two were dug by hand. Depth of the trenches varied widely from a metre to 5 m depending on relief and steepness of slope on hill side. Widths of the trenches average 1.5 m. A total of 4,346 horizontal channel samples were taken and because of the prevalence of shallowly dipping veins, 509 vertical channel samples were also taken where appropriate. Where the 2 m samples yield anomalous results, 1 m samples are re-taken from the interval. The vertical sample sections were usually spaced 2.5 m apart. The total number of samples sent to the laboratory for analysis was 7,925. Typical QA/QC protocols were observed.



9.3.4 Hand Augering

At Akwadum South 44 auger drill holes were sunk for a total of 237 m with 147 samples being collect and assayed (Figure 9.9). These drill holes were drilled to test a gold-in-soil anomaly.



9.3.5 Structural Analysis of Zone 2

SRK Consulting (Canada) investigated the structural geology of the Kibi Gold Trend project – Zone 2 in November 2011. They examined a number of trenches and drill holes from the Big Bend zone and South zone.

SRK concluded that:

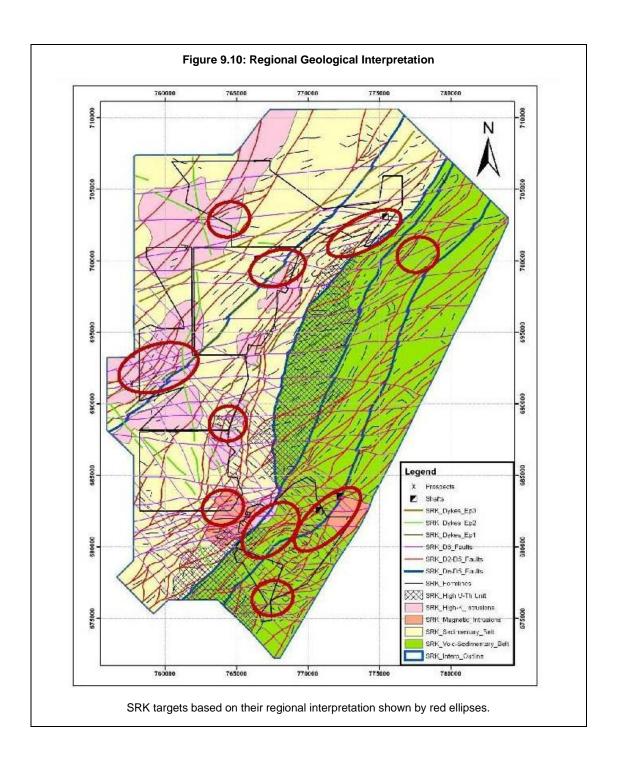
- The distribution of gold mineralization in the Big Bend Zone is controlled by two NNEtrending shear zones that bound the auriferous zone in a quartz diorite;
- Auriferous quartz veins in the Big Bend Zone comprise:
 - Shear and extensional veins related to the development of NNE-trending shear zones; and
 - Stockwork veins in a particular portion of the quartz diorite;
- Auriferous quartz veins in and around the shear zone have two dominant orientations (measured from drill core):
 - Steeply-dipping to the ESE (average 024°/78°); and
 - o Gently-dipping to the SE (average 035°/12°).
- Auriferous quartz veins in non-to weakly-foliated diorite in the Big Bend Zone form vein stockworks. Auriferous quartz veins within the quartz diorite (outside of shear zones) have two dominant orientations:
 - o Gently-dipping to the NNW (average 240°/15°); and
 - Moderately-dipping to the NW (average 215°/45°);
- Vein geometry, rare kinematic indicators and steeply plunging mineral lineations imply that deformation associated with gold mineralization in the Big Bend Zone resulted from a protracted episode of dominantly reverse SE over NW with minor sinistral movement; and
- In the South Zone, steeply-dipping auriferous laminated and breccia veins occur, in addition to the dominant gently dipping (<15°) extensional veins.
- The controls on gold mineralization at the South Zone and other zones are not well understood and require further oriented core drilling followed by structural geology investigations.

9.3.6 Regional interpretation

SRK Consulting (Canada) analysed the regional structural geology and interpreted the aeromagnetic data for the Kibi Gold Project in December 2011. They interpreted regional aeromagnetics and the VTEM data geological interpretation over the area (Figure 9.10).

Selected areas of structural complexity that are of interest for exploration in the Kibi area were highlighted, with three of the targets overlapping the Apapam Concession. The targets are based on the following criteria:

- Bends along fault corridors that acted as dilational jogs during sinistral strike-slip deformation;
- Areas of intersection between anastomosing NE-SW trending faults, or where these faults cross-cut intrusions:
- Areas of intersection between major NE-SW trending fault corridors and E-W trending faults; and
- The presence of intrusions at or near any of the above faults.



9.4 2012 - 2021 Exploration Programme

The exploration programme during the 2012 – 2021 period consisted of ongoing compilation of geological data, soil geochemical sampling, and scout trenching to identify and/or further advance grassroots targets; and drilling to extend mineralization and prepare an updated Mineral Resource estimate.

Exploration activities were primarily geared towards the continued advancement of early-stage gold shoots/showings within the Zone 2 - Zone 3 maiden Mineral Resource footprint area.

Activities from 2013 – 2016 focussed on preliminary work on the Cobra Creek Target (Zone 5), including trenching and outcrop stripping / channel sampling, and a scout drilling program.

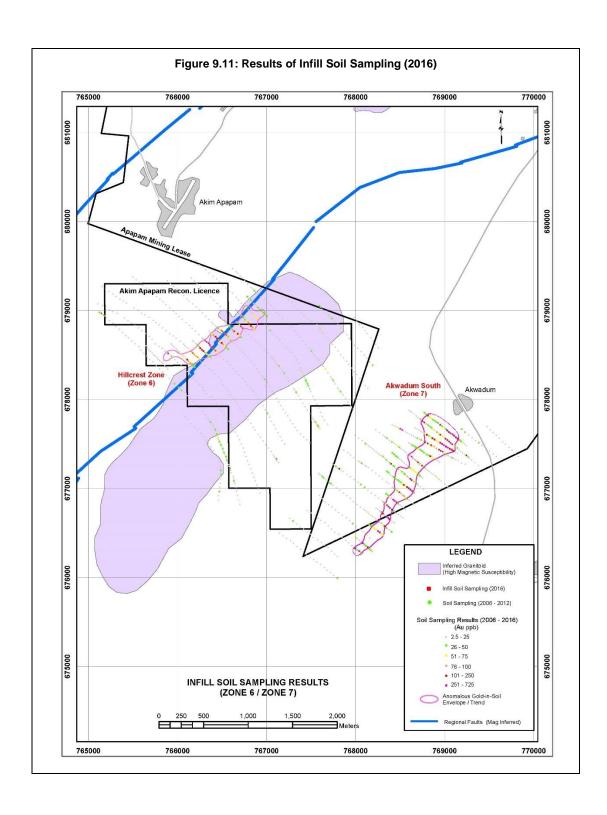
Exploration activities from 2018 – 2021 targeted resource expansion opportunities within the Zone 2 – Zone 3 maiden Mineral Resource footprint area. The drilling / trenching program was designed to follow up on early-stage gold shoots/showings discovered by previous drilling/trenching efforts (2008 – 2012), to test down-plunge extensions and/or fold limbs of existing resource bodies and to test prospective litho-structural gold settings identified by recently completed 3D geological modelling.

9.4.1 Soil Sampling

Further infill soil sampling was undertaken on the Cobra Creek (Zone 5) and on Akwadum South (Zone 7) targets. The sampling was undertaken to provide more detailed 100 m line-spacing coverage of previously identified gold-in-soil anomalies (i.e., infilling of 2012 sampling at 200 m line-spacing). Soil sampling methodology consist of the collection of approximately 2.5 kg of regolith material from hand-dug pits at depths of 50 to 60 cm with normal diameters not exceeding 30 cm using the soso digging tool.

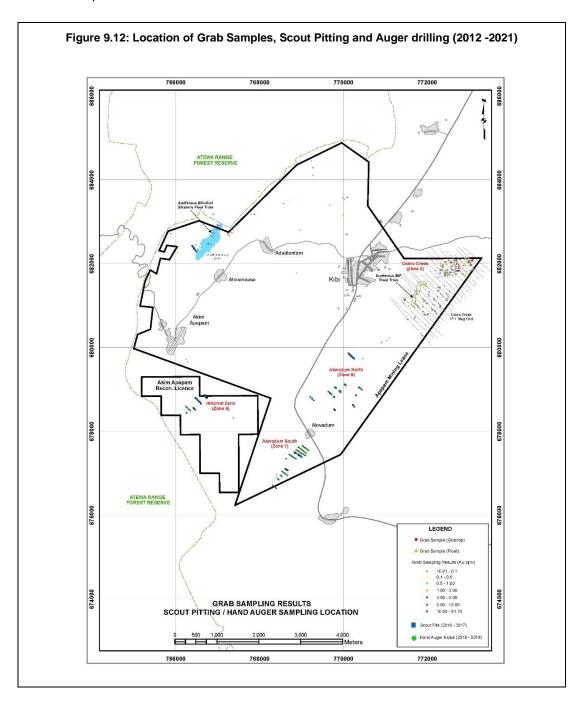
A total of 10.71 line-kilometres of soil sampling was completed, including 7.03 line-kilometres at Cobra Creek and 3.68 line-kilometres at Akwadum South. With a total of 458 samples collected at 25 m stations along the NW-SE grid lines (Cobra Creek – 310 and Akwadum – 148). The infill soil sampling permitted the delineation of an approximately 1,700 m long by 100 m – 200 m wide anomalous gold-in-soil trend at Akwadum South (Zone 7) (Figure 9.11). The southwest extremity (500 m) of the anomaly extends on Third Party ground (i.e., 1,200 m on Apapam mining lease) and the anomaly remains open at the northeast extremity with the soil coverage being interrupted by the hamlet of Akwadum.

The Akwadum South gold-in-soil trend is defined by an envelope of discontinuous/patchy, typically greater than 50 ppb, gold-in-soil values; with the anomalous threshold arbitrarily set at 50 ppb gold based on past exploration experience by Xtra-Gold in the Kibi Greenstone Belt. Out of the 152 samples within the anomalous trend, including both 2012 and 2016 sampling, 114 yielded gold values (75%) greater than the 50 ppb anomalous threshold, including: 58 samples from 51 ppb to 100 ppb gold; 49 samples from 101 ppb to 300 ppb gold; and seven (7) samples above 300 ppb gold (725 ppb maximum).



9.4.2 Grab Sampling

Sampling of outcrop and float material was performed at Cobra Creek (Zone 5) and other locations on the Apapam concession. A total of 910 grab samples were collected (Cobra Creek – 794 and 116 – other locations) as part of prospecting efforts geared towards the testing of quartz-bearing structures and the ground proofing of geophysical anomalies (Figure 9.12). Note that grab samples are selective in nature and may not be necessarily representative of the mineralization present on the concession.



At the Cobra Creek (Zone 5) prospect, systematic outcrop / rock float grab sampling helped delineate an approximately 550 m wide, NE-trending, multi-structure braided shear zone system traced over an approximately 850 m strike length (Figure 9.14). With the quartz feldspar porphyry (QFP) hosted mineralized corridor encompassing at least 9 auriferous shear zones ranging from approximately 1 m to 25 m in apparent width. Of the 794 grab samples collected on Zone 5: 167 samples yielded less than 0.01 g/t gold; 330 samples yielded gold values from 0.01 g/t to 0.1 g/t; 158 samples between 0.1 g/t and 1.0 g/t gold; 76 samples between 1.0 g/t and 5.0 g/t gold, 55 samples from 5.0 g/t to 20 g/t gold, and 8 samples returned values above 20 g/t gold, including a maximum value of 61.7 g/t gold.

Further Zone 5 prospecting yielded an approximately 650 m long, auriferous banded iron formation (BIF), rock float train spatially associated with a series of coincident high chargeability (IP) / weak – moderate resistivity anomalies and patchy to intermittent anomalous gold-in-soil values in the 50 ppb to 225 ppb range (Figure 9.17). The mineralized hematite BIF material is characterized by quartz stockworks, strong patchy to pervasive silica alteration, and pyritization. Of the 57 BIF samples collected: 6 samples yielded less than 0.01 g/t gold; 22 samples returned gold values from 0.01 g/t to 0.1 g/t; 28 samples between 0.1 g/t and 1.0 g/t gold; and 1 sample returned a maximum value of 2.18 g/t gold. In situ source of auriferous BIF material yet to be established by trenching / drilling.

Prospecting efforts in Zone 4 further defined an auriferous silicified / pyritized siltstone rock float field originally discovered in 2006. The mineralized siltstone floats are spatially associated with a NE-trending VTEM high resistivity anomaly situated along the same F1 isoclinal fold hinge as the JK West and JK East prospects (Figure 9.12, Figure 7.7, Figure 7.9). Of the 45 silicified siltstone samples collected to date: 3 samples yielded less than 0.01 g/t gold; 20 samples returned gold values from 0.01 g/t to 0.1 g/t; 13 samples between 0.1 g/t and 1.0 g/t gold; and 9 samples returned values above 1 g/t gold, including a maximum value of 11.3 g/t gold. In situ source of auriferous siltstone material yet to be established by trenching / drilling.

9.4.3 Trenching

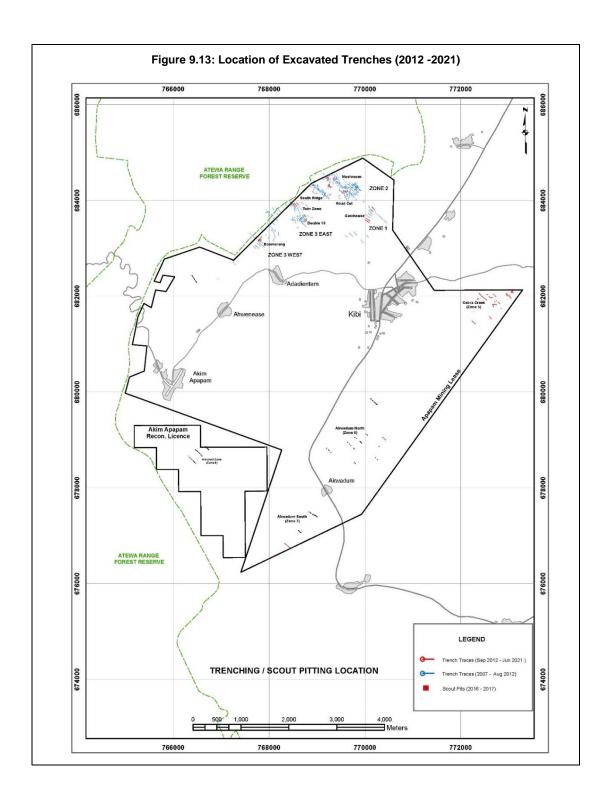
Trenching was undertaken from October 2018 – June 2021 on Zone 1 – Zone 2 – Zone 3 to supplement the drilling of the targets already recognised. Trenches were excavated on most of the targets that form part of the present mineral resource estimate (Table 9.2 and Figure 9.13). Trenching was also conducted intermittently from September 2012 - March 2016 to delineate the Cobra Creek (Zone 5) auriferous shear system.

Table 9.2: Summary of Trenches Excavated					
Oct 2018 - Jun 2021	No of Trenches	Trench Metres	Sampled Metres		
South Ridge	8	276			
Double 19	5	126			
Road Cut	8	201.1			
Mushroom	2	57.4			
Gatehouse	5	300.6			
Boomerang West	13	305.2			
JK East	10	260.8			
Other Targets	5	142.8			
Total Trenching	56	1669.9 m	1577.7 m		
Cobra Creek (Zone 5)	38	1194.3	1170.8 m		
Sept 2012 - Mar 2016	No of Trenches	Trench Metres	Sampled Metres		
Akwadum South (Zone 7) 2017	1	154.5	154.5 m		
Apapam Total Trenching	95	3018.7 m	2903.0 m		

Due to the high relief in the area, road cuts exposing saprolitic bedrock on hillsides are also categorize as trenches, and subject to the same surveying / sampling procedures described below for trenching. For differentiation purposes, the road cuts have the letters RS as part of the labelling prefix (i.e., Road Sampling).

The trenches were mechanically excavated with a width of 1.5 m and an average depth of 3 m, with some sections of trenches reaching 4 to 5 m in depth. Trenching typically extends down to the saprolite horizon, or locally to saprock, but often the saprolite cannot be reached due to safety concerns. The entire length of the trench is subjected to systematic geological mapping and channel sampling; with wooden pegs stuck to the side of the trench at 2 m intervals. Prior to sampling, the wall of the trench is cleaned of any loose material to avoid contamination.

Samples consist of continuous channels excavated along the bottom sidewall of the trench (~ 0.20 m above the floor) with emphasis on constant sample volume over the length of the sample interval. Samples are typically 2 m in length; with 1 m, to locally 0.5 m, samples being utilized in areas of geological interest and/or to delineate specific lithological/structural features. In addition to horizontal channel sampling, trenches where shallow-dipping quartz veining and/or shearing was observed were also subjected to vertical channel sampling at typically 2.5 m section spacing.



Saprolite/rock chips are collected on a clean plastic sheet laid on the trench floor and immediately placed into a labelled plastic sample bag containing a unique sample ticket stapled to the inside lip of the bag, and securely sealed by staples. The sample intervals (i.e., sample numbers) are marked on aluminium tags stapled to wooden pegs stuck to the sidewall of the trench. Samples are collected by a trained field assistant under the supervision of a company geologist.

Trenches are surveyed as three-dimensional features to permit 3D-plotting, with the trench data collected in standard drill database tables (i.e., collar, survey, geology, assays, structure). Along the Zone 1 – Zone 4 corridor and at Cobra Creek (Zone 5), trench collar coordinates (i.e., zero mark of channel sample string) are established by Total Station survey using DGPS-established control pillars, by the company's in-house surveyor. For trenching on early-stage projects with no established control pillars, the trench collars are surveyed-in utilizing handheld GPS established reference points (i.e., backsight & foresight).

The trace of the channel sample string is surveyed by Total Station from the collar to the end point, with azimuth and slope measurements collected at inflection points. The sample intervals are established to match the inflection points along the trace of the sample line. With the sample intervals (i.e., from and to measurements) representing slope measurements along the channel sampling line and not horizontally corrected distances.

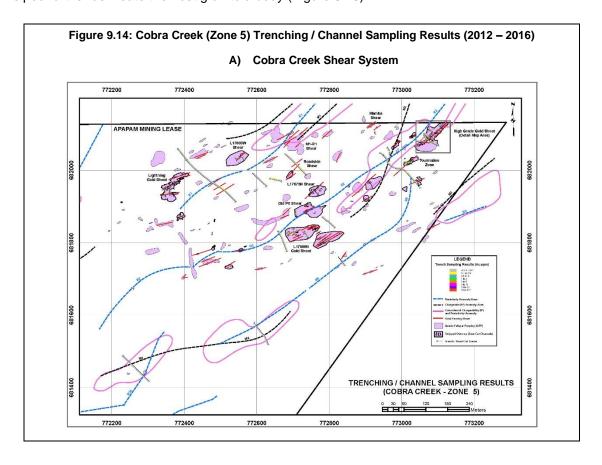
Ninety-five (95) trenches totalling 3,018.7 m were completed on the Kibi Gold Project during the 2012 – 2021 exploration program. Significant trench results are summarized in Table 9.3 and Figure 9.13 depicts the location of the trenches on the Apapam concession.

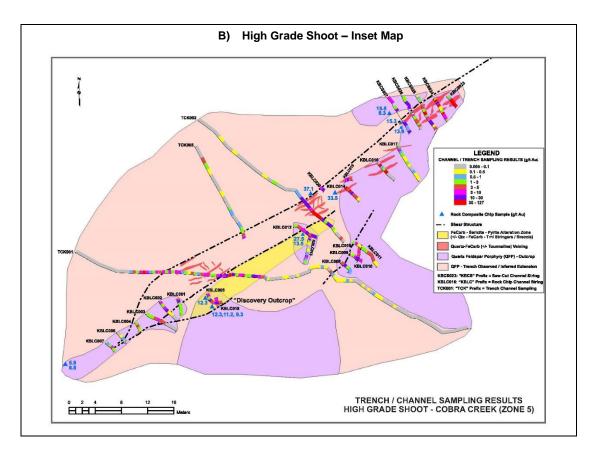
The reported trench results meet the following criteria: 4 m minimum length and minimum 0.3 g/t average grade over the interval or a minimum metal factor (grade x length) of 10 m-g/t if interval falls below the minimum 4 m criteria. In addition, mineralized intercepts are constrained with a 0.25 g/t gold minimum cut-off grade at top and bottom of intercept, with no upper cut-off applied, and maximum of five (5) consecutive samples of internal dilution (less than 0.25 g/t gold). Reported trench results correspond to trench-lengths in metres. The orientation / geometry of the mineralization is not fully understood in the saprolitic trench exposures resulting in unknown true thickness.

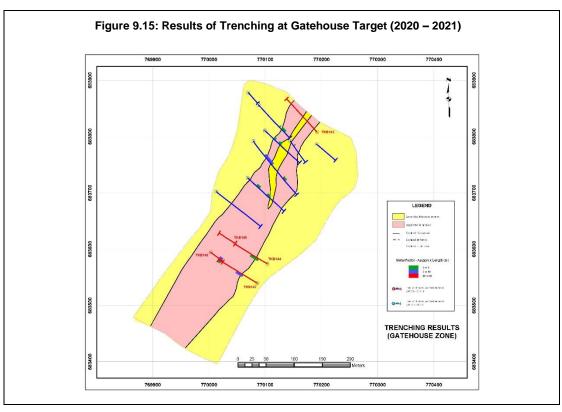
Table 9.3: Significant Trench Results (2012 – 2021)					
Trench ID	From (m)	To (m)	Trench-Length (m)	Au (g/t	
		South Ridge			
KBRS045	25.5	33.0	7.5	1.47	
KBRS047	7.3	12.0	4.7	4.32	
including	9	10	1	15.02	
		Double 19			
TAD044	0	7.2	7.2	1.85	
including	2	3.2	1.2	4.18	
TAD045	2.5	22.5	20.0	4.83	
including	14.0	19.5	5.5	9.60	
TAD046	1.0	5.5	4.5	0.87	
		Road Cut	<u> </u>		
TKB131	48.0	49.5	1.5	15.85	
TKB141	20.0	26.3	6.3	2.15	
		Mushroom			
TKB143	9.0	30.4	21.4	0.90	
including	17.4	24.4	7.0	1.86	
<u> </u>		Gatehouse			
TKB144	22	35	13.0	0.33	
TKB147	31.5	43.5	12.0	0.74	
including	37.5	39.5	2.0	2.30	
TKB148	19.0	24.4	5.4	1.18	
110110	10.0	Boomerang West	0.1	1.10	
TAH015	1.5	18.0	16.5	1.64	
including	4.5	13.5	9.0	2.35	
TAH016	25.5	36.0	10.5	1.93	
TAH023	18.0	31.0	13.0	0.5	
JK East	10.0	31.0	10.0	0.0	
TAD050	16.0	34.0	18.0	0.52	
ADRS018	9.0	25.0	16.0	0.52	
including	17.0	18.0	1.0	2.33	
including	17.0	Cobra Creek (Zone 5	1	2.33	
TCK004	0.5	,	,	7.06	
TCK001	9.5	30.0	20.5	7.26	
including	22.5	29.0	6.5	12.26	
TCK002	20.3	27.0	6.7	32.32	
including	23	25	2.0	82.22	
TCK003	4.0	12.0	8.0	1.45	
including	4.0	5.0	1.0	4.65	
TCK005	2.0	18.0	16.0	1.17	
including	17.0	18.0	1.0	6.15	
TCK006	44.9	53.0	8.1	0.38	
TCK008	61.0	73.0	12.0	0.31	
TCK013	3.0	5.0	2.0	14.98	
TCK017	0.0	9.1	9.1	0.42	
CCRS001	8.0	20.5	12.5	1.28	
including	14.7	15.2	0.5	6.98	
CCRS002	4.0	8.0	4.0	2.29	
CCRS003	3.5	9.5	6.0	5.96	
including	7.0	9.5	2.5	13.00	
CCRS005-A	0.0	2.4	2.4	6.25	
including	1.5	2.4	0.9	10.10	
CCRS009	0.0	4.0	4.0	1.67	
		Akwadum South (Zone	7)		
TAKS001A	18.0	43.25	25.25	0.35	
including	26.7	33.15	6.45	0.77	
TAKS001A	92.3	93.75	1.45	16.37	

At the Cobra Creek (Zone 5) prospect, systematic trenching helped delineate an approximately 550 m wide, NE-trending, multi-structure braided shear zone system traced over an approximately 850 m strike length. With the QFP hosted mineralized corridor encompassing at least nine auriferous shear zones ranging from approximately 1 m to 25 m in apparent width (Figure 9.14). Trenching was also utilized to test IP/Resistivity anomalies and to identify prospective areas for outcrop stripping and channel sampling.

At the Gatehouse target, trenching extended the gold mineralization approximately 150 m to the southwest, as well as confirmed the two, parallel zone geometry of the mineralized system, and helped further delineate the host granitoid body (Figure 9.15).







9.4.4 Outcrop Stripping and Channel Sampling

The QFP body hosting the gold mineralization at the Cobra Creek (Zone 5) prospect is characterized by an abnormally strong resistance to weathering, with the QFP forming unoxidized (fresh) rock outcrops in otherwise deeply laterized terrane.

A total of approximately $15,000 \text{ m}^2$ of bedrock exposures were mechanically stripped and power washed at the Cobra Creek project from 2012 - 2016 to permit systematic mapping and channel sampling of the auriferous shear system (Figure 9.13). A total of 24 stripped bedrock exposures, ranging from approximately 20 m^2 to $2,800 \text{ m}^2$, were excavated across an approximately $850 \text{ m} \times 375 \text{ m}$ extent of the host QFP body.

A total of 1,312.26 m of saw-cut channel sampling and 71 m of chip-channel sampling was completed on the stripped bedrock exposures (Figure 9.13). Individual rock channel samples range from 0.3 m to 2 m in length (0.7 m average length), with sample arrays ranging from single channel samples to continuous channel sample strings from 1.4 m to 37.6 m in length.

The saw-cut channel sampling involved the cutting of two continuous, parallel cuts approximately 3 cm - 5 cm apart to a typical depth of 5 cm - 7 cm utilizing a mechanical diamond blade saw. With the sample split between the two saw cuts with a chisel. The chip-channel samples involved the collection of contiguous rock chips along an approximate straight line forming an approximately 5 cm wide $x \cdot 0.5 \text{ cm} - 2 \text{ cm}$ deep channel; with sample depth depending on oxidation degree of rock.

The extremities of individual samples are marked with short cross-cuts, and the sample measurements (i.e., from and to) and channel sample string identification numbers, hand-painted on the rock surface with durable exterior paint. For differentiation purposes, the saw-cut channel string identification numbers have the KBSC prefix (i.e., Kibi Project – Saw-Cut Channel) and the chip-channel string identification numbers have the KBLC prefix (i.e., Kibi Project – Linear Chip-Channel).

Channel sample strings are surveyed as three-dimensional features to permit 3D-plotting, with the sampling data collected in standard drill database tables (i.e., collar, survey, geology, assays, structure). Channel sample string collar coordinates (i.e., zero mark of channel string) are established by Total Station survey using DGPS-established control pillars, by the company's in-house surveyor.

The trace of the channel sample string is surveyed by Total Station from the collar to the end point, with azimuth and slope measurements collected at inflection points. The sample intervals are established to match the inflection points along the trace of the sample string. With the sample intervals (i.e., from and to measurements) representing slope measurements along the channel sampling string and not horizontally corrected distances.

Of the total 2,046 channel samples collected: 105 samples yielded less than 0.01 g/t gold; 777 samples returned gold values from 0.01 g/t to 0.1 g/t; 614 samples between 0.1 g/t and 1.0 g/t gold; 357 samples between 1.0 g/t and 5.0 g/t gold; 98 samples between 5.0 g/t and 10.0 g/t

gold; 68 samples between 10.0 g/t and 25.0 g/t gold; 22 samples between 25.0 g/t and 50.0 g/t gold; and 5 samples above 5.0 g/t gold (121.0 g/t maximum).

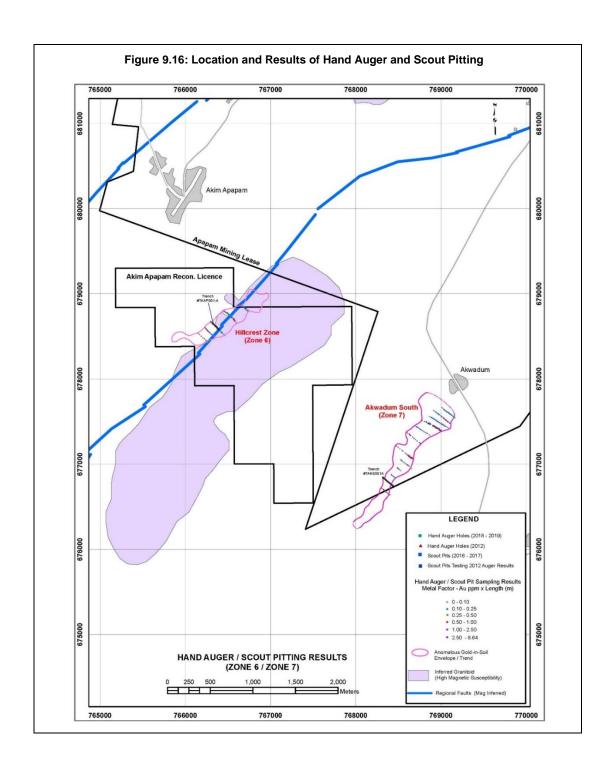
Significant channel sampling results are summarized in Table 9.4. The reported channel samples results meet the following criteria: 1 m minimum length and minimum metal factor (grade x length) of 15 m-g/t. Reported channel sample results correspond to channel-lengths in metres. Due to irregular bedrock surface, the channel sample results represent sample intersection lengths irrespective of mineralization topography and may not represent true width of mineralization.

	From (m)	To (m)	Channel-Length (m)	Au (g/t
	,	High Grade Shoot	<u> </u>	,,
KBCS023	0.0	2.4	2.4	33.21
KBCS024	0.2	4.5	4.3	15.13
KBCS025	0.0	4.9	4.9	8.68
KBCS026	0.0	7.0	7.0	6.31
KBCS027	0.0	9.0	9.0	6.74
KBCS023-2	1.72	9.65	7.93	4.10
KBCS023-11	0.0	3.95	3.95	3.91
KBCS023-12	0.0	4.3	4.3	12.00
KBCS023-45	2.9	6.8	3.9	12.61
KBCS023-46	2.0	7.0	5.0	23.62
including	6.1	7.0	0.9	67.90
KBCS023-47	4.12	8.7	4.58	20.48
including	5.9	7.8	1.9	35.47
KBCS023-48	3.0	7.55	4.55	15.11
KBLC010	0.0	4.0	4.0	8.9
KBLC012	0.0	4.0	4.0	7.69
KBLC014	0.0	3.0	3.0	16.14
KBLC016	0.7	3.0	2.3	40.6
KBLC018	0.0	2.2	2.2	8.72
KBLC020	0.0	1.0	1.0	18.05
		High Grade Shoot - Wes	st	
KBCS092-3	0.35	3.72	3.37	10.26
KBCS092-4	0.0	3.15	3.15	13.17
including	0.0	0.95	0.95	31.50
KBCS092-5	0.0	1.1	1.1	19.35
		Tourmaline Zone		
KBCS019	0.0	7.4	7.4	2.25
KBCS028B	0.0	7.6	7.6	3.13
		L17600N Shoot		
KBCS080-2	0.0	1.5	1.5	10.47
KBCS080-5	0.0	1.6	1.6	11.14
KBCS080-9	0.6	4.5	3.9	5.90
KBCS080-18	0.0	1.1	1.1	14.77
KBCS080-28	1.4	6.5	5.1	5.88
KBCS080-33	5.2	12.0	6.8	3.12
KBCS080-35	0.5	3.5	3.0	6.39
		Old Pit		
KBCS077-11	0.0	3.8	3.8	6.84
KBCS077-26	0.9	5.2	4.3	13.89
		L17600W Shear		
KBCS078-5	2.2	5.8	3.6	15.40
		Lightning Shoot		
KBCS085-57	0.0	5.6	5.6	3.30
KBCS085-61	0.0	2.0	2.0	27.89
KBCS085-67	0.0	1.1	1.1	44.00
KBCS085-68	0.0	1.2	1.2	12.6
KBCS085-74	2.0	4.28	2.28	8.10
KBCS085-99	2.27	6.35	4.08	5.06
		Main Shear - SW		3.00

9.4.5 Hand Augering

A hand auger sampling program encompassing a total of 118 drill holes with a cumulative meterage of 390.89 m was implemented at Akwadum South (Zone 7) during 2018 and 2019 (Figure 9.16). The auger sampling was designed to test the subsurface signature of the gold-insoil anomaly to identify follow up scout pitting and/or trenching targets. The sampling was carried out with a locally fabricated cutting tool made from a used drill rod with a bevelled cutting edge. The cylindrical sample barrel was driven into the ground to recover the sample at 1m intervals. Auger drill holes were sunk at 12.5 m spacing and typically to a depth of 3 m - 5 m (down to saprolite).

Of the 288 samples collected from the auger drill holes: 65 samples yielded less than 0.01 g/t gold; 150 samples returned gold values from 0.01 g/t to 0.1 g/t; 70 samples between 0.1 g/t and 1.0 g/t gold; and 3 samples above 1.0 g/t gold (4.68 g/t maximum). The auger sampling helped delineate the in situ, saprolitic bedrock, signature of the northeast extremity (250 m) of the Akwadum South gold-in-soil trend. With five consecutive auger drill holes covering a 50 m distance along Line 11900 N (i.e., 12.5 m spacing) yielding metal factors (grade x length) ranging from 0.26 – 1.81 m-g/t.



9.4.6 Scout Pitting

Exploration efforts included 62 scout pits designed to follow up on soil geochemistry and hand auger anomalies, and to help define the litho-structural setting of grassroots targets (Table 9.5 and Figure 9.16). The manually excavated pits were normally 2m long x 1m wide x 1.5m - 5m deep (down to saprolite). In addition to horizontal channel sampling, pits where shallow-dipping quartz veining and/or shearing was observed were also subjected to vertical channel sampling.

Table 9.5: Summary of Scout Pitting (2016 - 2017)						
Scout Pitting No of Pits Vertical Metres Horizontal Metres						
Zone 4 (2016)	8	19.35	15.75			
Akwadum North (Zone 8)	35	81.72	73.98			
Akwadum South (Zone 7) 2017	19	60.65	38.56			
Apapam Total Pitting:	62	161.72	128.29			

Of the total 233 samples collected from the pits: 52 samples yielded less than 0.01 g/t gold; 110 samples returned gold values from 0.01 g/t to 0.1 g/t; 64 samples between 0.1 g/t and 1.0 g/t gold; and 7 samples above 1.0 g/t gold (4.41 g/t and 27.37 g/t maximum values).

At Akwadum South (Zone 7), the scout pitting successfully confirmed the anomalous hand hander sampling results. With five consecutive pits covering a 50 m distance along Line 11300 N (i.e., 12.5 m spacing) yielding metal factors (grade x length) ranging from 0.69 - 4.85 m-g/t; and six of seven pits over a 75 m distance on Line 11700 N returning metals factors ranging from 0.51 - 1.96 m-g/t. The pitting also permitted the delineation of the volcaniclastic rock unit hosting the gold mineralization.

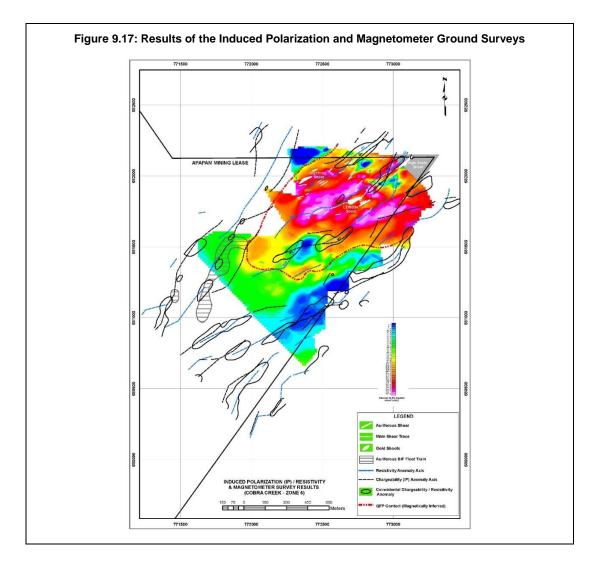
9.4.7 Ground Geophysics

Ground geophysics was conducted at Cobra Creek in 2013 (Figure 9.17). The work consisted of 32.2 line-kilometres of IP/Resistivity survey and 14.99 line-kilometres of magnetic survey. The IP/Resistivity survey utilised a Pole-Dipole Array with a dipole length of 50 m and dipole separations of n=1 to 10, and was undertaken by Sagax Afrique S.A. The magnetic survey was conducted by Xtra-Gold utilising a single Model G-856AX portable proton magnetometer. Magnetic readings were collected at 5 m stations along 100 m spaced grid lines with diurnal drift in the earth's magnetic field corrected via repeated sampling at a specific location during the survey.

Auriferous shear zones at Cobra Creek exhibit a strong spatial association with two prominent ENE-NE high resistivity trends appearing to reflect broad zones of strong iron carbonate (± silica) alteration; with gold mineralization traced over an approximately 850 m distance along the approximately 1,100 m long R2 resistivity trend (i.e., Main Shear structure). Resistivity trends spatially associated with the auriferous shear system appear to be abutting against and/or bending into an inferred NNE-trending second order fault defined by a high chargeability / weak-moderate resistivity domain occupying the western margin of the survey grid.

Consistent with the shear system orientation, the host QFP body exhibits a strong ENE-NE magnetic fabric exemplified by a series of low magnetic susceptibility trends separated by discontinuous, lozenge-shaped, higher susceptibility domains. The auriferous shear structures, as well as the associated resistivity trends, tend to be spatially associated with areas of low magnetic intensity and/or along the margins of the higher susceptibility domains. With the low magnetic susceptibility trends apparently attributable to magnetite-destructive shearing and alteration processes.

A deep-rooted (>225 m), moderate IP chargeability anomaly (M2) is spatially associated with the high grade gold shoot and tourmaline zone at the NE extremity of the prominent main shear structure resistivity trend (R2). The approximately 350 m long, NE-trending, coincident chargeability / resistivity anomaly appears to exhibit a steep northwesterly dip and moderate SW plunge; with the upper margin of the chargeable body lying at a vertical depth of approximately 125 m. The deeper M2 chargeability target remains untested but drill testing of the adjacent M3 chargeability anomaly, appearing to represent the up-dip, near surface expression of the M2 anomaly, intersected a proto-mylonite zone exhibiting barren pyrite mineralization.



9.4.8 Structural Study

TECT Geological Consulting has undertaken a structural re-interpretation of regional VTEM and radiometric geophysical data across all of Xtra-Gold's concessions (TECT Geological Consulting, 2020a; Figure 7.5), a preliminary structural mapping and 3D modelling study of Zone 5 (Cobra Creek Prospect, **Error! Reference source not found.**, TECT Geological C onsulting 2019) and a comprehensive structural mapping and 3D modelling exercise for Zones 1-4 (TECT Geological Consulting 2020b). A summary of the relevant findings is presented in Chapter 7, with the omission of Zone 5 (Cobra Creek), which does not form part of the present resource update/declaration.

Regardless, recent work has been consolidated to provide guidance to the exploration activities for the Cobra Creek Deposit, as this prospect host significant Au anomalies (TECT Geological Consulting 2019) (Figure 9.18). Preliminary findings by TECT Geological Consulting have suggested the following:

Structural Data Analysis and Interpretation

- Locally S₁ (gneiss) and S_{2m} (mylonite) trends within the hosting quartz feldspar porphyry (QFP) body are consistent with fabric trends interpreted from IP-Res and ground based magnetics survey datasets, and overall the D₂ structural evolution described for the Kibi Project (see Chapter 7.5);
- On an outcrop scale, individual S_{2m} shears exhibit an anastomosing geometry that, in part, envelope lenses of relict QFP, thereby creating low-strain shadows associated with lozenges. S_{2m} records distinct reverse (i.e., SW-up), left-lateral kinematic (i.e., oblique) movement, that is consistent with regionally-interpreted NNE-trending D₂ structure (see Chapter 7.5).
- Auriferous veins consist of fault-fill/shear, extension and hybrid extensional-shear veins of varying proportions of quartz-carbonate±tourmaline, which were cotemporaneous with the development of S_{2m}. Mineralization occurs in close association with late-stage Ser-Cb-Po-Py (±Chl) alteration and as native/visible gold;
- Syn-deformational (S_{2m}) hydrothermal fluid flow, veining and auriferous mineralization occurred within a sub horizontal- to shallow-dipping stress field, with a principle shortening axis trending NW to NNW:

Preliminary Model of Controls on Mineralization and Exploration

- Syn-deformational (S_{2m}) fault-fill/Shear, extension and hybrid extensional-shear veins exhibit directional brittle fracturing, interconnectivity and directional permeability, leading to the development of stockwork geometries representative of echelon arrays (planes). Along these planes, stockworks have a preferred directional (linear) permeability or shoot geometry along very shallow (<20°) to sub-horizontal plunges towards the SW (236° 228°).
- Work towards characterizing the extent and geometry of the host QFP intrusion within the volcano-sedimentary Kibi Belt to establish QFP (sheared) contacts and possible strain

Figure 9.18: Structural Mapping of the Cobra Creek Target

High Grade Shoot West

Tourmaline Zone South East

shadows or dilational zones (jogs/bends), which would be highly favourable sites for more continuous and enhanced Au mineralization;

9.5 2021 – 2023 Exploration Programme

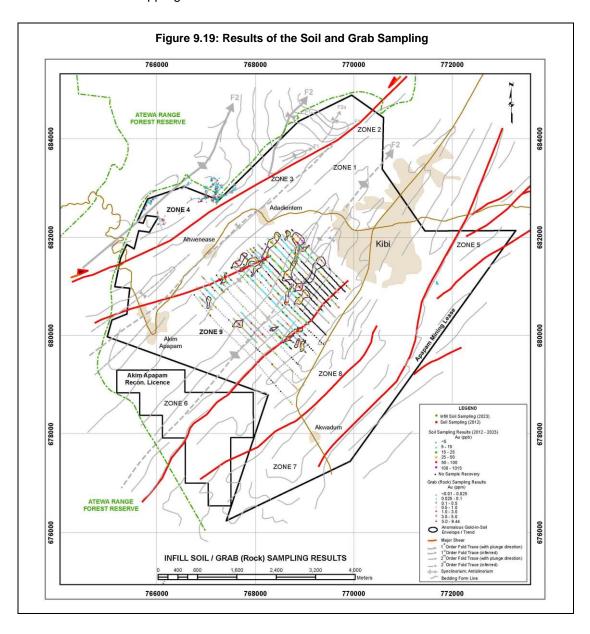
9.5.1 Soil Sampling

Further soil sampling was conducted on the Zone 9 gold-in-soil anomaly in 2023 with the aim of ground proofing an approximately 2 km segment of the grassroots Central Fold target generated by the 3D VTEM / TMI inversion modelling (Figure 9.19). A total of 542 samples was collected from 17.3 line-km of cut lines on a 100 m x 25 m infill sampling grid (i.e., infilling of 2012 sampling at 200 m line-spacing). A total of 165 sites representing disturbed ground (artisanal mining) and/or low swampy terrain along the 17.3 line-km of gridlines could not be sampled. Soil sampling methodology consist of the collection of approximately 2.5 kg of regolith material from hand-dug pits at depths of 50 to 60 cm with normal diameters not exceeding 30 cm using the soso digging tool.

The infill soil sampling permitted the further delineation of a series of linear gold-in-soil trends ranging from approximately 300 m to 1,000 m in length. The NE-trending gold-in-soil anomalies appear to exhibit a broad spatial relationship with inferred parasitic folds associated with the 1st-order F2 anticlinorium fold structure generated from 3D geophysical modelling. Grab sampling and scout trenching delineated an approximately 650 m long quartz vein

structure spatially associated with the most prominent gold-in-soil trend (Items 9.5.2 and 9.5.3). With the remaining gold-in-soil anomalies yet to be subjected to follow up work.

The gold-in-soil trends are defined by envelopes of discontinuous / patchy, typically greater than 25 ppb, gold-in-soil values; with the anomalous threshold arbitrarily set at 25 ppb gold based on past exploration experience by Xtra-Gold in the Kibi Greenstone Belt. Of the 542 soil samples collected, 81 samples yielded less than 5 ppb gold; 323 samples returned gold values from 5 ppb to 25 ppb gold; 117 samples from 26 ppb to 100 ppb gold; 16 samples from 101 ppb to 300 ppb gold; and 5 samples returned greater than 300 ppb gold, including a maximum value of 756 ppb gold.



9.5.2 Grab Sampling

Traditional prospecting, including the collection of 611 composite chip samples (i.e., grab samples) from rock float and outcrop, was performed on grassroots targets in 2023. With 438 samples collected from the Zone 4 auriferous silicified siltstone float target, 165 samples from the Central Fold (Zone 9) target and eight (8) samples from other isolated locations (Figure 9.19). Cautionary Note: Grab samples are selective in nature and may not be necessarily representative of the mineralization present on the concession.

Prospecting efforts in Zone 4, covering an area of approximately 2 km², successfully expanded the auriferous silicified / pyritized siltstone rock float target originally discovered in 2006. Systematic grab sampling identified three well-defined auriferous rock float fields, including: an approximately 375 m by 500 m, gold-in-float anomaly exhibiting a spatial association with interpreted bedding / S1 form lines, appearing to demark a possible fold nose structure (i.e., Float Target 1); and Float Target 2, an approximately 375 m long, NE-trending gold-in-float anomaly lying approximately 900 m west of Float Target 1. Of the 438 silicified siltstone float samples collected, 154 samples yielded less than 0.01 g/t gold; 137 samples returned gold values from 0.01 g/t to 0.1 g/t; 107 samples between 0.1 g/t and 1.0 g/t gold; 29 samples between 1.0 g/t to 3.0 g/t gold; and 11 samples returned values above 3.0 g/t gold, including a maximum value of 9.44 g/t gold. In situ source of auriferous siltstone material yet to be established by trenching / drilling.

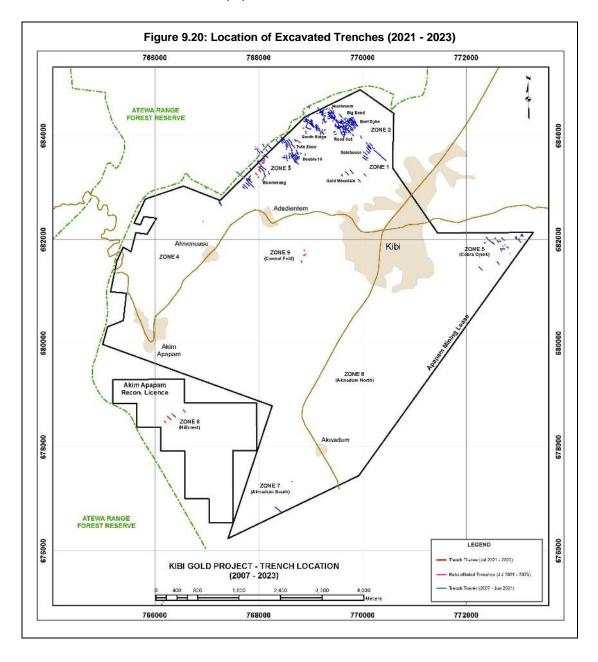
<u>Cautionary Note:</u> Readers are cautioned that the northeastern half of Float Target 1 (112 samples) lies in the Forest Reserve, outside the Apapam concession, and the Company does not currently have the rights to explore or mine this potential mineralization zone. An application for a Prospecting Permit to conduct mineral exploration activities within the fringes of the Atewa Forest Reserve has been submitted to the Minerals Commission of Ghana by the Company. Although Forest Reserve prospecting permits are routinely granted by the Ghana Government, there is no assurance that the Company will ever be granted a permit to explore or mine these potential mineralization zone (Item 4.4.4 for details on the Forest Reserve Prospecting Licence Application).

A total of 165 composite chip samples (i.e., grab samples) was collected in support of the ground proofing of the Central Fold (Zone 9) target, an inferred 1st-order F2 anticlinorium fold structure generated from 3D geophysical modelling. The grab samples were collected from float and outcrop along an approximately 650 m strike-length of a prominent quartz – iron carbonate – tourmaline vein structure exhibiting a coincidental gold-in-soil anomaly. With the quartz vein structure appearing to occupy the hinge of an inferred parasitic fold associated with the larger anticlinorium. Of the 165 quartz veining samples collected: 152 samples yielded less than 0.01 g/t gold; 11 samples returned gold values from 0.02 – 0.04 g/t gold; and two samples returned values of 0.1 g/t gold.

9.5.3 Trenching

Trenching was primarily undertaken in Zone 3 to supplement the drilling of the targets already recognised. Trenches, including road cut sampling, were excavated on three of the targets

that form part of the present mineral resource estimate, including: the new Boomerang and Twin Zone targets and the existing Double 19 target. Scout trenching was also conducted on the Central Fold (Zone 9) grassroots target generated by the 3D VTEM / TMI inversion modelling. The trenching details are summarised in Figure 9.20 and Figure 9.139.20 depicts the location of the trenches on the Apapam concession.



Tal	ole 9.6: Summary of Trenchi	ng July 2021 - December 2	023
Trer	Trench Type		Trench Metres
Trench	New	8	303.8
Trench	Rehabilitation	5	92.6
Road Cut Sampling	New	12	336.5
Total Trenching -	- Apapam Concession	25	732.9
Trenching / Road Cut Sampling Per Target			
Double 19	Existing Resource Target	1	21.8
Total Trenching - Existing Resource Targets		1	21.8
Boomerang	New Resource Target	15	395.4
Twin Zone	New Resource Target	5	186.4
Total Trenching - New Resource Targets		20	581.8
Central Fold (Zone 9)	Grassroots (Scout) Target	4	129.3
Total Trenching – Gr	assroots (Scout) Targets	4	129.3

The trenches were either mechanically or manually excavated. The information relating to the methods of excavation, trench dimensions, surveying, mapping and sampling and rehabilitation are detailed in Item 9.4.3.

Significant trench sampling results from the Boomerang target are presented in Table 9.7. Trenching efforts from the Twin Zone and Double 19 targets failed to return any significant assay results.

Table 9.7: Significant Trench / Road Cut Results – Boomerang Target (2021 – 2023)								
Trench / Road Cut ID	From (m)	To (m)	Trench-Length (m)	Au (g/t)				
	Trench Sampling							
TAD014A	4.7	18.7	14.0	0.71				
including	11.0	15.0	4.0	1.01				
TAD014B	1.7	12.0	10.3	1.04				
including	7.0	9.0	2.0	1.46				
TAD014C	1.0	8.0	7.0	0.57				
including	3.8	5.4	1.6	1.25				
TAD016A	1.6	16.6	15.0	1.16				
including	11.0	12.0	1.0	5.01				
TAD018B	2.5	3.5	1.0	92.32				
	Road Cut Sampling							
ADRS023	5.0	6.8	1.8	2.50				
ADRS024	14.0	18.5	4.5	0.97				
including	14.0	15.0	1.0	2.85				
ADRS028	0.0	6.6	6.6	0.41				
including	2.3	3.0	0.7	1.67				
ADRS029	0.0	12.0	12.0	1.70				
including	0.0	0.9	0.9	4.58				
ADRS031	8.9	15.0	6.1	0.58				
including	13.0	14.0	1.0	2.19				

Significant mineralized intercepts meet the following criteria: minimum metal factor (grade x length) of 2.5; with minimum 0.25 g/t gold average grade over interval. Intercepts also constrained with a 0.25 g/t gold minimum cut-off grade at top and bottom of intercept, with no upper cut-off applied, and maximum of five (5) consecutive samples of internal dilution (<0.25 g/t gold). All internal intervals above 5 g/t gold indicated. Reported trench results correspond to trench-lengths in metres. The orientation / geometry of the mineralization is not fully understood in the saprolitic trench exposures resulting in unknown true thickness.

Scout trenching was conducted on the Central Fold (Zone 9) target to test a prominent, NEtrending, quartz – iron carbonate – tourmaline vein structure traced over an approximately 650 m strike-length by rock float and outcrop grab sampling (Figure 9.19). With the quartz vein structure appearing to occupy the hinge of an inferred parasitic fold associated with an interpreted 1^{st} -order F2 anticlinorium fold structure generated from 3D geophysical modelling. The quartz structure was tested over an approximately 235 m strike-length by four manual trenches ranging from 9 m to 64 m in length. Of the 113 quartz vein and metasedimentary wall rock samples collected: 68 samples yielded less than 0.01 g/t gold; 41 samples returned gold values from 0.01 - 0.1 g/t gold; and 4 samples returned values above 0.1 g/t gold, including a maximum value of 0.29 g/t gold.

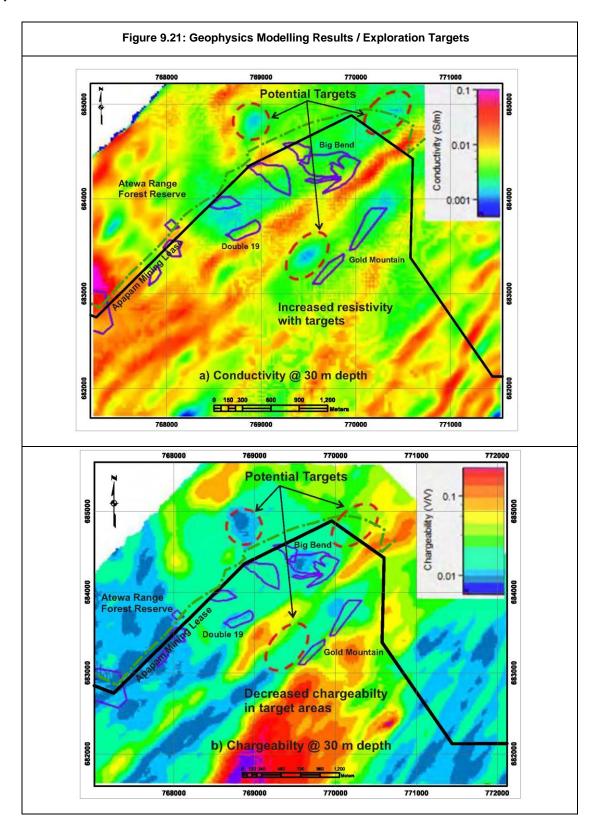
9.5.4 Airborne Geophysics Modelling

In 2022, Technolmaging LLC of Salt Lake City, Utah, USA, undertook 3D geophysical modelling of the broader Apapam concession area (~70 km²), encompassing a 585 line-km subset of the over 4,000 line-km regional helicopter-borne VTEM — Mag survey, completed by Geotech Airborne Limited in 2011 (Item 9.3.1). Geophysical modelling included 3D joint inversion for conductivity and chargeability of the VTEM survey data, as well as 3D inversion of the Total Magnetic Intensity (TMI) to magnetic susceptibility and magnetization vector models. The modelling work was undertaken to generate high-priority exploration targets to help guide ongoing resource expansion drilling and to support 3D litho-structural modelling efforts.

The remanent component of the magnetization vector and the chargeability inversion of the VTEM produced the best correlations with known geology. The remanent component of the magnetization vector accurately mapped the auriferous shear-hosting Cobra Creek quartz feldspar porphyry (QFP) body. The airborne derived chargeability exhibits a very good correlation with ground-based chargeability from Induced Polarization (IP) surveys.

From a mineralization targeting viewpoint, the VTEM-derived chargeability exhibits a negative correlation (i.e., chargeability decrease) with some of the known mineralization bodies, including the Big Bend and Double 19 mineral resource bodies. A broad resistivity increase (i.e., conductivity decrease) also tends to be associated with the general footprint of the known mineralization zones. Geophysical interpretation by Technolmaging generated three exploration targets within the mineral resource footprint area, including a high-priory, coincidental low chargeability / high resistivity anomaly lying to the northwest of the Zone 2 mineral resource bodies, and a high resistivity target located to the northwest of the Gold Mountain resource body

in Zone 1 (Figure 9.21). None of these exploration targets have been subjected to field proofing yet.

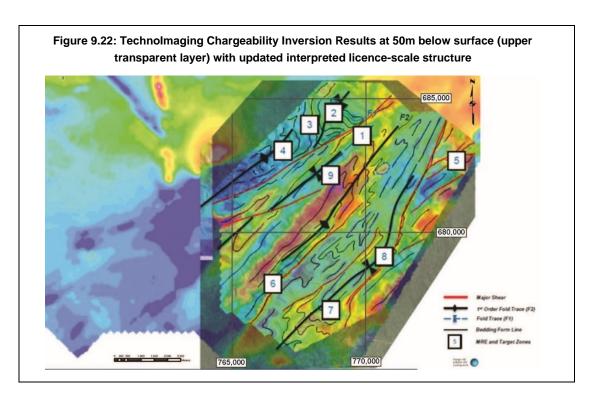


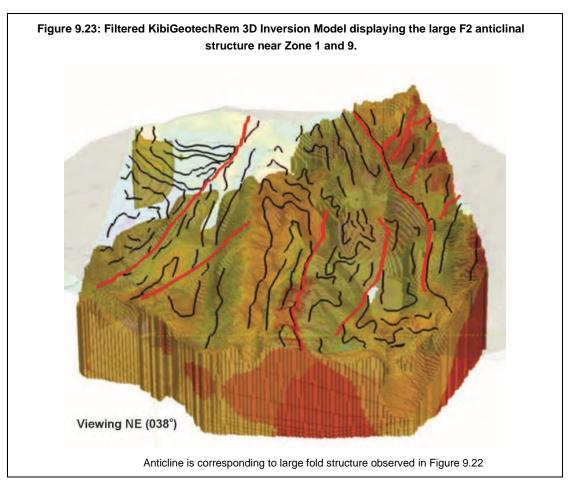
<u>Cautionary Note:</u> Readers are cautioned that two of the above exploration targets lie in the Forest Reserve, outside the Apapam concession, and the Company as presently no rights to explore or mine these potential mineralization zones. An application for a Prospecting Permit to conduct mineral exploration activities within the fringes of the Atewa Forest Reserve has been submitted to the Minerals Commission of Ghana by the Company. Although Forest Reserve prospecting permits are routinely granted by the Ghana Government, there is no assurance that the Company will ever be granted a permit to explore or mine these potential mineralization zone (Item 4.4.4 for details on the Forest Reserve Prospecting Licence Application).

9.5.5 Structural Study

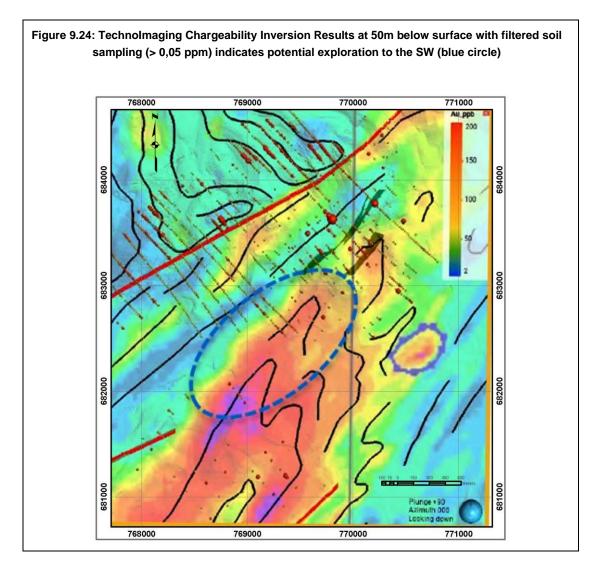
TECT Geological undertook additional structural interpretation and 3D litho-structural modelling covering an area of approximately 41 km² based on additional drilling from July 2021 – January 2023 which considered of 90 diamond drill holes (15,551.5 m) in combination with the 3D VTEM / TMI inversion models produced by Technolmaging. The intention being to further define the structural controls of the gold mineralization and to generate high-priority exploration targets to help guide ongoing resource expansion drilling. The resultant models formed the basis for the 2024 Mineral Resource estimate.

Zones 1 and 9 are broadly hosted within a tectonostratigraphy that defines a 1st-order F2 anticlinorium, plunging to the NE (Figure 9.22). This anticline is identifiable on the Technolmaging Chargeability Inversion Results at 50m below surface (Figure 9.23) and on the KibiGeotechRem 3D Inversion Model. The large anticline is associated with smaller-scale parasitic folds. The anticline is comprised of metavolcanics surrounded by K-rich metasediments and in the north bound by metasediments. The F2 anticlinal structure is bound to the east by a D2 shear (Figure 9.22 and Figure 9.23).

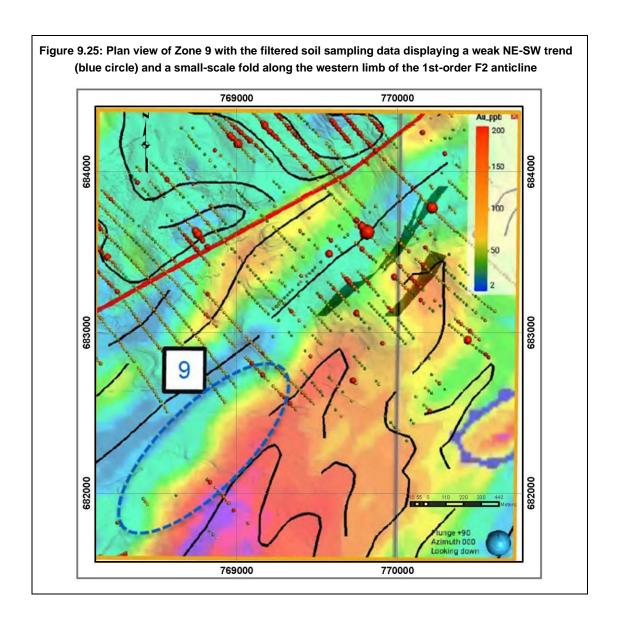




Gatehouse and Gold Mountain (Zone 1) and Zone 9 are situated on the western limb of the F2 anticline (Figure 9.24), potentially related to limb-parallel shears. The filtered regional soil sampling (> 0,05 ppm) indicates that southwest of Gatehouse and Gold Mountain, along strike and following the same northwestern limb, there is potential for exploration (Figure 9.24). This should include field mapping, grab samples, and trenching between and parallel to the soil sampling in order to gain knowledge on rock types, foliation and quartz veins, subsequently followed by confirmational drilling

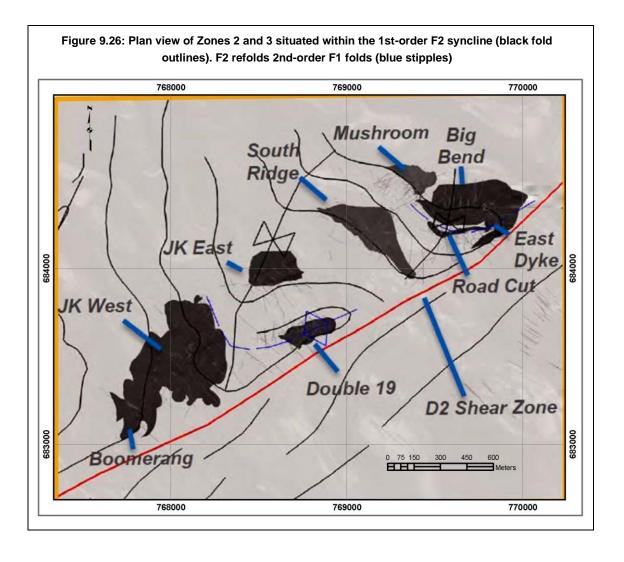


Zone 9 is located closer to the hinge of the 1st-order F2 anticlinorium and along the western limb of the fold (Figure 9.25 and Figure 9.26); here several similar NE-trending soil-geochemical Au anomalies are observed. These are relatively weak, but their trend matches deposit geometries. There is also smaller-scale folding present that could complicate the deposit geometries (Figure 9.26). Further exploration or targeting is advised along NE-striking anomalies, viz. field mapping, trenching and grab sampling.



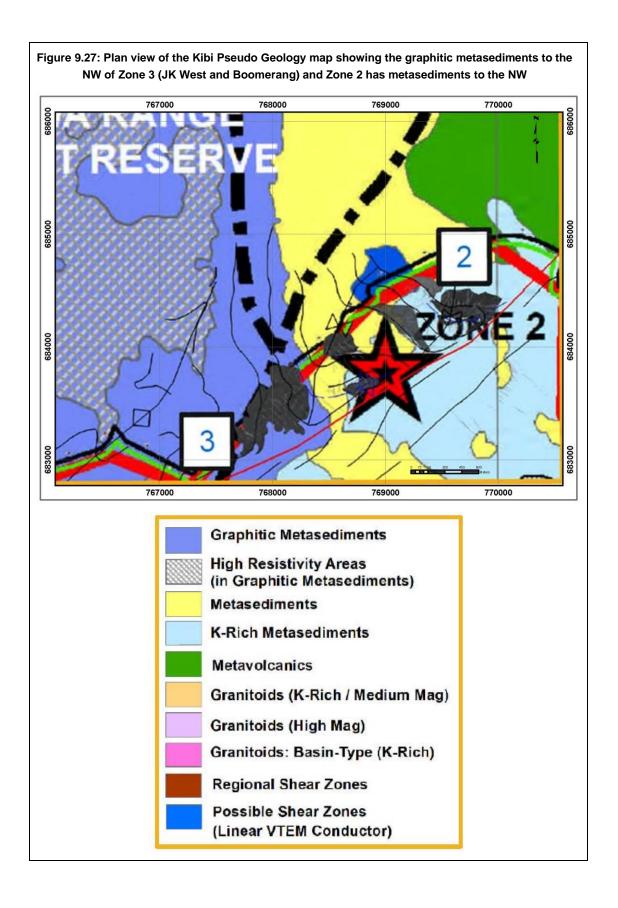
Zones 2 and 3 are situated on the western side of the D2 shear, hosted within a broad tectonostratigraphic package, characterized by interlayered volcano-sedimentary and granodiorite / granite units. The tectonostratigraphy forms part of a licence-scale 1st-order F2 syncline, where the F2 folding refolds the 2nd-order F1 folds (Figure 9.27). Kinematics on the D2 shear are sinistral (left lateral).

Deposits are consistently constrained within or associated with F1- or F2-folded and/or sheared granodiorite units interlayered with volcano-sedimentary contacts. This scenario is, for all intents and purposes, the loci for brittle-ductile dilation that induces hydrothermal fluid-flow and associated Au mineralization.



In this context, and to seek new deposits and expand Zones 2's and 3's overall in-situ resource, the most prospective area to continue exploring would be just NW of the Apapam Mining Lease, following the 1st-order F2 folded stratigraphy, along strike from proven deposits. Around Zone 2 there is regional soil sampling showing potential extension to the NW (Figure 9.26) along with some of the ore deposits not fully constrained by drilling. On the regional Kibi Pseudo Geology map there are metasediments to the NW of Zone 2 (Figure 9.27).

Many of the JK West deposits (Zone 3) are open along strike to the NW and have not been constrained by drilling. One thing to note, on the Kibi Pseudo Geology map there is mainly graphitic metasediments to the NW of Zone 3 (Figure 9.28. These sediments are characteristically ductile and would need to be interlayered with competent units in order to form brittle-ductile shears/fractures to host Au mineralization. We would suggest trenching, soil sampling, and subsequent drilling to the NW of Zones 2 and 3 to establish potential extensions.



9.6 Akim Apapam Reconnaissance License Application

9.6.1 2012 – 2021 Exploration Program

Exploration efforts on the Akim Apapam Reconnaissance License Application, including infill soil sampling, trench rehabilitation / deepening, scout pitting and hand auger sampling, were geared towards the further definition of the Hillcrest (Zone 6) target. An approximately 1,200 m long by 100 m – 200 m wide anomalous gold-in-soil trend spatially associated with a prospective lithostructural setting consisting of an inferred NE-trending regional fault bounding an apparent Belt-type granitoid (i.e., high magnetic susceptibility body) at the intersection with a series of SE-trending linking faults.

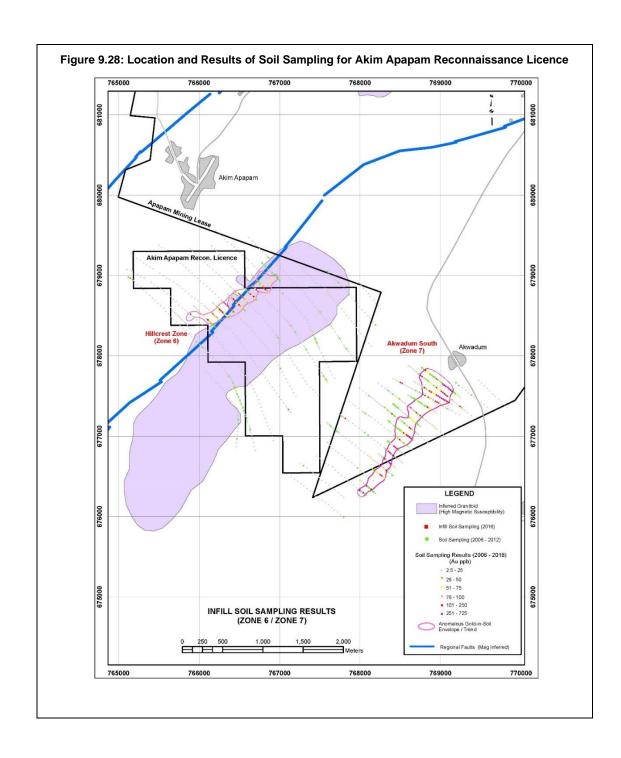
A total of 2.25 line-kilometres of infill soil sampling was completed in 2016 to provide tighter coverage of the gold-in-soil anomaly at 100 m line spacing (i.e., infilling of 2012 sampling at 200 m line spacing). A total of 97 soil samples were collected at 25 m stations along the NW-SE grid lines (Figure 9.28). The Hillcrest gold-in-soil trend is defined by an envelope of discontinuous/patchy, typically greater than 50 ppb, gold-in-soil values; with the anomalous threshold arbitrarily set at 50 ppb gold based on past exploration experience by Xtra-Gold in the Kibi Greenstone Belt. Out of the 80 samples within the anomalous trend, including both 2012 and 2016 sampling, 58 samples (72%) yielded gold values greater than the 50 ppb anomalous threshold, including: 29 samples from 51 ppb to 100 ppb gold; 23 samples from 101 ppb to 300 ppb gold; and 6 samples above 300 ppb gold (493 ppb maximum).

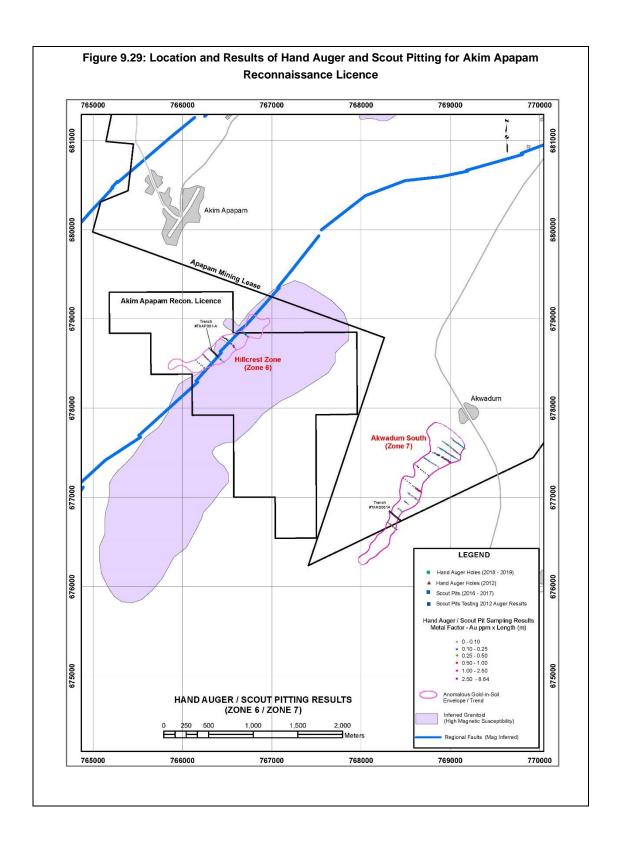
A 108.6 m section of an existing 2012 trench was manually rehabilitated / deepened in 2017 to provide continuous saprolitic bedrock exposure for detailed mapping / sampling of the Hillcrest shear structure (Figure 9.29). Of the 98 samples were collected from the deepened #TAAP001-A trench: 54 samples yielded gold values from 0.01 g/t – 0.1 g/t, 30 samples between 0.1 g/t to 1.0 g/t gold; 9 samples from 1.0 g/t to 5.0 g/t gold; and 5 samples above 5.0 g/t gold (10.7 g/t maximum). Trench #TAAP001-A delineated an approximately 45 m wide structural zone encompassing a series (7) of parallel, SE-dipping (30° – 40°), auriferous shears ranging from approximately 0.5 m to 4 m in trench-length (Figure 9.30). Channel sampling of the iron carbonate-amphibole-quartz schist seams returned mineralized intercepts ranging from 0.5 m grading 0.35 g/t gold to a high of 5.0 m grading 3.27 g/t gold. Note that the above auriferous intercepts represent trench-lengths with true widths estimated to be 50% - 60% of the reported intervals.

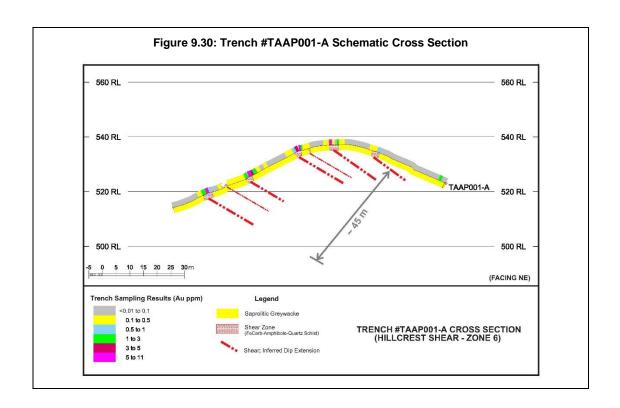
Additional 2017 work on the Hillcrest target included the excavation of 21 scout pits totalling 66.3 vertical metres and 40.7 horizontal metres (Figure 9.30). The pitting was designed to follow up on soil geochemistry and/or hand auger anomalies, and to help define the mineralization zone's litho-structural setting. Of the 42 samples collected from the pits: 15 samples yielded less than 0.01 g/t gold; 17 samples returned gold values from 0.01 g/t to 0. g/t; seven (7) samples between 0.1 g/t and 1.0 g/t gold; and three (3) samples returned maximum values of 1.81 g/t, 3.88 g/t, and 5.0 g/t gold respectively. Scout pitting efforts successfully traced the Hillcrest shear structure approximately 200 m northeast of trench #TAAP001-A. With

iron carbonate-amphibole-quartz schist seams in pits #PKB062 and #PKB063, located 25 m apart, returning 5.0 g/t gold and 1.81 g/t gold over 1.0 m and 1.1 m trench-lengths respectively.

A hand auger sampling programme encompassing 17 drill holes totalling 66.1 m was implemented on the Hillcrest zone in 2019 (Figure 9.30). The auger sampling was designed to test the subsurface signature of gold-in-soil anomalies to identify follow up scout pitting and/or trenching targets. Of the 51 samples collected from the auger drill holes: 13 samples yielded less than 0.01 g/t gold; 18 samples returned gold values from 0.01 g/t to 0.1 g/t; 18 samples between 0.1 g/t and 0.35 g/t gold; and 2 samples returned maximum values of 0.9 g/t gold and 1.02 g/t gold. The auger sampling appears to have delineated the apparent southwest strike extension of the Hillcrest shear structure, approximately 100 m southwest of trench #TAAP001-A. With five consecutive auger drill holes covering a 50 m distance (i.e., 12.5 m spacing) yielding metal factors (grade x length) ranging from 0.45 – 2.1 m-g/t.







9.6.2 2021 – 2023 Exploration Program

In 2021, further scout trenching was conducted on the Hillcrest Shear (Zone 6) target located on the Akim Apapam Reconnaissance License Application (Figure 9.21). The grassroots target is characterized by a braided system of auriferous carbonate-amphibole-quartz schist seams, spatially associated with a 1,200 m x 100 – 200 m anomalous gold-in-soil trend (Figure 9.29). With the NE-trending shear structure appearing to straddle the contact between an apparent Belt-type granitoid (i.e., high magnetic susceptibility body) and a metasedimentary rock sequence, along the western limb of an interpreted 1st-order F2 anticlinorium fold structure generated from 3D geophysical modelling (Figure 7.6).

Three (3) manually excavated trenches totaling 201.0 m were completed to follow up on anomalous hand auger and/or scout pitting results. The NW-trending trenches tested an approximately 400 m strike-length of the shear structure, including one trench located approximately 100 m to the northeast of the #TAAP001-A trench completed in 2017, and two trenches located 100 m and 200 m respectively to the southwest of the #TAAP001-A trench. A 30 m section of trench #TAAP003 was deepened by 0.9 metres to permit resampling of more continuous saprolite material (i.e., #TAAP003-A).

The information relating to the methods of excavation, trench dimensions, surveying, mapping and sampling and rehabilitation are detailed in Chapter 9.4.3.

Significant trench sampling results are presented in Table 9.49.8. Significant mineralized intercepts meet the following criteria: minimum metal factor (grade x length) of 2.5; with minimum 0.25 g/t gold average grade over interval. Intercepts also constrained with a 0.25 g/t

gold minimum cut-off grade at top and bottom of intercept, with no upper cut-off applied, and maximum of five (5) consecutive samples of internal dilution (<0.25 g/t gold). All internal intervals above 5 g/t gold indicated. Reported trench results correspond to trench-lengths in metres. The orientation / geometry of the mineralization is not fully understood in the saprolitic trench exposures resulting in unknown true thickness.

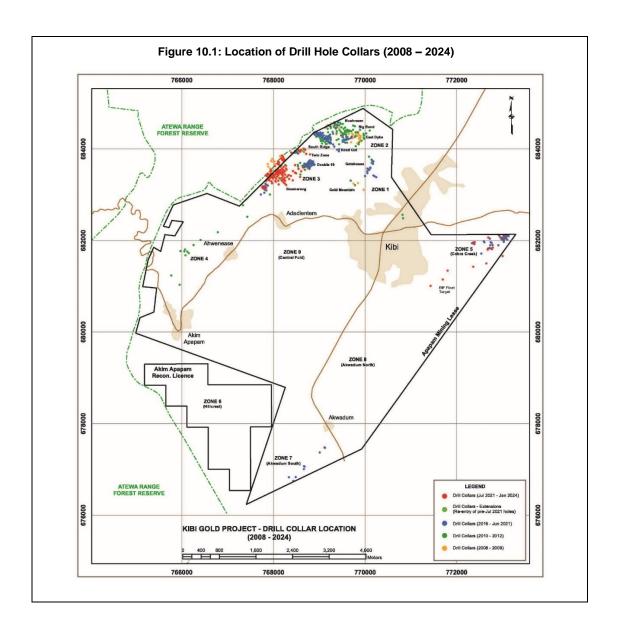
Table 9.8: Significant Trench Results – Hillcrest Shear (Zone 6) Target						
From (m)	To (m)	Trench-Length (m)	Au (g/t)			
8.5	20.5	12.0	0.39			
8.5	9.5	1.0	1.67			
9.5	11.0	1.5	1.89			
17.7	30.8	13.1	0.72			
20.6	21.4	0.8	2.22			
27.0	27.9	0.9	3.00			
53.0	84.1	31.1	0.39			
81.5	83.5	2.0	0.96			
2.0	19.0	17.0	0.52			
17.0	18.0	1.0	1.15			
39.6	47.6	8.0	0.6			
42.0	43.2	1.2	1.72			
	From (m) 8.5 8.5 9.5 17.7 20.6 27.0 53.0 81.5 2.0 17.0 39.6	From (m) To (m) 8.5 20.5 8.5 9.5 9.5 11.0 17.7 30.8 20.6 21.4 27.0 27.9 53.0 84.1 81.5 83.5 2.0 19.0 17.0 18.0 39.6 47.6	From (m) To (m) Trench-Length (m) 8.5 20.5 12.0 8.5 9.5 1.0 9.5 11.0 1.5 17.7 30.8 13.1 20.6 21.4 0.8 27.0 27.9 0.9 53.0 84.1 31.1 81.5 83.5 2.0 2.0 19.0 17.0 17.0 18.0 1.0 39.6 47.6 8.0			

10 DRILLING

Significant drilling has been completed on the Kibi Gold Project. The drilling activity is summarised in Table 10.1 and the location of the drill hole collars are depicted on Figure 10.1.

	Table 10.1: Summary of Drilling						
	Reverse Circulation		Diamond Drill		Total		
Year	No. of Drill Holes	Meterage	No of Drill Holes	Meterage	No. of Drill Holes	Meterage	
2008		-	18	3,001.00	18	3,001.00	
2009	50	4,715.00		-	50	4,715.00	
2010		-	36	6,458.12	36	6,458.12	
2011		-	81	21,878.32	81	21,878.32	
2012		-	71	13,035.40	71	13,035.40	
2016			43	2,639.3	43	2,639.3	
2018		-	26	3,413.60	26	3,413.60	
2019		-	33	3,604.8	33	3,604.8	
2020		-	74	9,615.45	74	9,615.45	
2021			75	12,111.9	75	12,111.9	
2022			79	14,974.5 *	79	14,974.5	
2023			56	13,495.1	56	13,495.1	
Total	50	4,715.00	592	104,227.5	642	108,942.5	

^{* 2022} drill footage includes 165.5 m from the re-entry / extension of a 2012 drill hole



10.1 Drilling Campaigns

10.2 2008 – 2010 Exploration Program

The first drilling campaign on the Apapam Concession area was conducted by Xtra-Gold in 2008. The Xtra-Gold drilling focused on the Kibi Gold Project consisting of a > 5.5 km long mineralized trend delineated from gold-in-soil anomalies, trenching, and geophysical interpretations along the northwest margin of the Apapam Concession; and characterized by widespread gold occurrences of the granitoid hosted-type.

A total of 68 drill holes totalling 7,716 linear metres were drilled on the Apapam Concession, including 18 diamond drill holes in 2008 (3,001 m) and 50 reverse circulation (RC) drill holes in 2009 (4,715 m). The diamond drilling and RC drilling was conducted by Burwash Drilling and Boart Longyear, respectively. The drill holes targeted the Zones 1, 2, and 3 gold-in-soil anomalies.

Diamond drill core was HQ size (63.5 mm diameter) in upper oxidized material (regolith) and NQ2 size (50.6 mm diameter) in the lower fresh rock portion of the drill hole. RC drilling was typically conducted with a 5 ¾ inch diameter bit; but reduction to 5 ½ inch or 5 ¼ inch diameter bits was on occasion required due to rock hardness and/or increasing drill hole depth. Core from five (5) of the 18 diamond drill holes was oriented utilizing the Ezy- Mark core orientation device.

All drill collar locations were surveyed-in by a professional surveyor utilizing a combination of DGPS-established benchmarks and theodolite surveying; and all drill casing (PVC) secured by a cement base.

All drill holes (except for 3 drill holes) were downhole surveyed (azimuth/inclination) at nominal 30 m interval utilizing an electronic single shot survey instrument. Diamond drill and RC drill holes were surveyed with the Flexit and Reflex EZ-Shot tools, respectively. Out of the 68 drill holes, two (2) diamond drill holes were only subjected to dip surveys using the acid etch method due to electronic survey tool technical difficulties, and one (1) short RC drill hole (62 m) was not subjected to any type of downhole survey due to ground collapse.

10.3 2010 – 2012 Exploration Program

A total of 188 drill holes were drilled for 41,372 m of drilling (Table 10.2). From this, 33,961 samples were taken from 39,088 m of core. Core size used was HQ diameter (63.5 mm) in upper oxidized material (regolith) and NQ2 diameter (50.6 mm) in the lower fresh rock portion of the drill hole.

Table 10.2: Summary of Diamond Drilling (2010 - 2012)									
July 25, 2010 to May 28, 2012	Drill Holes	Metres	# Samples	Sample Metres					
Big Bend	44	12,569	10,344	11,628					
East Dyke	18	4,489	3,124	3,804					
Mushroom	18	3,254	2,759	3,200					
South Ridge	24	4,652	4,445	4,586					
Double 19	28	4,925	4,092	4,690					
Other Targets	56	11,483	9,197	11,180					
Total	188	41,372	33,961	39,088					

Core recoveries recorded were generally very good with an average core recovery in the upper regolith (saprolite/transition zone) of 88.7%, and in the fresh rock of 99.6%. Drill core is stored in wooden trays that are stacked in storage sheds. The core was logged, and various structural measurements were taken where possible. Half core sampling of drill core was undertaken. One half of the core was sent to the laboratory and the other half was retained in the core tray for reference. Samples are usually 1 m in length.

10.4 2012 – 2021 Exploration Program

A total of 212 diamond core drill holes totalling 25,198.55 m was drilled on the Kibi Gold Project during the 2012 – 2021 reporting period.

10.4.1 Mineral Resource Area (Zones 1,2,3)

With most of the drilling, 158 drill holes totalling 21,321.45 m (85%), completed from February 2018 – June 2021 on targets within the Zone 1 – Zone 2 – Zone 3 Mineral Resource estimate footprint area (Table 10.3). Samples were taken from 21,742 m of core. Core size used was HQ diameter (63.5 mm) in upper oxidized material (regolith) and NQ2 diameter (50.6 mm) in the lower fresh rock portion of the drill hole. Drilling was undertaken by Xtra-Gold's in-house drilling team utilizing Atlas Copco Christensen CS1000 and Odyssey ODR100 drill rigs.

A detailed review and investigation of structural readings from drill core, trench and outcrops was also undertaken by TECT Geological Consulting (TECT Geological Consulting, 2020b). The body of work was aided by the use of digital mapping devices for field mapping and a Reflex IQ-Logger to obtain more precise and dense structural datasets. These datasets where integrated with 3D Geological/Target Modelling (Chapter 14.2), exploration drillhole targeting and downstream resource estimation (Chapter 14.8 – 14.10).

Table 10.3: Summary of Drilling (2012 - 2021)							
Feb 2018 - Jun 2021	Drill Holes	Drill Metres	Sample Metres				
South Ridge	45	5629.1					
Double 19	40	6,400.95					
Road Cut	18	2446					
Mushroom	5	731					
Gatehouse	16	2,122.5					
Boomerang	17	2,196.7					
JK East	8	986.7					
Other Targets	9	808.5					
Total Drilling:	158	21,321.45 m	17,907.67				
Cobra Creek (Zone 5) Jun - Aug 2016	43	2639.3	2,599.4				
Akwadum South (Zone 7) May - Sep 2019	11	1,237.8	1,236.3				
Apapam Total Drilling:	212	25,198.55 m	21,743.37				

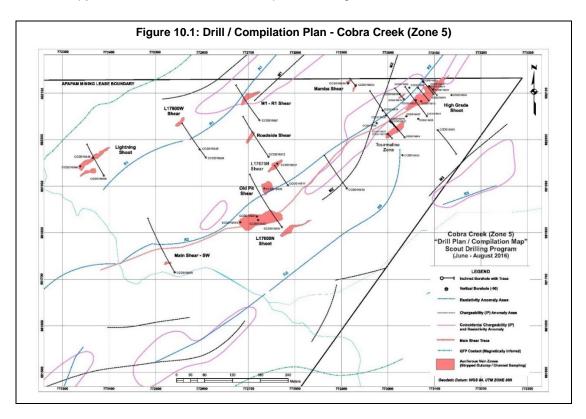
10.4.2 Cobra Creek (Zone 5) / Akwadum South (Zone 7) Scout Drilling Campaigns

Drilling activities during the 2012 – 2021 exploration program also included scout drilling campaigns on the Cobra Creek (Zone 5) and Akwadum South (Zone 7) targets located at the northeastern and southwestern extremities of the Apapam concession respectively (**Error! R eference source not found.**). These early-stage targets, exhibiting distinct litho-structural settings, fall outside the footprint area of the present Zone 1 – Zone 2 – Zone 3 mineral resource estimate.

A 43 drill hole (2,639.3 m) Phase I diamond core drill program was implemented from early June to late August 2016 on the Cobra Creek (Zone 5) target; an approximately 550 m wide, NE-trending, quartz-feldspar porphyry ("QFP") hosted, multi-structure braided shear zone system traced by trenching / outcrop stripping over an approximately 850 m strike length. With the drilling testing an approximately 700 m x 200 m - 300 m segment of the Cobra Creek auriferous structural corridor, down to a maximum vertical depth of approximately 175 m (Figure 10.1).

The Cobra Creek drill program included: 12 scout drill holes ranging from 56 m to 220 m in length (1,576 m) designed to test auriferous shear targets identified by extensive outcrop stripping / channel sampling efforts and priority Induced Polarization ("IP") / Resistivity anomalies spatially associated with auriferous shears; and 31 short, predominantly vertical (-90°) drill holes ranging from 16 m to 63 m in length (1,063 m) designed to better target / dissect flat-lying to shallow dipping gold-bearing extensional veining arrays and/or shallow plunging auriferous shoots.

A scout drilling program was undertaken on the Akwadum South (Zone 7) target located at the southwestern extremity of the Apapam concession from late May to mid-September 2019 (Figure 9.16). A total of 11 diamond core drill holes (1,237.8 metres) ranging from 51.6 m -190 m in length were completed by the company's in-house drilling crew. The scout drilling traced anomalous gold mineralization over an approximately 1,000 m strike-length, and to a vertical depth exceeding 100 m, along a NE-trending volcaniclastic rock package exhibiting widespread silica-sericite-pyrite alteration with associated quartz veining.



Significant drill results for the Cobra Creek (Zone 5) and Akwadum South (Zone 7) targets are summarized in Table 10.4. The reported drill results meet the following criteria: 1 m minimum length and minimum metal factor (grade x length) of 3 m-g/t with minimum 0.3 g/t average grade over the interval or a minimum metal factor of 5 m-g/t if interval falls below the minimum 1 m criteria. In addition, mineralized intercepts are constrained with a 0.25 g/t gold minimum cut-off grade at top and bottom of intercept, with no upper cut-off applied, and maximum of five (5) consecutive samples of internal dilution (less than 0.25 g/t gold). Reported drill results correspond to core-lengths in metres. The orientation / geometry of the mineralization is not fully understood at this time resulting in unknown true thickness.

nole ID	From (m)	To (m)	Core-Length (m)	Au (g/t)				
Cobra Creek (Zone 5)								
016002	3.0	4.5	1.5	19.50				
016004	31.7	32.3	0.6	8.05				
016004	45.6	52.2	6.6	1.00				
16005	128.6	129.5	0.9	6.31				
16009	28.4	30.7	2.3	2.68				
016010	115.0	115.6	0.6	9.95				
016012	110.0	141.0	31.0	0.36				
uding	122.4	123.3	0.9	2.40				
016013	0.0	5.5	5.5	6.57				
uding	2.5	4.5	2.0	11.70				
016015	1.0	6.2	5.2	9.51				
uding	4.6	5.7	1.1	37.95				
016016	4.5	7.2	2.7	3.00				
uding	4.5	5.0	0.5	8.08				
016016	19.6	34.4	14.8	1.43				
uding	28.2	30.2	2.0	4.39				
016018	30.0	31.5	1.5	3.54				
016020	7.1	11.6	4.5	10.90				
uding	10.0	10.6	0.6	57.08				
016021	24.5	26.8	2.3	2.41				
016022	1.5	3.0	1.5	48.10				
016022	12.0	13.2	1.2	7.27				
016023	4.0	9.3	5.3	4.46				
016024	10.4	18.0	7.6	2.09				
uding	14.2	14.8	0.6	7.67				
016024	27.6	28.3	0.7	58.73				
016025	0.5	8.5	8.0	2.93				
uding	1.1	2.0	0.9	9.61				
016037	18.0	22.0	4.0	1.15				
016039	0.0	4.0	4.0	1.57				
016042	3.0	4.5	1.5	3.10				
	Ak	wadum South (Zon	e 7)					
19002	35.5	52.7	17.2	0.37				
19002	71.8	72.8 1.0		4.19				
19002	79.0	9.0 80.0 1.0		3.25				
019002 019004	79.0 71.0	80.0 74.0	1.0 3.0					

Table 10.4: Significant Drill Results - Cobra Creek (Zone 5) / Akwadum South (Zone 7)								
Drill hole ID	Drill hole ID From (m) To (m) Core-Length (m) Au (
AKDD19006	28.0	53.0	25.0	0.47				
including	51.2	52.0	0.8	2.98				
AKDD19007	62.0	73.5	11.5	0.62				
including	63.7	64.5	1.8	1.45				
AKDD19008	50.5	52.0	1.5	4.05				
AKDD19011	42.0	54.5	12.5	0.37				

10.5 2021 - 2023 Drilling Program

A total of 174 diamond core drill holes totalling 34,737.1 m was completed on the Kibi Gold Project from July 2021 – January 2024. The details are presented in **Error! Reference source n ot found.** Drilling utilised HQ diameter size core (63.5 mm) in the upper oxidized material (regolith) and NQ2 diameter size core (50.6 mm) in the lower fresh rock portion of the drill hole. Approximately 93% of the drill core was sampled (32,264 m). The drilling was implemented by Xtra-Gold's in-house drilling team utilizing Atlas Copco Christensen CS1000 (x 2) and Odyssey ODR100 drill rigs.

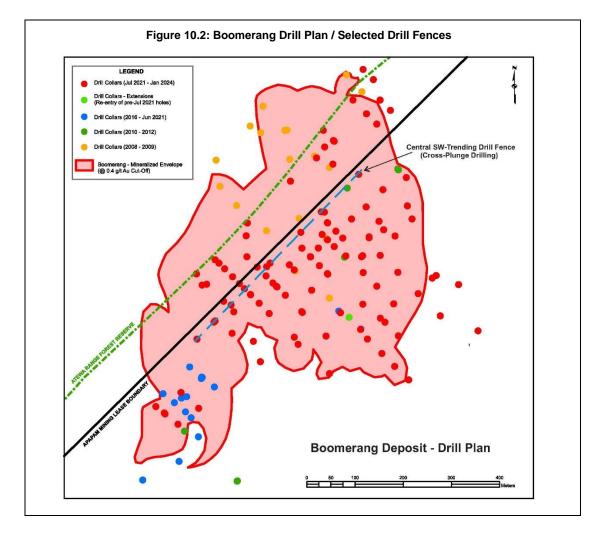
Any drill results reported in Item 10.5 meet the following Significant Result criteria: minimum metal factor (grade x length) of 2.5; with minimum 0.25 g/t gold average grade over interval. Intercepts also constrained with a 0.25 g/t gold minimum cut-off grade at top and bottom of intercept, with no upper cut-off applied, and maximum of five (5) consecutive samples of internal dilution (<0.25 g/t gold). All internal intervals above 15 g/t gold indicated. Reported drill results correspond to core-lengths in metres. The orientation / geometry of the mineralization is not fully understood at this time resulting in unknown true thickness.

Table 10.5: Summary of Drilling (2021 - 2024)								
Concession	No of Drill Holes	Drill Metres	Comments					
Apapam Lease	147	30,184.5	Drill meterage includes three (3) drill hole extensions totalling 222.0 m (re-entry of pre-July 2021 drill holes)					
Buffer Zone	27 4,552.6		Drill holes collared in area between concession and forest reserve boundaries					
Total Drilling (Lease Area)	174	34,737.1	~93% of drill core sampled (32,264 m)					
Drilling Per Target / Zone								
Boomerang - Zone 3 (Mining Lease)	93	21,969.9	Drill meterage includes one (1) drill hole extension (165.5 m)					
Boomerang – Zone 3 (Buffer Zone)	27	4,552.6						
Twin Zone (Zone 3)	24	3,529.8	Drill meterage includes two (2) drill hole extensions totalling 56.5 m					
Double 19 – SW (Zone 3)	14	2,445.0	Scout drilling to test 3D litho-structural modelling targets					
Cobra Creek (Zone 5)	12	1,690.8	Scout drilling					
Other	4	549.0	Grassroots target: auriferous iron formation float trains with coincidental IP/Res anomalies					

10.5.1 Mineral Resource Area

Most of the drilling, 144 drill holes totalling 30,052.3 m (87%), completed during the present July 2021 to January 2024 reporting period targeted the new Boomerang and Twin Zone (i.e., formerly JK East) resource bodies within Zone 3 of the mineral resource footprint area.

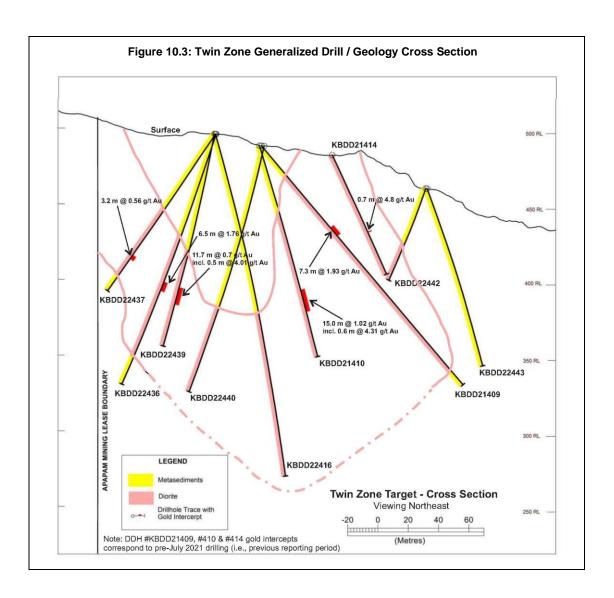
With present drilling efforts, 120 drill holes totalling 26,522.5 m (76%), primarily dedicated to the advancement of the Boomerang target to the mineral resource stage. The drilling was initially conducted along a series of NW-trending drill fences at nominal 50 m spacing, with the steeply inclined (-650 to -750) drill holes targeting the shallow SE-dipping mineralization package. The Boomerang gold system was subsequently tested by three SW-trending drill fences at approximately 90 m - 120 m spacing, with the moderately inclined drill holes (-500 - -600) designed to dissect the moderate NE-plunging, stacked, multi-shoot gold system, at the approximate mid-point between the existing NW-trending drill sections. With the central SW-trending drill fence (10 drill holes) covering an approximately 425 m down-plunge extension of the mineralization package (Figure 10.2).



<u>Cautionary Note:</u> A total of 27 drill holes totalling 4,552.6 m (BFDD22001 – BFDD23027) were collared within the "Buffer Zone" corresponding to the area between the current Apapam Mining Lease and Atewa Forest Reserve boundaries. This "Buffer Zone" area lies outside the current Apapam mining lease, and the Company presently does not hold title to this land. As stated in the Apapam mining lease status opinion prepared by REM Law Consultancy the renewal / extension of the Apapam mining lease by the Government of Ghana will in all probability be granted to cover the updated lease map that has been certified by the Technical Committee of the Minerals Commission as conforming to the Cadastral System. Consequently, upon issuance of the lease renewal / extension by the Government of Ghana, the lease area will most probably include the "Buffer Zone" area between the current Lease Boundary up to the edge of the Forest Reserve. Nonetheless, there is no assurance that the Company will ever be granted title to this land by the Government of Ghana.

The current drilling successfully amalgamated the former Boomerang East and West, and JK West targets into a structurally coherent gold complex (i.e., Boomerang Gold Complex). The gold system is characterized by a series of structurally parallel gold shoots or sub-deposits (i.e., JK1 – 11) occupying the north-western limb of an apparent, moderate NE-plunging, open (meso-scale) F2 synclinal hinge structure. With the gold mineralization characterized by tensional arrays of auriferous quartz-carbonate veins primarily hosted within or spatially associated with folded / strained diorite, and/or metasedimentary rock – diorite contacts. Present drilling established the stacked, multi-shoot Boomerang Gold System over approximately 650 m strike, 750 m down-plunge and 370 m cross-plunge distances, respectively.

A total of 24 drill holes totalling 3,529.8 m was completed on the Twin Zone mineral resource body during the present reporting period. The drilling was primarily conducted along three NW-SE trending drill fences (50 m spacing) targeting the down-plunge extension of the mineralized system. The current drilling successfully extended the gold mineralization over an approximately 330 m down-plunge distance. The Twin Zone auriferous body occupies an ENE-trending, moderately plunging, open to tight F2 synclinal hinge zone. With the gold mineralization characterized by diorite-hosted extensional (brittle) vein-arrays emplaced along the inner arc of the controlling synclinal fold structure. Significant drill results are presented in **Error! Reference s ource not found.** and a generalized cross-section of the drilling and geology for the Twin Zone auriferous body is depicted in Figure 10.3.



10.5.2Cobra Creek (Zone 5)

The 2021 – 2023 exploration program also included drilling campaigns on the Cobra Creek (Zone 5) and BIF Float targets located at the northeastern extremity of the Apapam concession (**Error! Reference source not found.**). These early-stage targets, exhibiting distinct litho-s tructural settings, fall outside the footprint area of the present Zone 1 – Zone 2 – Zone 3 mineral resource estimate.

A 12 drill holes (1,690.8 m) Phase II diamond core drill program was implemented from April to June 2022 on the Cobra Creek (Zone 5) target; an approximately 550 m wide, NE-trending, quartz-feldspar porphyry ("QFP") hosted, multi-structure braided shear zone system traced by trenching / outcrop stripping over an approximately 850 m strike length (Figure 10.4). With the drilling campaign including: eight (8) drill holes (773.6 m) designed to better target / dissect flatlying to shallow dipping gold-bearing extensional veining arrays and/or shallow plunging auriferous shoots; and 4 scout drill holes (917.2 m) targeting high-priority induced polarization

(IP) / resistivity anomalies along the southeastern margin of the QFP body. Significant drill results are presented in Table 10.6.

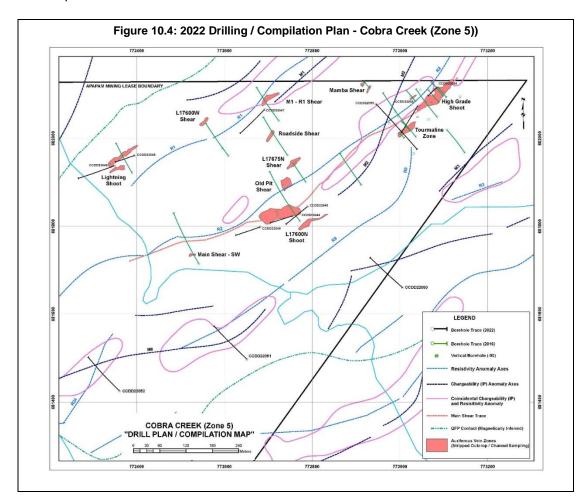


Table 10.6: Significant Drill Results – Cobra Creek (Zone 5) Target										
Drill hole ID	From (m)	From (m) To (m) Core-Length (m)								
	Cobra Creek Shear Zone System Drilling (Structural Targets)									
CCDD22044	18.0	28.4	10.4	2.00						
including	24.0	25.0	1.0	10.05						
CCDD22045	12.0	14.0	2.0	1.35						
CCDD22046	18.0	19.0	1.0	2.85						
CCDD22047	25.1	42.0	16.9	1.61						
including	40.0	41.0	1.0	12.92						
CCDD22047	57.0	59.0	2.0	4.63						
CCDD22047	94.0	99.0	5.0	1.91						
including	95.0	96.0	1.0	4.71						
CCDD22048	10.0	18.0	8.0	2.05						
including	12.0	13.0	1.0	6.5						
CCDD22049	0.0	20.0	20.0	0.48						
including	16.0	17.0	1.0	1.5						
CCDD22049	84.0	109.0	25.0	0.36						
including	84.0	89.0	5.0	0.65						
CCDD22053	76.0	80.0	4.0	0.79						
including	78.0	79.0	1.0	2.43						
CCDD22053	115.0	117.0	2.0	2.16						

Table 10.6: Significant Drill Results – Cobra Creek (Zone 5) Target								
Drill hole ID From (m) To (m) Core-Length (m) Au (g								
CCDD22054	24.0	28.0	4.0	4.44				
including	including 26.5 27.0 0.5		0.5	11.28				
Scout Drilling (Geophysical Targets)								
CCDD22052	81.0	84.0	3.0	1.13				

Drilling efforts also included 4 scout drill holes (917.2 m) on the BIF Float target, an approximately 700 m long auriferous banded iron formation ("BIF") rock float train spatially associated with a high chargeability (IP) / high resistivity geophysical trend, and partly coincidental gold-in-soil anomaly. None of the 4 scout drill holes targeting the BIF Float target returned any significant auriferous intercepts.

10.6 Drilling Quality

10.6.1 Diamond Drill Core

Diamond drill core is HQ diameter (63.5 mm) in upper oxidized material (regolith) and NQ2 diameter (50.6 mm) in the lower fresh rock portion of the drill hole.

All drill collar locations are surveyed-in by the company's in-house surveyor utilizing a combination of DGPS-established benchmarks and Total Station surveying; and completed drill holes are identified by a drill casing (PVC) secured by a cement base with an inscribed collar number.

Topographical control for the Zone 1 and Zone 2 drill collars, including the digital terrain model (DTM) employed for the Big Bend, East Dyke, Mushroom, South Ridge, and Gatehouse and Gold Mountain MRE bodies, is based on a satellite topography survey (i.e., photogrammetry) conducted in 2012 by PhotoSat of Vancouver, Canada (1 m bare earth survey grid). Topographical control for the Zone 3 drill collars, including the digital terrain model (DTM) employed for the Boomerang, Twin Zone and Double 19 MRE bodies, is based on a LiDAR drone topographical survey conducted in 2022 by Rockstats Geo Limited of Accra, Ghana (Absolute Accuracy: 10 cm horizontal & 4 cm vertical).

Directional surveys (azimuth/inclination) are conducted at 30 m downhole intervals as the drill hole is being drilled and at 6 m intervals during the pulling of the NQ rods upon completion of the drill hole (i.e., until reaching the HQ casing). The directional surveys are conducted utilizing a Reflex EZYTRAC multi-shot electronic survey tool with the instrument periodically serviced and calibrated by Imdex Global BV, the official Reflex Instrument distributor in Ghana. Approximately 50% of the NQ2 core (i.e., in fresh rock) is oriented utilizing the Reflex Act III core orientation tool.

10.6.2 Reverse Circulation (RC) Drill Samples

All Reverse Circulation (RC) drill hole collar locations were surveyed-in by a professional surveyor utilizing a combination of DGPS-established benchmarks and theodolite surveying;

and all drill casing (PVC) secured by a cement base. Directional surveys (azimuth/inclination) were conducted at 30 m downhole intervals utilizing an electronic single-shot drill hole survey tool.

11 SAMPLE PREPARATION, ANALYSES AND SECURITY

11.1 Sample Processing and Storage

Core is transported to and stored at the Kwabeng exploration camp. The core shack has the trained personnel and facilities to receive and layout core and RC samples in preparation for logging and sampling by the site geologist.

11.2 Sampling Methodology

11.2.1 Diamond Drill Core Samples

Drill core obtained from diamond drilling is targeted directly from the core tube into wooden core boxes, marked with the drill hole number and depth information. In the case of saprolite material the core is laid directly onto a strip of plastic wrap placed inside the box and then securely wrapped around the core to stabilize and prevent the dehydration of the saprolite. Core recovery and any drilling problems are noted by company staff at the drill site.

The NQ2 core (i.e., in fresh rock) is laid out along an angle iron on a work bench and meticulously re-assembled piece by piece with the core aligned with the orientation marks at the bottom of each 3 m drill run. The core orientation line (i.e., bottom of core) is marked along the length of the core with down-hole pointing arrows. The core is then measured, core recovery and rock quality designation (RQD) information collected, and a dashed sampling line perpendicular to the core orientation line marked along the length of the core. Each individual core box is photographed with the photographs taken from both dry and wet core.

A company geologist subsequently conducts geological logging of the core and marks the sample intervals. Structural logging is conducted with the Reflex IQ-Logger structural logging laser-device with the structural measurements digitally collected. The core is sampled over nominal 1 m intervals; with adjustments where necessary for mineralized structures. More comprehensive geological logging (on half core) is routinely conducted after the reception of assays on significant gold intercepts and petrographic studies periodically undertaken.

The diamond drill core is then saw-split lengthwise by trained company personnel, and half the core is immediately placed into a labelled plastic bag with a unique sample ticket stapled to the inside lip of the bag, and securely sealed by staples. The remaining half of core is returned to the core box and the box stored in a secure facility. The samples are then laid-out in sequence in the designated sample room to avoid duplications and omissions of samples in the laboratory submission orders, and the sample bags placed in labelled rice sacks in sequence. The shipping sacks are immediately secured with a numbered security seal (i.e., nylon zap strap) and stored in a locked room pending shipment to the laboratory.

11.2.2Reverse Circulation (RC) Drill Samples

Reverse circulation drill samples are collected immediately at the drill hole site under the supervision of a company geologist. The drill sample cuttings are collected in a cyclone over one (1) meter sample intervals; with the cyclone being purged after every 6 m drill run (i.e., hit

with sledge hammer and air blown). Reverse circulation drilling is conducted under dry ground conditions to ensure sample integrity. If water is encountered in the RC drill hole, the drill target is subsequently drilled by the diamond core method.

The dry RC bulk chip sample (\sim 25 to 30 kg) is then weighed and passed through a two – stage riffle splitter to produce a nominal 2 – 3 kg sample for assay which is also weighed on site. The splitter is thoroughly cleaned by hitting it with a wooden club and a wire brush is used after every sample. The split sample is immediately placed into a labelled plastic bag with a unique sample ticket stapled to the inside lip of the bag, and securely sealed by staples. The remaining portion of the bulk drill chip samples is then stored in large, labelled plastic bags at the drill site for future reference.

Drill cuttings from each sample interval are screened – washed and a quick log of the rock chips is completed at the drill site by a company geologist; noting amongst other things the sample quality/recovery, weathering profile, main lithologies, prominent alteration, and the character of the mineralization (i.e., oxide versus sulphide). Representative rock chips are also collected into a plastic sample tray for subsequent detailed geological logging of the drill hole by a senior geologist with the aid of a binocular microscope.

Upon transport to the Kwabeng exploration camp by company personnel at the end of the drill shift the samples are laid-out in sequence in a designated sample room to avoid duplications and omissions of samples in the laboratory submission orders, and the sample bags placed in labelled rice sacks in sequence. The shipping sacks are immediately secured with a numbered security seal (i.e., nylon zap strap) and stored in a locked room pending shipment to the laboratory. The bulk reference samples for economically significant gold intercepts are subsequently transported to the Kwabeng exploration camp for safe keeping.

11.2.3 Auger Sampling

Hand auger sampling is routinely utilized to test the geochemical signature of gold-in-soil anomalies at depth within the saprolite horizon to better define trenching targets. The augering is carried out with a locally fabricated cutting tool made from a used drill rod; the cylindrical cutting edge being driven into the ground to recover the sample. At auger sites where strong quartz scree is present, a collar pit (~ 0.5 m -1 m deep) is dug to facilitate penetration through the surficial quartz scree. Auger drill holes are typically sunk to a depth of 3 m – 5 m depending on the depth of the saprolite horizon. Sampling is typically conducted at 1 m intervals starting from the drill hole's 1 m mark. Auger drill hole spacing is typically at 25 m, with local 12.5 m infilling. To avoid any contamination only dry samples are collected.

11.3 Data Management

Xtra-Gold Resources utilizes the Datamine Fusion geological data management system for the collection, reporting and management of its geological data. The integrated software system allows the company to manage all drillhole/trench/channel, surface sampling, and Quality Control data from one location including managing reporting, analysis, and exporting data to GIS or modelling packages.

Fusion Server: Fusion Server provides for one central location for the administration and storage of all data (drillhole, samples, QA-QC, etc) in one central database.

DHLogger: Complete drillhole/trench data capture and management with embedded QC (Quality Control) Management. Fully customizable interface for all types of geological data collection and management including geological, geochemical, and geotechnical. parameters

Sample Station: Complete surface sample data capture and management with integrated QC. Stores all data for rock, soil, and stream samples. All data is validated on input and all sample data is stored and can be reported on and used in Mine Modelling and GIS systems.

QueryBuilder: A querying tool to extract data from the Fusion Database. Includes querying, graphing and reporting capabilities. The QC Charting Wizards plots for Standards, Duplicates, Lab Checks, Thompson Howarth, etc may be quickly generated.

Lab import: permits direct import of sample results from a commercial laboratory. Lab import checks the contents of the file for errors, validates sample numbers and directly imports samples, field QC and lab QC into the database. *Lab import* automatically performs QC checks and presents a control chart showing the QC for that analytical batch.

11.4 Sample Preparation

11.4.1SGS Laboratory Services

From August 2006 to August 2008 all samples were submitted to SGS Laboratory Services Ghana Ltd for analysis at their analytical facility located in Tarkwa, Ghana. The laboratory operates a Quality System, that accords to ISO 17025 standards. As part of the international group of SGS laboratories, the SGS Tarkwa laboratory takes part in a regular Round Robin sample analysis to check for bias or systematic error. Every sample batch is assayed alongside certified reference standards, a blank and a repeat. Aqua regia samples are done in batches of 20 and Fire assay samples in batches of 50. The Quality Control systems in place are such that analysis of blanks, standard reference materials, repeats and re-splits account for up to 25% of all determinations conducted. All such data is available to the company on final certificates if requested.

11.4.2ALS Chemex (ALS Ghana Limited)

From September 2008 to February 2017 sample preparation and analysis for all Xtra-Gold's samples, including diamond core and RC samples, trench channel, soil, and surface chip/grab samples were conducted by ALS Chemex (ALS Ghana Limited) at their analytical facility located in Kumasi, Ghana. ALS Chemex laboratory operations are covered by ISO 9001:2000 certification for the provision of assay and geochemical analytical services by QMI Quality Registrars and are accredited to ISO 17025 standards in various jurisdictions. The quality system and work procedures used in the Kumasi laboratory are identical to those used in the ALS Chemex laboratory in Vancouver, Canada and are subject to regular internal audits by their

global quality assurance team. The Kumasi laboratory also participates in a number of proficiency tests and round robins.

ALS Chemex's automated Laboratory Information Management System (LIMS) inserts quality control samples (reference materials, blanks and duplicates) on each analytical run, based on the rack sizes associated with the method. Quality control samples are inserted based on the following rack sizes specific to the method (Table 11.1).

	Table 11.1: Summary of Quality Control Protocols							
Rack Size	Methods	Quality Control Sample Allocation						
20	Specialty methods including specific gravity, bulk density, and acid insolubility	2 standards, 1 duplicate, 1 blank						
28	Specialty fire assay, assay- grade, umpire and concentrate methods	1 standard, 1 duplicate, 1 blank						
39	XRF methods	2 standards, 1 duplicate, 1 blank						
40	Regular AAS, ICP-AES and ICP-MS methods	2 standards, 1 duplicate, 1 blank						
84	Regular fire assay methods	2 standards, 3 duplicates, 1 blank						

ALS Chemex has developed the Open Lab™ system which allows clients on-line access to not just data reports generated by the laboratory, but to all the underlying QC data and audit trails. In addition, all lab QC data is automatically provided in the data files and analytical certificates posted/stored on their Webtrieve system or sent directly to the client.

11.4.3Intertek Minerals Limited

From March 2017, sample preparation and analysis for all Xtra-Gold's samples, including diamond drill programs, trench channel, soil, and surface chip/grab samples were and are conducted by Intertek Minerals Limited (Intertek) at their analytical facility located in Tarkwa, Ghana. The Tarkwa laboratory is accredited by the South African National Accreditation System (SANAS) (Accreditation No: T0796) in accordance with ISO17025 for Fire Assay, Aqua Regia and Carbon digestion.

As part of the Intertek's QC process, a full range of blanks, in house reference materials, certified reference materials, re-splits and checks are analysed with each job. Each job will contain approximately 4% reference materials, 4% re-splits, 4% checks and 2% blanks which can be control blanks (which monitor the digestion, analytical processes and instrument) or Prep blanks (which monitor contamination and analytical processes).

11.5 Chain of Custody

Reverse circulation drill samples are collected immediately at the drilling site under the supervision of a company geologist who escorts the samples back to the Xtra-Gold exploration camp in Kwabeng at the end of the shift. Drill core samples are saw-split and bagged under the supervision of a company geologist in a sample cutting room located adjacent to the core shack at the exploration camp. Trench, auger, soil, and surface rock chip/grab samples are transported from the field to the Kwabeng camp under Xtra-Gold's field team supervision.

All samples are laid-out in sequence in the designated sample room to avoid duplications and omissions of samples in the laboratory submission orders, and the sample bags placed in labelled rice sacks in sequence. The shipping sacks are immediately secured with a numbered security seal (i.e., nylon zap strap) and stored in a locked room pending shipment to the laboratory. Only the Project Geologist and the Chief Geotechnician have access to the key. The Kwabeng exploration camp is a fenced in compound with 24 hr security. A record of all samples shipped, as well as the actual samples within the individual sacks and their security seal numbers, is kept by the Project Geologist.

Depending on sample shipment size, samples are either transported by Xtra-Gold personnel directly to the analytical facility or scheduled for pickup by the laboratory from the Kwabeng exploration camp. A Tracking Record of all sample deliveries/pickups and pending analytical orders is kept by the Project Geologist, including personnel in custody of samples and time of departure; laboratory drop-off site; status of security seals; and assay turnaround time.

Upon delivery or pickup of samples a signed copy of the Sample Submittal Form is provided to Xtra-Gold personnel by the laboratory's sample reception staff. The laboratory has been instructed to notify Xtra-Gold's VP, Exploration and/or Senior Project Geologist immediately of any signs of tampering with the security seals or damage to the shipping sacks.

11.6 Analytical Procedure

11.6.1Sample Preparation and Analyses (SGS Laboratory Services – Pre September 2008)

Trench Channel, Hand Auger, and Rock Sample Samples

PRP86 Prep Method: Samples are dried in trays, crushed to a nominal 6 mm using a Jaw Crusher, then <1.5 kg is split using a Jones type riffle. Reject samples are retained in the original bag and stored. The split is pulverised in a chrome steel bowl to a nominal 75 μ m. An approximately 200 g sub-sample is taken for assay, with the pulverised residue retained in a plastic bag. All the preparation equipment is flushed with barren material prior to the commencement of the job.

FAA505 Analytical Method: Fuse a 50 g sample with a litharge-based flux, cupel, dissolve prill in aqua regia, extract into DIBK and determine Gold by flame AAS – Detection Limit 0.01 ppm.

Soil Samples

SCR30 Prep Method: Samples are dried (105° C) and disaggregated to break up lumps prior to sieving to -80 mesh; oversize material discarded.

FAE505 Analytical Method: Fuse a 50 g sample with a litharge-based flux, cupel, dissolve prill in aqua regia, extract into DIBK and determine Gold by flame AAS – Detection Limit 2 ppb.

11.6.2 Sample Preparation (ALS Chemex) (September 2008 – February 2017)

Prior to July 2009, drill core, trench channel, hand auger, and rock samples were prepared utilizing ALS Chemex's PREP-31 method as follows: The samples are logged in the tracking system, weighed, dried and finely crushed to better than 70% passing a 2 mm (Tyler 9 mesh, US Std. No.10) screen. A split of up to 250 g is taken utilizing a riffle splitter and pulverized to better than 85 % passing a 75 micron (Tyler 200 mesh, US Std. No. 200) screen.

From July 2009 onward, RC chip and drill core samples were prepared utilizing the PREP31-B method; in which a 1 kg split of the sample is pulverized to better than 85% passing 75 microns.

From October 2010 onward, including all the 2010 – 2012 drilling campaign sampling, drill core samples with observed visible gold and/or exhibiting typical Kibi-type granitoid hosted mineralization characterized by liberated, particulate gold grains were typically prepared utilizing ALS Chemex's PREP-22 method as follows; pulverize entire sample in multiple stages to 85% passing 75 microns or better. Recombine and homogenize by riffling and /or re-pulverize.

The soil samples were prepared using ALS Chemex's PREP-41 method as follows: samples are logged in the tracking system, weighed, low temperature dried and sieved to 180µmm (-80 mesh); both fractions are retained.

11.6.3 Analytical Method (ALS Chemex) (September 2008 - February 2017)

Prior to December 2011, all samples, including drill core and RC chips, trench channel, hand auger, rock, and soil samples, were typically analysed for gold only utilizing ALS Chemex's Au-AA24 method: Fire Assay Fusion with Atomic Absorption Spectroscopy (AAS) Finish. A prepared sample (50 g nominal sample weight) is fused with a mixture of lead oxide, sodium carbonate, borax, silica and other reagents as required, inquarted with 6 mg of gold-free silver and then cupelled to yield a precious metal bead. The bead is digested in 0.5 mL dilute nitric acid in the microwave oven. An amount of 0.5 mL concentrated hydrochloric acid is then added and the bead is further digested in the microwave at a lower power setting. The digested solution is cooled, diluted to a total volume of 4 mL with de-mineralized water, and analysed by atomic absorption spectroscopy against matrix-matched standards. Reporting range is 0.005 ppm – 10 ppm. Samples yielding over limit gold values (> 10 ppm) are re-assayed by Fire Assay (50 g nominal sample weight) with Gravimetric Finish (Au-GRA22).

Selective pulp duplicates submitted for re-assay, in addition to the standard Au-AA24 gold assay, were also submitted to a conventional exploration geochemistry ICP-AES analysis utilizing the ALS Chemex ME-ICP41 method (35 elements). Over limit values for base metal

and pathfinder elements of interest are subsequently re-analysed utilizing an Ore Grade ICP-AES method (OG46 method code).

From December 2011 onward, all samples except for soil samples, were analysed utilizing ALS Chemex's Au-AA26 method: Fire Assay Fusion (50 g aliquot) with Atomic Absorption Spectroscopy (AAS) Finish. Reporting range is 0.01ppm – 100 ppm.

From August 2011 onward, drill core samples with observed visible gold and/or exhibiting typical Kibi-type granitoid hosted mineralization characterized by liberated, particulate gold grains were typically analysed four (4) times either by the Au-AA24 (Au-GRA22 if >10 ppm) method or the Au-AA26 method; with the arithmetic average of the four assays reported. With the four fire assays conducted on four separate 250 g pulp subsamples (splits).

11.6.4Sample Preparation and Analytical Method (Intertek Minerals Limited) (Post March 2017)

Currently all assays are undertaken by Intertek Minerals Limited at their Tarkwa facility in Ghana. Batches of samples are typically collected by the laboratory from the Kwabeng field camp on a two-week schedule.

Drill core, trench channel, saw-cut channel, hand auger, and rock grab samples are typically prepared utilizing Intertek's SP12 prep-method as follows: samples are logged in the tracking system, weighed, dried, and crushed to nominal 10 mm screen size. A split of up to 1.2 kg is taken utilizing a riffle splitter and pulverized to better than 85 % passing a 75 micron. The milling specifications (grind size) are checked regularly.

All samples are analysed for gold utilizing method FA51/AA which is a lead collection fire assay (50 g aliquot) fusion method with an Atomic Absorption Spectroscopy finish (AAS) finish. Reporting range is 0.01 ppm – 800 ppm.

Drill core samples with observed visible gold and/or exhibiting typical Kibi-type granitoid hosted mineralization characterized by liberated, particulate gold grains are typically pulverized in their entirety to better than 85% passing 75 microns (SP13 prep-code) and analysed three (3) times by the FA51/AA method; with the arithmetic average of the three assays reported. With the three fire assays conducted on three separate 300 g pulp subsamples (splits).

Soil samples are dry-sieved to 180 microns (SV12 prep-code) and the undersize fraction analysed by 50-gram Trace Level fire assay fusion with atomic absorption spectroscopy finish (FA50/AA). Reporting range is 0.005 ppm – 10 ppm.

In March 2023, the analytical methods coding was changed to FA50/AA from FA51/AA, and to FA50L/AA from FA50/AA, by Intertek Minerals. The analytical methods remain the same as the original fire assay with an Atomic Absorption Spectroscopy finish methodology described above.

12 DATA VERIFICATION

12.1 Accurate Placement and Survey of Drill Hole Collars

Pre 2010: Drill collar locations were surveyed-in by a professional surveyor utilizing a combination of DGPS-established benchmarks and Total Station surveying.

Post 2010: Drill collar locations were surveyed-in by the company's in-house surveyor utilizing a combination of DGPS-established benchmarks and Total Station surveying.

Topographical control for the Zone 1 and Zone 2 drill collars, including the digital terrain model (DTM) employed for the Big Bend, East Dyke, Mushroom, South Ridge, and Gatehouse and Gold Mountain MRE bodies, is based on a satellite topography survey (i.e., photogrammetry) conducted in 2012 by PhotoSat of Vancouver, Canada (1 m bare earth survey grid). Topographical control for the Zone 3 drill collars, including the digital terrain model (DTM) employed for the Boomerang, Twin Zone and Double 19 MRE bodies, is based on a LiDAR drone topographical survey conducted in 2022 by Rockstats Geo Limited of Accra, Ghana (Absolute Accuracy: 10 cm horizontal & 4 cm vertical).

12.2 Downhole Surveys

Pre 2010: Downhole surveys (azimuth/inclination) were conducted at 30 m downhole intervals utilizing an electronic single-shot drill hole survey tool.

Post 2010: Downhole surveys (azimuth/inclination) were conducted at 30 m downhole intervals as the drill hole is being drilled and at 6 m - 9 m intervals during the pulling of the NQ rods upon completion of the drill hole (i.e., until reaching the HQ casing), utilizing an electronic multi-shot drill hole survey tool.

12.3 Analytical Quality Assurance and Quality Control Data

Quality-Control Programmes have been implemented to ensure best practice in the sampling and analysis of the diamond drill core, reverse circulation (RC) chip samples, saprolite trench and saw-cut channel samples, and soil samples and that the data can be used to inform subsequent work and the progression of the project.

12.4 Quality Assurance and Quality Control (QA/QC) Procedures and Results

12.4.12008 – 2010 Exploration Campaign

Analytical protocols utilized by Xtra-Gold involved the insertion of quality control samples into the sample stream of assay samples submitted to the laboratory. As of September 2008 certified reference standards, coarse analytical blanks, and field duplicate samples were inserted within sample sequences at the following rate: one (1) of each for every 20 samples within batches of Drill Core, RC Chip, and Trench Channel samples; and one (1) of each for every 40 samples within batches of Geological/Characteristic (i.e., grab, composite chip), Hand Auger, and Soil samples.

In February 2010, Xtra-Gold commissioned SEMS to conduct a detailed technical audit of Xtra-Gold's drill sample QA/QC program. Datasets assessed for quality control include:

Laboratory standards, blanks, duplicates and check repeats (Fire Assay and Gravimetric determinations). The audit reviewed the following:

- Client introduced standards, blanks and RC field duplicates (Fire Assay and Gravimetric determinations)
- Client resubmitted pulps
- Client quartered core
- SEMS quarter core control
- SEMS RC duplicates resubmitted to ALS Chemex and also Intertek, Tarkwa
- Check sieve test analysis
- Laboratory and field splitting error
- Results for Screen Fire assay versus 50 g Fire Assay

The following results were found from the investigation:

- Comparison of laboratory standards indicates that precision and accuracy are well within industry tolerance.
- Standards obtained from Canada indicate that some batches sent to ALS were poorly analysed. Those batches were re-analysed by ALS.
- Blanks obtained from Rocklabs indicate that there is no evidence from the results of laboratory or client blanks to suggest low-level contamination or repeated crosscontamination during crushing or pulverisation
- Laboratory duplicates and check repeats generally indicate that precision is well within industry tolerance.
- Xtra-Gold duplicates indicate that there is insignificant bias in the correlation between duplicates and original samples.
- Xtra-Gold quarter core submission to compare with original half core results highlighted the presence and effect of coarse gold grains on assays (nugget effect).
- RC chip resubmission indicated that there is possibly a problem with splitting of the sample in the field while resubmission of core indicates that there is also a problem with splitting in the laboratory.

12.4.2Umpire analysis for period 2008 – 2012

In 2021, 200 samples from the period 2008 – 2012 were reanalysed by Intertek.

The following results of the internal QA samples were found from the investigation:

- Although a relatively small population, no analyses of the blanks were greater than 3x the detection limit indicating that no significant contamination has occurred.
- The two internal certified reference materials (CRMs) didn't perform well. The one CRM performed within specification but the other failed.
- Typical of the deposit, a relatively large proportion of the duplicates have differences between the original and repeats.

The analysis of the umpire analysis indicates that there is a little bias between the datasets. However, there are a number of differences between the original analysis and the repeat analysis that are beyond generally acceptable limits. Having noted this and when looking at various other duplicate analyses utilising a number of other laboratories, it is clear that the

nature of the gold deportment within the deposit is the reason for the significant differences between the original and repeat analyses.

12.4.32010-2012 Exploration Campaign

Sampling carried out by Xtra-Gold was conscientiously and diligently pursued. The quality control programme involved inserting and analysing blanks standards, and duplicates, checking laboratory repeats to check precision and resampling core (quarter core). Overall the results of the QA/QC checks were reported as being very encouraging

The results were as follows:

- Contamination: All the QC blanks are below detection indicating that contamination is not significant.
- Accuracy: Some 30 CRMs were used by the laboratories. In all cases the analysis indicates that the laboratory's accuracy is within specification. Xtra-Gold used 11 CRMs in the QC programme. Generally the analysis of the CRMs indicates that the assay data can be considered accurate and suitable for mineral resource estimation.
- Precision: Various tests of the precision were undertaken. In general the analysis of the duplicates indicates that there is very little bias between the original and repeat analyses. However there are a number of individual differences between the original analysis and the repeat analysis that are beyond generally acceptable limits. Having noted this and when looking at various other duplicate analyses utilising a number of other laboratories, it is clear that the nature of the gold deportment within the deposit is the reason for the significant differences between the original and repeat analyses.
- Independent quarter core re-assay in mineralised zones: The independent assessment of mineralised intersections from the Big Bend, East Dyke, Mushroom, Double 19 and South Ridge prospects, yielded a precision of the original and resampling programmes that is within specification. The results suggest the presence of particulate/coarse grained gold.

12.4.42012 - 2021 Exploration Programme

Sampling carried out by Xtra-Gold was conscientiously and diligently pursued. The quality control programme involved inserting and analysing blanks standards, and duplicates, checking laboratory repeats to check precision and resampling core (quarter core). Overall the results of the QA/QC checks were reported as being very encouraging

The results were as follows:

- Blanks: Failed blanks are random, and all 2714 laboratory blanks are below detection indicating that contamination has been minimised and is statistically insignificant.
- Standards: On average, precision of standards is 14.3% which is slightly high, but bias is mostly below 5%, and shows both negative and positive sign. Four standards marginally exceeded 5%. Results for standards are acceptable although additional follow-up on failures is recommended in future work. The laboratory standards analysed yielded good precision, below 6.6%, and accuracy is generally very good and mainly less than +1% from the recommended value.
- Duplicates: Precision of field duplicates is high but consistent with the presence of particulate gold. After removal of "Flyers", regression analysis shows an absence of bias;

results are accurate. Precision of pulp duplicates is similarly high. Regression analysis shows a skewed dataset where bias is zero at 0.7 ppm Au increasing to 8% at 4 ppm gold. Evaluation of pulp duplicates, represented by four separate analyses for each sample, shows reasonable comparison in grade between the original and duplicated datasets. Average grade obtained using data only for pulp duplicates is 2.02 ppm +7.4%. This variability is acceptable with ore containing particulate gold. The laboratory duplicates yielded a precision of 9.7% for Fire Assay in the resource range =>0.5 ppm gold. Regression analysis indicates accurate results. Similar precision and accuracy was achieved with gravimetric and Screen Fire Assay determinations.

- Check assays: Laboratory check assays yielded a precision of 13.2% in the resource range =>0.5 ppm Au. Results are accurate.
- Independent quarter core re-assay in mineralised zones: The independent assessment of mineralised intersections from the Big Bend, East Dyke, Mushroom, Double 19 and South Ridge prospects, yielded high precision of the original and resampling programmes. However, the results indicate the presence of particulate gold. Regression analysis shows a weak bias where the re-assay values are slightly higher than the original assays. This is also observed comparing the mean grades of each dataset; mean original is 3.17g/t Au and re-assay gives 3.36g/t Au. The difference in the means is 6% and this is entirely satisfactory and underlines the accuracy of the database as a whole.

It was concluded that the geochemical data used in the mineral resource estimation was satisfactory, with variations most probably due to the "nuggetty gold effect".

12.4.52021 - 2023 Exploration Programme

Sampling carried out by Xtra-Gold, trenching and drilling, was conscientiously and diligently pursued utilising the Fusion - DHLogger database. The quality control programme involved inserting and analysing blanks standards, and duplicates, checking laboratory repeats to check precision and resampling core (quarter core). Overall the results of the QA/QC checks were reported as being satisfactory.

The results were as follows:

- Contamination: Blank samples are checked to ensure they do not return assay values greater than 3 x detection limit (0.03 ppm) which is the same as the threshold used by Intertek Minerals. Random failures occurred. Contamination is considered statistically insignificant
- Accuracy: Level 2 or complete failure was considered when the result was greater than the expected value for the designated CRM ± 3x standard deviation. Level 1 or part failure was considered when the result was greater than the expected value for the designated CRM ± 2x standard deviation. Where a failure is noted the pulps of samples surrounding the failed samples are re-assayed by Intertek. Re-assaying is typically conducted on Level 2 failures, with number of re-assayed pulps dependent on number of anomalous values in batch section. Re-assaying of Level 1 failures is also routinely conducted on batch sections exhibiting significant assay values.

A total of 654 pulp samples from 62 individual analytical batches were re-assayed during the current 2021 – 2023 period. The pulp re-assays are averaged with the original assays to produce a final assay value in the database.

The average value of the assays against the expected values of various CRMs, were within the required specification which does not indicate any significance bias. Based on the assessment completed the data is considered accurate.

■ **Duplicates:** Precision of field duplicates (¼ core) is poor but consistent with the presence of particulate gold. This variability is acceptable with material containing particulate gold. The Bias of the original vs the duplicates is 4.3%, which is an acceptable level – should be less than 5%.

Laboratory Quality Control Samples: The Laboratory CRM / Blank failures were also recorded.

Where drill core samples were observed with visible gold or exhibiting strong sulphide mineralization, the samples are typically pulverized in their entirety and subjected to three fire assay analyses, with the arithmetic average of the three assays being reported. The intention of conducting three assays on three separate pulp sub-splits, was to negate the "nugget" effect.

12.5 Conclusions and Recommendations

The following conclusions and recommendations are made:-

- The operator is diligent in the use of the QA/QC programme with the recording of data for analysis. An important aspect is that an effective and dynamic QC programme is utilised to review data as it comes in from the laboratory; a practice currently being applied on site.
- The assessment of the blanks confirms that there is minimal contamination at the laboratory.
- The assays were undertaken utilising a 50 g aliquot for the fire assay whereas the CRMs generally utilised a 30 g aliquot. The expectation is that the larger aliquot should produce results that are better grouped (precision) and more accurate.
- In the earlier programmes numerous failures of the Xtra-Gold CRMs have been noted. Most of these have been attributed to the misidentification of the CRMs. The laboratory CRMs demonstrate that the data can be considered to be accurate.
- The precision is tested by analysis of the duplicate data. The results of the duplicate analysis presented suggests that the precision is an issue. This is probably truer for the higher grades and may be related the presence of coarser grains of gold..

It was concluded that the geochemical data used in the resource estimation was satisfactory, with variations most probably due to the nature and deportment of the gold and probably related to the presence of coarse gold in the deposits.

13 MINERAL PROCESSING AND METALLURGICAL TESTING

A Gold Deportment Study which assessed the mineralogical and metallurgical aspects of the gold mineralization in the Kibi Gold Project was completed in October 2011 by SGS South Africa (Pty) Ltd. Two 10 kg samples of oxide material (average grade of 7.28 g/t Au) and sulphide material (average grade of 3.47 g/t Au) were analysed. The composite oxide sample was created from trench samples that were crushed and combined. The mineralogical test work included metallurgical and mineralogical tests that were undertaken in conjunction with the gravity test work. The tests performed included:

- Test work to determine the amenability of the material to gravity recovery;
- Gold distribution across size fractions (grading analysis);
- Heavy liquid separation to determine the amount of free gold or gold in heavy particles such as sulphides;
- Exposure and mineral association analysis of the particulate gold grains in the gravity concentrate;
- Chemical composition of the material and metallurgical test products;
- general mineralogical characterization of the material;
- Identification and quantification of gold minerals including native gold, gold-tellurides, etc.
 In the gravity concentrates;
- Grain size distribution of the gold grains in the gravity concentrate;
- Test work to determine the gold recovery by direct cyanidation; and
- Diagnostic leach analysis of the gravity tailings to determine the gold deportment in the gravity tails.

The preliminary conclusions made indicated that:

- The gold in the sulphide samples (3.49 g/t Au) was highly amenable to cyanidation leaching with ~97% recoverable by means of direct cyanidation. This material is also amenable to gravity upgrading, with ~67% of the gold recovered at a mass pull of ~3%. In the gravity concentrate (97.5 g/t Au), a total of 143 particulate gold grains were observed in the gravity concentrate of this sample.
- The grading analysis on the sulphide sample indicated a very high upgrading of gold in the +106µm size fraction (~69%). This indicates that the gold is either large gold grains or locked in large gold-bearing particles. From the liberation and mineral association characteristics determined by QEMSCAN, on the gravity concentrate, the gold was found to be ~63% liberated and ~25% was associated with pyrite. This indicates that the gold is either large, liberated gold grains or locked in large gold-bearing pyrite particles.
- The direct cyanidation and diagnostic leach indicates that the sample is highly amenable to cyanide leaching, with ~97% of the gold recovered from the head sample at a grind of 80%-75µm by direct cyanidation and ~96% for the gravity tailings at a grind of~50%-75µm. This is corroborated by the exposure and the mineral association characteristics as determined by QEMSCAN analysis of the gravity concentrate. Approximately 90% of the particulate gold grains are ≥10% exposed and should be leachable.
- The gold in the composite oxide sample (7.28 g/t Au) is also highly amenable to cyanidation, with ~97% of the gold recoverable by means of direct cyanidation. The material is also amenable to gravity upgrading, to some degree, with only ~56% of the gold recovered at a mass pull of ~3%. In the gravity concentrate (134.83 g/t Au) a total of 125 particulate gold grains were observed by QEMSCAN.
- The grading analysis on the composite oxide sample indicated a very high upgrading of gold in the +106µm size fraction (~74%). This indicates that the gold is either large gold grains or locked in large gold-bearing particles. From the liberation and mineral

- association characteristics determined by QEMSCAN analysis of the gravity tailings, it was found that the gold grains were moderately liberated (~76%) and that ~10% was occurring in silicates and ~14% in oxides. This indicates that the gold is either large, liberated gold grains or locked in large gold-bearing silicate/oxide particles.
- The direct cyanidation and diagnostic leach tests indicated that the sample is highly amenable to cyanide leaching, with ~98% of the gold recovered from the head sample at a grind of 80%-75μm and ~99% of the gold in the gravity tailings at a grind of 50%-75μm. This is corroborated by the exposure and mineral association characteristics of particulate gold in the gravity 28 concentrate, as determined by QEMSCAN analysis. Approximately ~96% of the gold grains are ≥10% exposed and should be leachable.
- The most simplistic processing option would be to mill the material to ~80%-75μm followed by carbon-in-leach cyanidation. Another option, which may result in somewhat lower operational cost is to mill the material relatively coarsely (say 80%-106μm) followed by gravity concentration and intensive cyanidation of the gravity concentrate. The gravity tailings could then be milled finer to ~80%-75 μm, followed by carbon-in- leach cyanidation. Taking out the coarse gold and some of the sulphides by gravity, will allow shorter retention times in the leach tanks and possibly even lower cyanide consumption.

14 MINERAL RESOURCE ESTIMATES

14.1 Methodology

Geological models for each target were generated based on the geology of the target and the identified structural trends. Block models were generated for Big Bend, Double 19, Road Cut, East Dyke (North Limb, Core, South Limb), South Ridge, Mushroom, GH and GM, Twin Zone and Boomerang. The Boomerang target was further split into the deeper (Boomerang) and shallower part (JK West). Each target was estimated separately form the data within the assigned geological model using variograms and search parameters that are aligned to the identified structural direction. The approach was based on the premise that mineralising fluids would have flowed through the identified structures.

The database is comprised of collar coordinates, downhole surveys, lithological core logs, bulk density data and assay data for the 642 drill holes and 417 trenches completed on the project.

Detailed descriptive statistical analyses of the data for each target were competed prior to starting the estimate in order to understand the data thoroughly. An assessment of the various data populations was made and capping values for each target were determined.

The estimations for each target were undertaken using ordinary kriging as lognormal kriging had produced results that could not be validated. Directional variograms were developed for each target utilising the direction of the veins within the target. This direction was also used as the search parameter for each estimate. The block size was considered after testing various parameters and with consideration to the geometry of the targets.

14.2 Geological Models

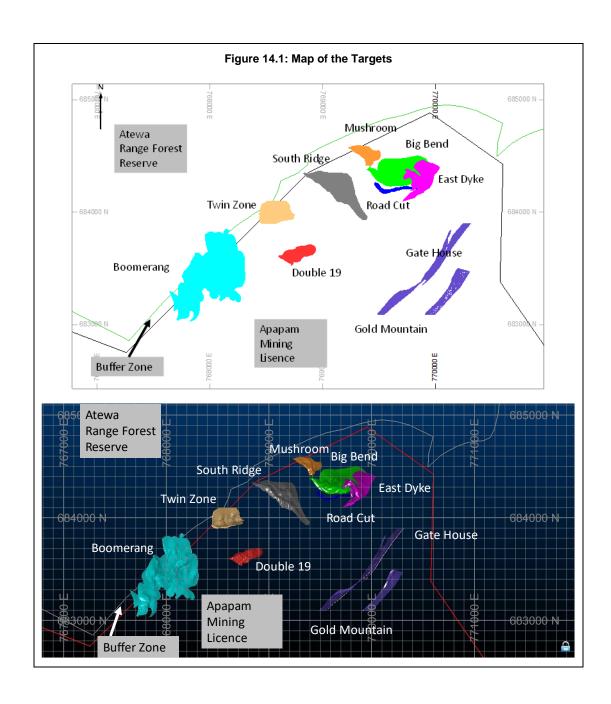
Three dimensional models of each target were created based on the known geology and structural interpretation. 3D Geological Modeling in Leapfrog Geo™ was performed and utilised the relevant interpretations described in Sections 7.4 - 7.6. The most recent structural and lithological data was integrated and incorporated into the 3D target models (including Big Bend, Double 19, East Dyke, Gate House, Gold Mountain, Mushroom, Road Cut and South Ridge, Twin Zone and Boomerang). These target models serve a dual purpose, as they are used as well-constrained mineralization envelopes for mineral resource estimation, and as approximations of structurally controlled mineralization zones that may be subject to further exploration targeting (Figure 14.1 and Figure 14.2).

Gold assays with a 0.4 g/t cut-off were used to constrain the outer limit of mineralization envelops, otherwise referred to as target models. No compositing of the assay data was applied prior to creating the envelopes. Each target model was informed by the structural setting of the unconstrained mineralized zone, as indicated by the geometry of combined lithological contacts and intersected gold mineralization.

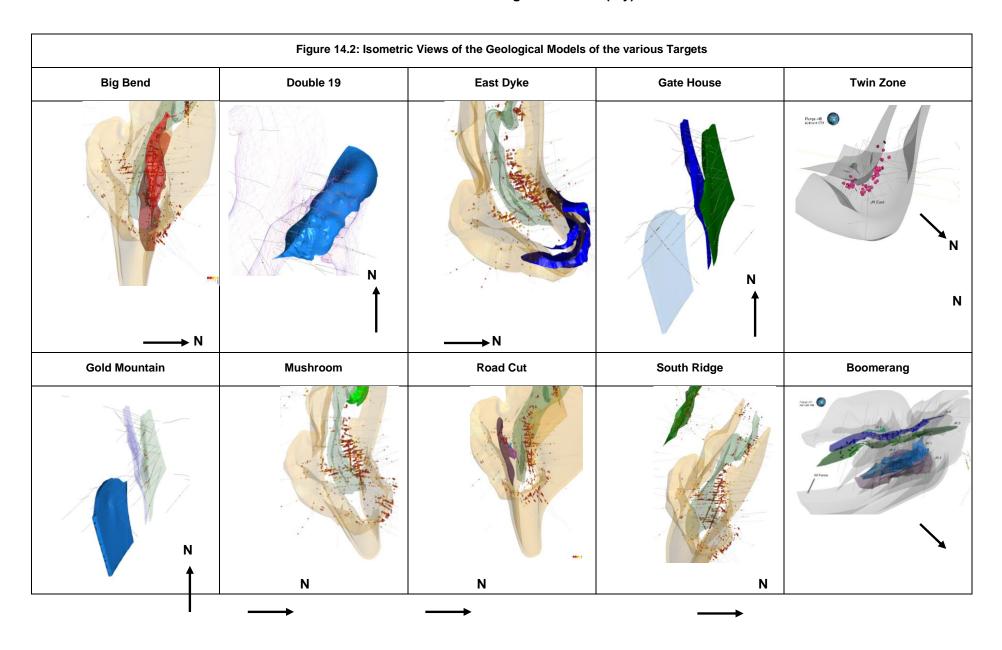
The structural setting of targets Big Bend, Double 19, East Dyke, Mushroom and Road Cut, in particular, are further enhanced by additionally modelled lithological contacts of undifferentiated mafic units (M1 and M2), diorite and metasediments.

Triangulation of target models in Leapfrog Geo[™] was undertaken using the Vein System tool in conjunction with manual interval selection of 0.4 g/t cut-off intervals. This methodology was preferred largely because of the level of confidence in the structural setting for each target vs. the pure implicit modelling (Intrusion tool) guided by LVA (locally variable anisotropy) or structural trends.

Using the deposit tool, a base surface (or contact) was modelled for a fresh rock-saprolite/laterite (TR-SAP-LAT) transitional zone and complete saprolite/laterite (SAP-LAT) zone, as recorded by drillhole and trench logs. Both TR-SAP-LAT and SAP-LAT surfaces approximate the level of oxidation and weathering from topography.



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14.3 Compositing

The data was composted to 1m intervals utilising the weighting by the drill hole length and density.

14.4 Descriptive Statistics: Composites

An intensive statistical analysis was undertaken on the data for each of the targets. The summary of analysis are presented in Table 14.14. and Figure 14.3. with comparisons of trench and drill hole data. It was determined that the drill hole and trench data represent the same geological environment and are compatible. It is noted that the trench data represents the oxide horizon.

The analysis of the data confirms that the gold grade populations of each of the targets is lognormal.

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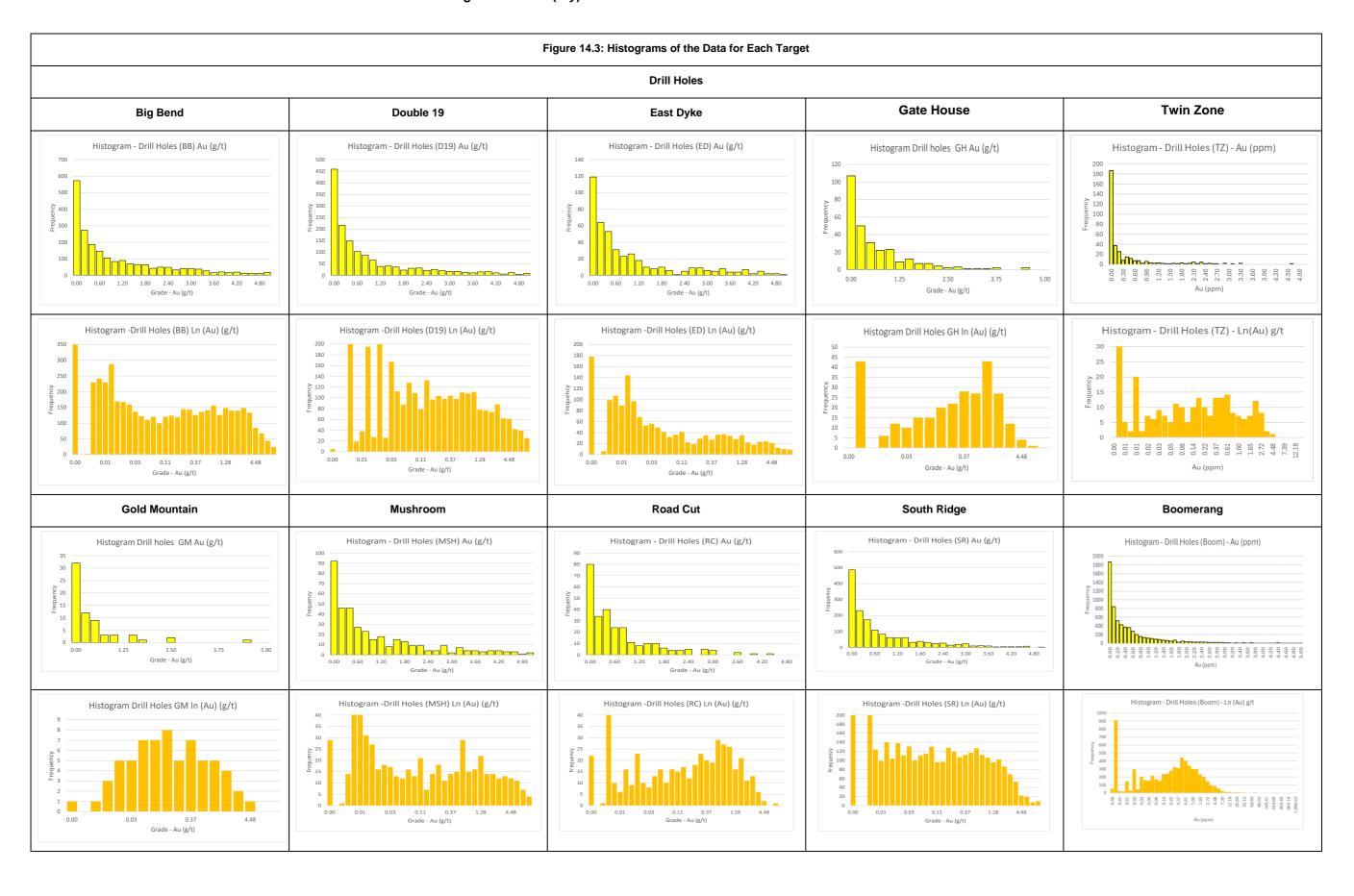
Table 14.14.: Descriptive Statistics per Target

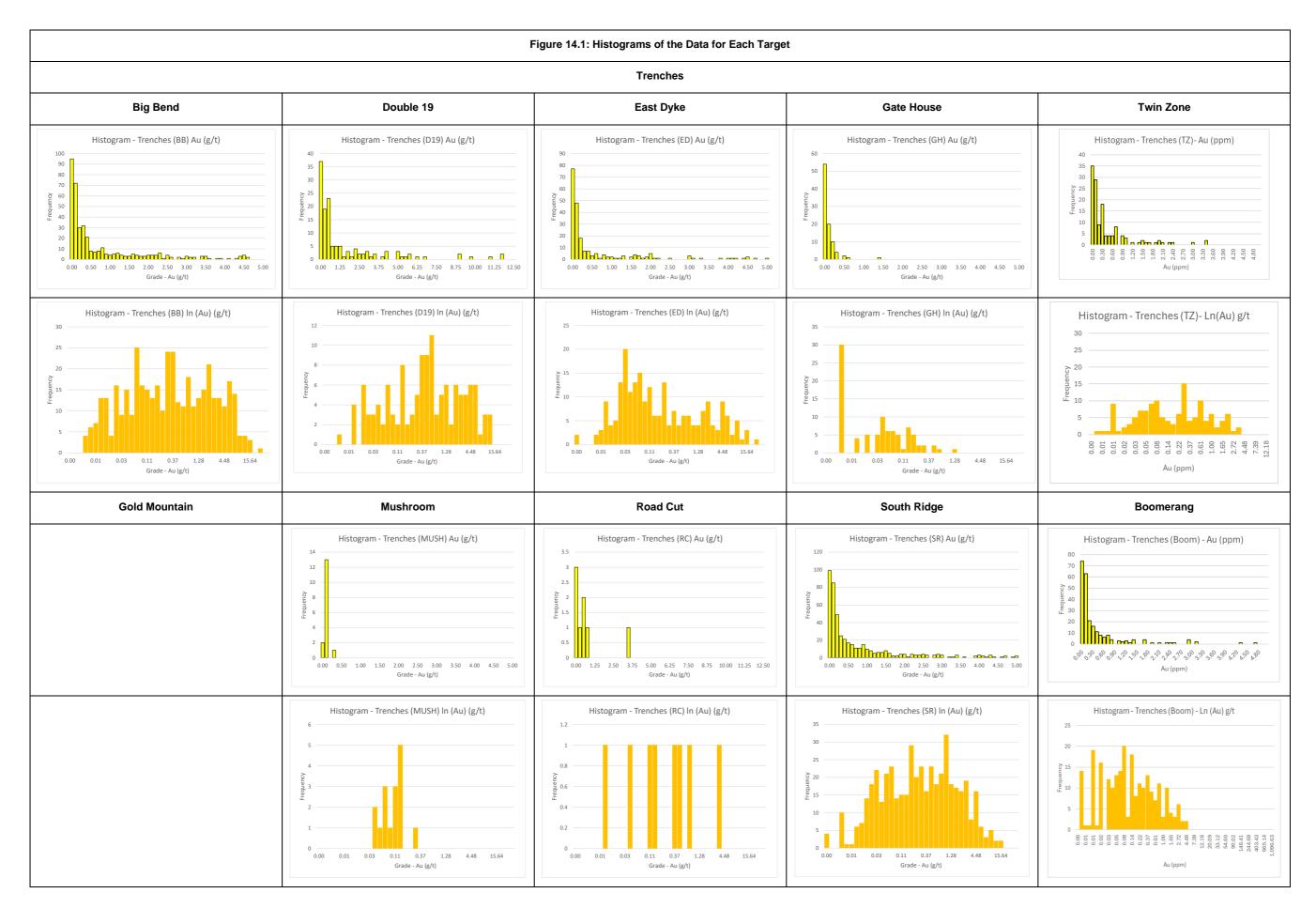
Drill	\Box	
Drill	по	ies

	Big Bend	Double 19	East Dyke	Gate House	Gold Mountain	Mush room	Road Cut	South Ridge	Twin Zone	Boomerang
Count	4858	3660	1556	285	66	905	576	3778	347	6520
Average	0.83	0.81	0.68	0.63	0.44	1.16	0.44	0.52	0.35	0.72
Min	0.00	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.01	0.00
Max	45.50	114.34	42.40	5.76	4.56	272.00	7.40	40.80	4.59	66.27
Range	45.50	114.34	42.40	5.76	4.56	272.00	7.40	40.80	4.58	66.27
Std Dev	2.06	2.86	2.37	0.86	0.76	9.34	0.78	1.64	0.69	1.78
Median	0.07	0.06	0.02	0.27	0.13	0.07	0.08	0.05	0.03	0.25
Mode	0.00	0.01	0.00	0.01	0.02	0.00	0.01	0.01	0.01	0.01
Geomean	0.28	0.24	0.14	0.47	0.32	0.29	0.22	0.19	0.14	0.44
CoV	249%	351%	349%	138%	172%	804%	176%	316%	196%	247%

Trenches

	Big Bend	Double 19	East Dyke	Gate House	Gold Mountain	Mush room	Road Cut	South Ridge	Twin Zone	Boomerang
Count	429	132	232	92		16	8	478	133	241
Average	1.63	1.59	1.36	0.09		0.10	0.66	1.04	0.45	0.40
Min	0.01	0.01	0.00	0.01		0.04	0.01	0.00	0.01	0.01
Max	45.30	11.80	29.10	1.38		0.27	3.44	17.60	3.44	4.72
Range	45.30	11.80	29.10	1.38		0.23	3.43	17.60	3.44	4.72
Std Dev	3.52	2.51	3.23	0.18		0.06	1.15	2.01	0.68	0.73
Median	0.31	0.49	0.11	0.04		0.11	0.27	0.28	0.18	0.09
Mode	0.01	0.02	0.03	0.01		0.07	#N/A	0.01	0.01	0.01
Geomean	0.30	0.42	0.19	0.03		0.09	0.20	0.26	0.35	0.26%
CoV	216%	158%	237%	200%		53%	173%	193%	151%	182



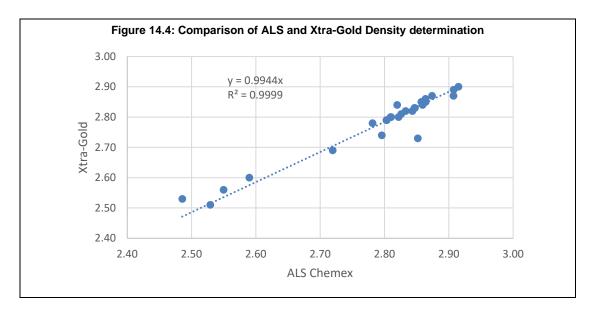


14.5 Density

Density determinations were undertaken by ALS and by Xtra-Gold.

The density was measured by pycnometer (Method: OA-GRA08). In addition, density determinations were undertaken on site using the Archimedes method. From the available information, cognisance of the porosity was noted with samples being immersed in wax prior to weighting in water.

A full analysis of both datasets was undertaken to confirm that the data was equivalent and that all the data could be used to determine the densities of all the blocks in the block model. Figure 14.4 shows a strong correlation coefficient and a very low bias between data sets. As a result, the two datasets were considered compatible and were used to determine the density of the mineralised material. A summary of the determinations is presented in Table 14.114.2.



Using all the data, densities were estimated independently of the gold and merged into the block model.

Table 14.114.2: Statistics of ALS and Xtra-Gold Density Determinations									
	Count	Average	Average Minimum		Standard Deviation				
		Big Ben	d	1					
Saprolite	22	1.56	1.37	2.34	0.25				
Transition	12	2.34	2.17	2.53	0.11				
Fresh	927	2.84	2.60	3.43	0.06				
		Double 1	9	1					
Saprolite	13	1.82	1.44	2.15	0.22				
Transition	8	2.34	2.03	2.64	0.09				
Fresh	15	2.90	2.76	3.08	0.09				
		East Dyk	ке	1					
Saprolite	12	1.58	1.41	2.06	0.19				
Transition	2	2.55	2.49	2.61	0.08				
Fresh	87	2.86	2.66	3.11	0.08				
	- 1	Mushroo	m	1					
Saprolite	13	1.68	1.26	2.40	0.30				
Transition	5	2.46	2.14	2.59	0.22				
Fresh	159	2.84	2.57	3.11	0.09				
	- 1	South Rid	lge	1					
Saprolite	15	1.53	1.29	1.77	0.16				
Transition	13	2.39	2.25	2.57	0.07				
Fresh	1451	2.83	2.55	3.83	0.08				
		Boomerar	ng						
Saprolite	23	1.61	1.12	1.90	0.18				
Transition	26	2.23	1.73	2.76	0.27				
Fresh	90	2.86	2.71	3.06	0.08				

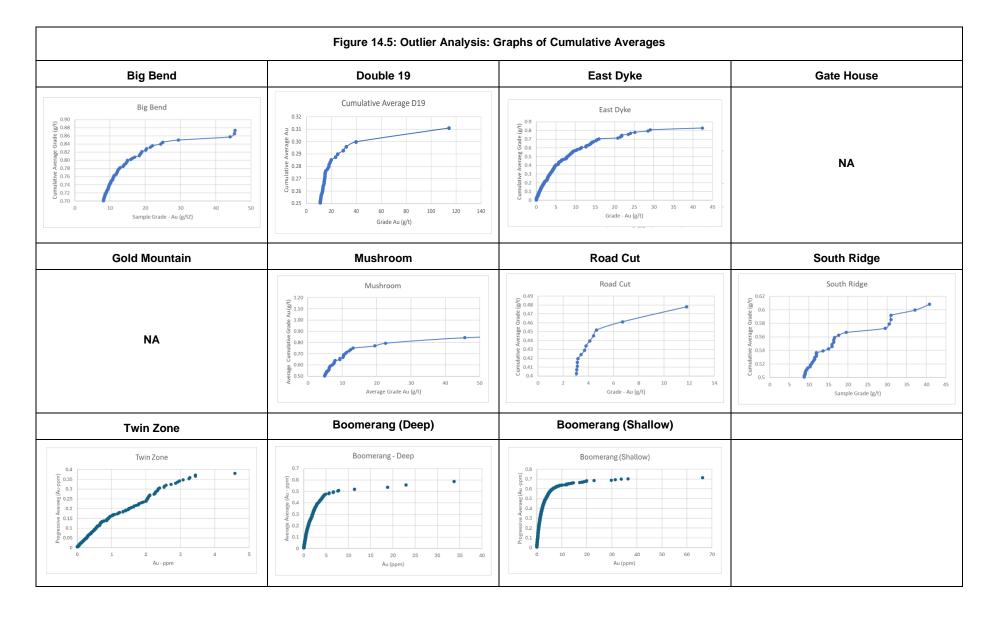
14.6 Outlier Analysis

The data was examined to determine if any of the data would be considered as an outlier that may have a negative effect on the estimation. An assessment of the high-grade samples was completed for each target to determine the requirement for possible high-grade cutting or capping. The approach is summarised as:

- Detailed review of histograms and probability plots with significant breaks in populations interpreted as possible outliers.
- Investigation of clustering of the higher-grade data.
- High-grade data which clustered was considered to be real while high grade samples not clustered with other high-grade data was considered to be a possible outlier that required further consideration either through cutting and/or search restriction.

Based on the analysis of the data set, capping was considered appropriate. The capping value was established for each target independently. The results of the outlier analysis are summarised in Table 14.3 and the cumulative averages graphically represented in Figure 14.5.

	Table 14.3: Ou	tlier Analysis	
Target	Capping value - Au (g/t)	Target	Capping value - Au (g/t)
Big Bend	20	Gold Mountain	None
Double 19	25	Mushroom	20
East Dyke	15	Road Cut	None
Gate House	None	South Ridge	15
Twin Zone	None	Boomerang (Deep)	15
Boomerang (Shallow)	19.75		



14.7 Block Model Development

A three dimensional (3D) model was developed. The block model cell size of 5 m \times 5 m \times 5 m was established after considering the Kriging Neighbourhood Analysis (KNA) (Table 14.4).

Centroid Based	Big Bend	East Dyke	Double 19	South Ridge	Road Cut
		Block Model Ori	igin (Centroid)		
Χ	769,300	769,700	768,500	768,800	769,40
Υ	684,200	684,000	683,500	684,000	684,10
Z	-200	-100	200	200	20
		Block Model M	ax (Centroid)		
Х	770,000	770,100	768,900	769,300	769,90
Υ	684,500	684,500	683,800	684,400	684,30
Z	600	600	500	600	50
		Block Model	- No of Cells		
Х	140	80	80	100	10
Υ	60	100	60	80	4
Z	160	140	60	80	6
		Block Model Pa	rent Cell Size	_	
Х	5	5	5	5	5
Υ	5	5	5	5	5
Z	5	5	5	5	5
		Sub cell s	splitting	1	
	No	No	No	No	No
Centroid Based	Mushroom	GM/GH	Twin Zone	Boomerang (Deep)	Boomerang (Shallow)
		Block Model Ori	igin (Centroid)		
Χ	769,200	769,500	767,000	767,000	767,70
Υ	684,400	683,000	683,000	683,000	683,00
Z	300	100	-10	-10	
	<u> </u>	Block Model M	ax (Centroid)	<u>"</u>	
Х	769,600	770,400	768900	768900	76880
Υ	684,600	683,800	684100	684100	68410
Z	600	500	590	590	57
		Block Model	- No of Cells	<u>"</u>	
Х	80	180	380	380	22
Υ	40	160	220	220	22
Z	60	80	120	120	10
		Block Model Pa			
Х	5	5	5	5	5
	5	5	5	5	5
Υ) J	0			
Υ	5	5	5	5	5
			5	5	5

14.8 Search Criteria

The search strategy used three search criteria and is summarised in Table 14.5. The shape and orientation of the search ellipse was informed by the results of the structural analysis (Table 14.5 and Figure 14.7).

A three-pass estimation strategy was applied to each domain, applying progressively expanded and less restrictive sample searches to successive estimation passes, and only considering blocks not previously assigned an estimate. The search criteria are designed to be informed from adjacent drill holes rather than be informed from data in the same drill hole.

							Table	14.5: Sı	ummary	of Sea	ch Para	ameters	;							
	Orientation		Searc Distan	се		rch Dist		Sam	ples		ch Dista		San	nples		rch Dista		Sa	mples	Max per BH
		Х	Υ	Z	Х	Min.	Max.	Min.	Max.	х	Υ	Z	Min.	Max.	Х	Υ	Z	Min.	Max.	
Big Bend	300/45	45	0	-60	50	50	25	12	24	100	100	50	12	24	250	250	125	12	24	
Double 19	060/45	45	0	60	50	10	60	12	24	100	20	120	4	12	250	50	300	4	12	
East Dyke				•	•	•	•		•	•	•	•		•	•		•	•		
CORE	060/30	0	30	60	20	250	250	12	24	40	500	500	12	24	100	1250	1250	12	24	
North limb	270/15	0	15	270	200	500	75	12	24	400	1000	150	12	24	1000	2500	375	12	24	
South limb	180/15	0	15	180	500	500	50	12	24	1000	1000	100	12	24	2500	2500	250	12	24	
Gate House																				
Gold Mountain																				
Mushroom	210/45	0	90	-45	50	50	25	12	24	100	100	50	12	24	250	250	125	12	24	
Road Cut	300/45	0	0	0	50	50	50	12	24	100	100	100	12	24	250	250	250	12	24	
South Ridge	090/15	0	0	0	50	50	50	12	24	100	100	100	12	24	250	250	250	12	24	
Twin Zone																				5
Boomerang (Deep)	Dynami	c Anis	otropy	•	50	50	10	20	30	75	75	15	20	30	100	100	20	20	30	5
Boomerang (Shallow)	Dynami	c Anis	otropy		50	50	10	20	30	75	75	15	20	30	100	100	20	20	30	5

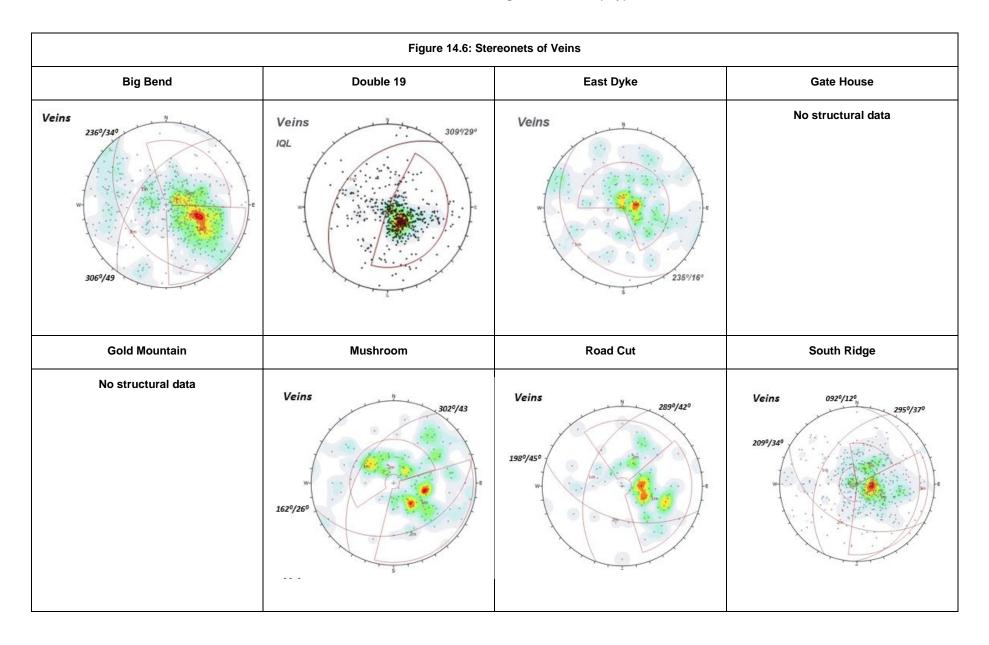


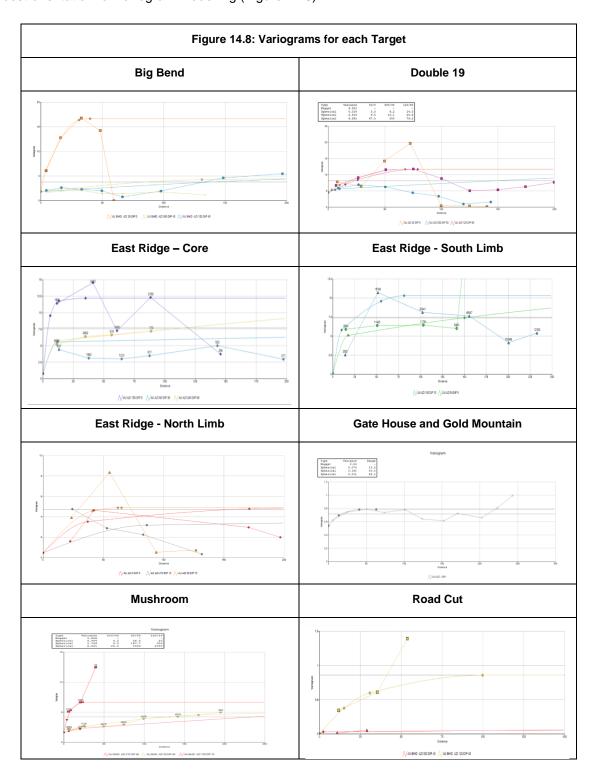
	Figure 14.6: Ste	reonets of Veins	
Twin Zone	Boomerang (Deep)		
Fold			
	Stereonet of two subhorizontal vein sets within the three modelled Boomerang orebodies	Stereonet of the updated three Boomerang Au grade shells. Orebodies are folded but broadly trend N-S	

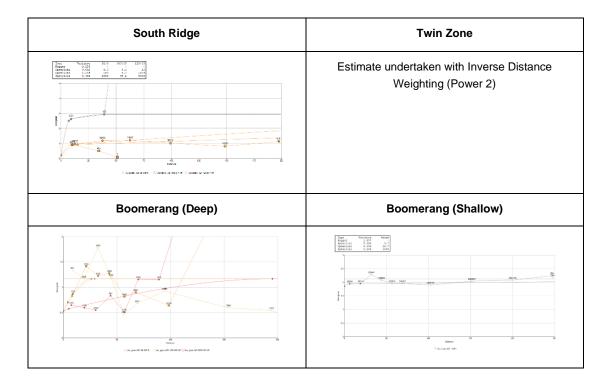
	Figure 14.7: Isometric views of the Geo	logical Models and the Vein Directions	
Big Bend	Double 19	East Dyke- Core	East Dyke- South Limb
236*/49*	300/30	60°/30°	180°/15°
Gate House & Gold Mountain	Mushroom	Road Cut	South Ridge
No structural data	302°/43°	289° / 42°	209*/34*

	Figure 14.7: Isometric views of the Geological Models and the Vein Direction	ns
Twin Zone	Boomerang (Deep)	Boomerang (Shallow)
Fold	763750 E 767500 E	

14.9 Variography

Variographic modelling was undertaken separately for each target. Anisotropic Variogram models were developed for the vein orientation. In general this direction presented a good or best orientation for variogram modelling (Figure 14.8).





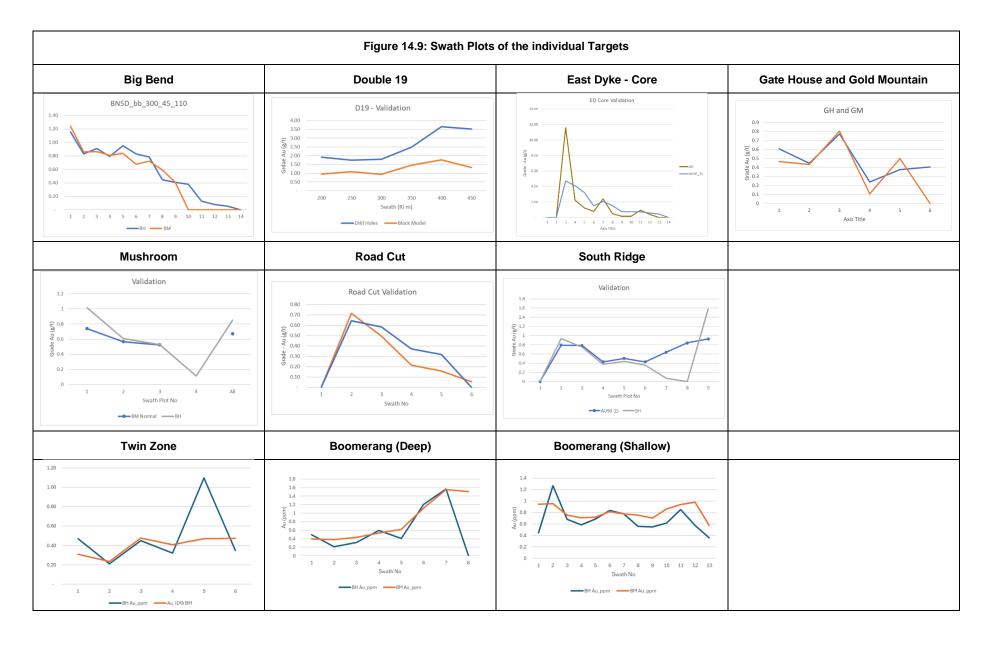
14.10 Estimation

As the populations for all the targets are lognormal, lognormal kriging was initially tested. However, the result was evidently incorrect, as in general very high grades were estimated. On further investigation it was established to be as a result of the high variance in the data which had a negative influence on the back transformation from the log values to the real values. As a result Ordinary kriging was utilised as being a more appropriate approach. Where a variogram could not be modelled, inverse distance weighting (power 2) was used for the estimation.

The estimation of each target was undertaken independently utilizing the structural information available. The premise used is that the main vein direction was the pathway for the mineralising fluid with their orientation being oblique to the lithological model.

14.11 Validation

Swath plots were undertaken for each target to demonstrate the veracity of the estimates (Figure 14.9). This analysis confirms the veracity of the estimates.



14.12Reasonable Prospects for Eventual Extraction

Consideration of the Reasonable Prospects for Eventual Economic Extraction (RPEEE) was undertaken using a simple financial assessment, assuming initially open pit extraction transitioning in some targets to an underground mining operation. An appropriate gold processing facility was assumed.

This simple financial model is presented, with the included assumptions, in Table 14.6. Based on these assumptions, there is potential for an open pit operation initially and subsequently an underground mine. Therefore a Mineral Resource can be declared.

Table 14.6:	Assessment of t	he Reasonable	Prospe	ects for Event	ual Economic Ex	traction
	Assumptions				Financial	
	Open Cast	Underground			Open Cast	Underground
	Au	Au			\$ (annual)	\$ (annual)
In Situ Grade	1.24	1.24	g/t	Revenue	\$26,737,947	\$26,737,947
Tonnage	2,880,000	2,880,000	t			
Mining losses - dilution etc	10%	10%		Capex	\$ 40,000,000	\$5,000,000
Mined -grade	1.12	1.12	g/t			
Price	\$2,300	\$2,300	\$/oz	Operating Co	sts (Opex)	
					\$ (annual)	\$ (annual)
Production				G&A	\$3,600,000	\$8,100,000
	Monthly	Monthly		Mining	\$5,400,000	\$10,800,000
Tonnage	30,000	30,000		Processing	\$9,000,000	\$7,200,000
oz	1,076	1,076		Total	\$18,000,000	\$26,100,000
	Annual	Annual				
Tonnage	360,000	360,000		Gross Profit	\$8,737,947	\$637,947
OZ	12,917	12,917			7 years	13 years
Recovery	90%	90%		Per tonne	\$/t	\$/t
				Revenue	\$74.27	\$74.27
Cost per tonne	at 360000 tpa			Cost	\$ 50.00	\$72.50
Cash on Mine	\$/t	\$/t		Value recovered	\$24.27	\$1.77
G&A	\$10.00	\$22.50		Per oz	\$/oz	\$/oz
Mining	\$15.00	\$30.00		Revenue	\$ 2,070.00	\$ 2,070.00
Processing	\$25.00	\$20.00	_	Opex	\$ 1,393.53	\$ 2,020.61
Total	\$50.00	\$72.50		Profit	\$ 676.47	\$49.39

Based on this assessment a cut-off grade of 0.5 g/t Au was considered appropriate.

14.13 Classification

The classification of each target was undertaken on an individual basis and broadly based on the efficiency of the search criteria i.e., where a block could be informed by the first search strategy, it was considered as Indicated and the other blocks were considered as Inferred. The Mineral Resource estimate was constrained by the expected depth at which the mine would potentially become uneconomic.

14.14 Mineral Resource Reporting

The Mineral Resource is declared in terms of the guidelines of the Canadian Institute of Mine (CIM) Standards (Table 14.8 and Table 14.7). A cut-off grade of 0.5 g/t was applied after consideration of the reasonable expectation of eventual economic extraction (Section 14.12).

	Table 14.7: Summary	of the Mineral Reso	ource Declaration	
	Lice	ense and Buffer Area		
	Tonnage (t)	Density (t/m³)	Grade - Au (g/t)	Au (oz)
Measured				
Indicated	27,532,000	2.79	1.20	1,058,200
M+I	27,532,000	2.79	1.20	1,058,200
Inferred	5,694,000	2.80	0.99	180,700
	Buffer Zone betv	veen License and Fo	rest Reserve	
	Tonnage (t)	Density (t/m³)	Grade - Au (g/t)	Au (oz)
Measured				
Indicated	3,769,000	2.83	1.20	145,400
M+I	3,769,000	2.83	1.20	145,400
Inferred				
		License Area		
	Tonnage (t)	Density (t/m³)	Grade - Au (g/t)	Au (oz)
Measured				
Indicated	22,179,000	2.79	1.18	842,400
M+I	22,179,000	2.79	1.18	842,400
Inferred	5,694,000	2.80	0.99	180,700

<u>Cautionary Note:</u> The Mineral Resource Estimate calculation includes the area termed the "Buffer Zone", which encompasses the "open ground" between the existing Apapam mining lease boundary up to the Forest Reserve boundary. Xtra-Gold applied for this Buffer Zone in the original application of extension/renewal of the Apapam mining lease on June 17th, 2015, and also in the updated extension/renewal dated June 28th, 2019. These applications were approved by the Technical Committee of the Minerals Commission and certified to be in conformity with the official cadastral system introduced under Ghana Mineral Law. Xtra-Gold is currently awaiting formal approval of the extension/renewal. Refer to the Apapam mining lease

Status Opinion prepared by REM Law Consultancy (Appendix B) for further details on the lease renewal / extension, including the Buffer Zone area.

Although the Company has taken all legal steps to extend the lease with the addition of the Buffer Zone, there is no assurance that such approval will be granted. In the interim, under Ghanaian mining law, the existing lease continues to remain in full force and effect.

The respective target grade/tonnage curves are shown in Figure 14.10.

Table 14.8: Mineral Resource Declaration - September 2024

Declared in terms of the CIM Standards

Cut-off: Au 0.5 g/t

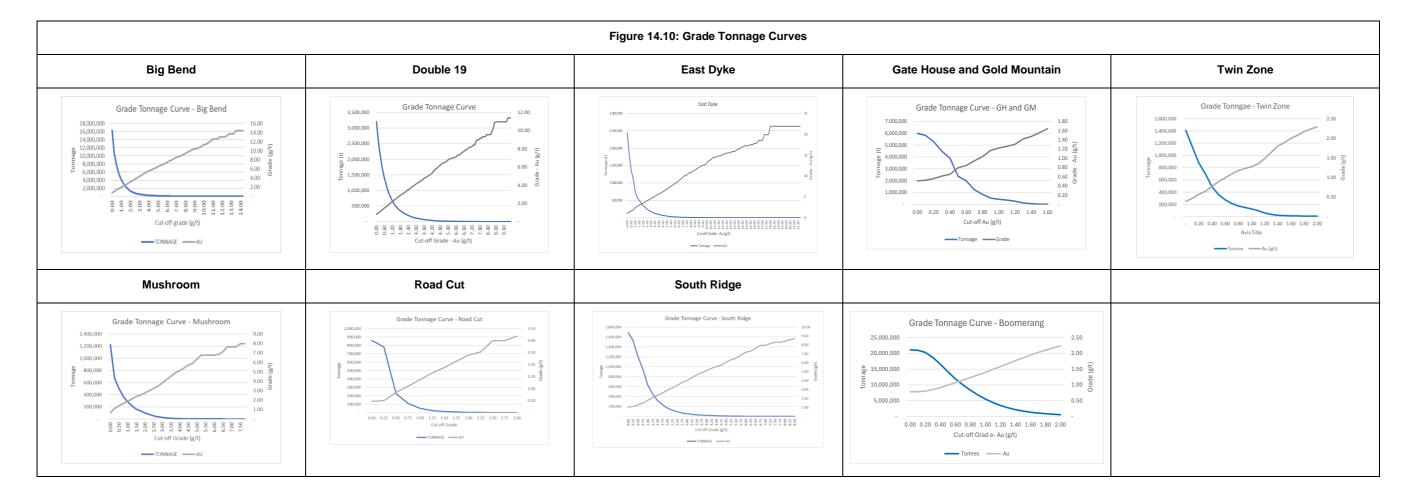
								Cut-off: Au 0	g								
		Licence Area and Bu	ffer Zone (B	Big Bend)				Ві	ıffer Zone((Big Bend)				Lic	ence Area (Big Bend)
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
Oxide	Indicated	284,000	2.00	2.06	18,800	Oxide	Indicated					Oxide	Indicated	284,000	2.00	2.06	18,800
ő	M+I	284,000	2.00	2.06	18,800	ő	M+I					ő	M+I	284,000	2.00	2.06	18,800
	Inferred	-	-	-	<u> </u>		Inferred						Inferred	-	-	-	-
چ	Measured				-	ءِ	Measured					Ę	Measured				-
sitio	Indicated	115,000	2.41	2.13	7,900	sitio	Indicated					sition	Indicated	115,000	2.41	2.13	7,900
Transition	M+I	115,000	2.41	2.13	7,900	Transition	M+I					La La	M+I	115,000	2.41	2.13	7,900
	Inferred	-	-	-	<u> </u>		Inferred						Inferred	-	-	-	-
	Measured				-		Measured						Measured				-
Fresh	Indicated	6,072,000	2.82	1.44	280,700	Fresh	Indicated					Fresh	Indicated	6,072,000	2.82	1.44	280,700
Fre	M+I	6,072,000	2.82	1.44	280,700	F	M+I					E E	M+I	6,072,000	2.82	1.44	280,700
	Inferred	1,257,000	2.82	1.03	41,400		Inferred						Inferred	1,257,000	2.82	1.03	41,400
	Measured				-		Measured						Measured				-
Total	Indicated	6,472,000	2.78	1.48	307,400	Total	Indicated					Total	Indicated	6,472,000	2.78	1.48	307,400
욘	M+I	6,472,000	2.78	1.48	307,400	º	M+I					မိ	M+I	6,472,000	2.78	1.48	307,400
	Inferred	1,257,000	2.82	1.03	41,400		Inferred						Inferred	1,257,000	2.82	1.03	41,400
		Licence Area and Buf	fer Zone (E	ast Dyke)				Bu	ffer Zone(I	East Dyke))			Lice	ence Area (East Dyke	e)
		Licence Area and Buf	fer Zone (Ea Density (t/m3)	Grade (g/t)	Content (ozs)			Bu Tonnage (t)	ffer Zone(I Density (t/m3)	East Dyke) Grade (g/t)	Content (ozs)			Lice Tonnage (t)	ence Area (I Density (t/m3)	East Dyke Grade (g/t)	Content (ozs)
	Measured		Density	Grade			Measured		Density	Grade	Content		Measured				
de	Measured Indicated		Density	Grade		de	Measured Indicated		Density	Grade	Content	ge	Measured Indicated				
Oxide		Tonnage (t)	Density (t/m3)	Grade (g/t)	(ozs)	Oxide			Density	Grade	Content	Oxide		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
Oxide	Indicated	Tonnage (t) 269,000	Density (t/m3)	Grade (g/t)	(ozs)	Oxide	Indicated		Density	Grade	Content	Oxide	Indicated	Tonnage (t) 269,000	Density (t/m3)	Grade (g/t)	Content (ozs) 13,700
	Indicated M+I	Tonnage (t) 269,000 269,000	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58	13,700 13,700		Indicated M+I Inferred Measured		Density	Grade	Content		Indicated M+I	Tonnage (t) 269,000 269,000	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58	Content (ozs) 13,700
sition	Indicated M+I Inferred	Tonnage (t) 269,000 269,000	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58	13,700 13,700		Indicated M+I Inferred Measured		Density	Grade	Content		Indicated M+I Inferred	Tonnage (t) 269,000 269,000	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58	Content (ozs) 13,700
sition	Indicated M+I Inferred Measured	Tonnage (t) 269,000 269,000 -	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58 1.58	13,700 13,700		Indicated M+I Inferred Measured		Density	Grade	Content	sition	Indicated M+I Inferred Measured	Tonnage (t) 269,000 269,000 -	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58 1.58	13,700 13,700
	Indicated M+I Inferred Measured Indicated	Tonnage (t) 269,000 269,000 - 92,000	Density (t/m3) 2.17 2.17	Grade (g/t) 1.58 1.58 - 1.46	13,700 13,700 - - 4,300	Transition Oxide	Indicated M+I Inferred Measured		Density	Grade	Content		Indicated M+I Inferred Measured Indicated	Tonnage (t) 269,000 269,000 92,000	Density (t/m3) 2.17 2.17 - 2.45	Grade (g/t) 1.58 1.58	13,700 13,700 - - 4,300
sition	Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 269,000 269,000 - 92,000 92,000	Density (t/m3) 2.17 2.17	1.58 1.58 1.46	13,700 13,700 - - 4,300		Indicated M+I Inferred Measured Indicated M+I		Density	Grade	Content	sition	Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 269,000 269,000 - 92,000 92,000	Density (t/m3) 2.17 2.17 2.45 2.45	Grade (g/t) 1.58 1.58 - 1.46 1.46	13,700 13,700 - - 4,300
Transition	Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 92,000 92,000	Density (t/m3) 2.17 2.17	1.58 1.58 1.46	(ozs) 13,700 13,700 - 4,300 4,300	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred		Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 92,000 92,000	Density (t/m3) 2.17 2.17 2.45 2.45	Grade (g/t) 1.58 1.58 - 1.46 1.46	13,700 13,700 - - 4,300
sition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 269,000 269,000 - 92,000 92,000	Density (t/m3) 2.17 2.17 - 2.45 2.45	1.58 1.58 1.46 1.46	13,700 13,700 4,300 4,300		Indicated M+I Inferred Measured Indicated M+I Inferred Measured		Density	Grade	Content	sition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 269,000 269,000 - 92,000 92,000 - -	Density (t/m3) 2.17 2.17 - 2.45 2.45	Grade (g/t) 1.58 1.58 1.46 1.46	13,700 13,700 - - 4,300 4,300
Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 269,000 269,000 92,000 92,000 2,742,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48	(ozs) 13,700 13,700 - 4,300 4,300 - 130,900	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated		Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 269,000 269,000 92,000 92,000 - 2,742,000	Density (t/m3) 2.17 2.17 2.45 2.45 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48	13,700 13,700 4,300 4,300 - 130,900
Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 - 2.79 2.79	Grade (g/t) 1.58 1.58 - 1.46 1.46 - 1.48 1.48	13,700 13,700 4,300 4,300 130,900 130,900	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I		Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 - 2.79 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48 1.48	13,700 13,700 4,300 4,300 - 130,900
Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 92,000 92,000 - 2,742,000 2,742,000 1,128,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 - 2.79 2.79	Grade (g/t) 1.58 1.58 - 1.46 1.46 - 1.48 1.48	(ozs) 13,700 13,700 - 4,300 4,300 - 130,900 130,900 43,300	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred		Density	Grade	Content	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated	Tonnage (t) 269,000 269,000 92,000 92,000 - 2,742,000 2,742,000 1,128,000	Density (t/m3) 2.17 2.17 - 2.45 2.45 - 2.79 2.79	Grade (g/t) 1.58 1.58 1.46 1.46 1.48 1.48	13,700 13,700 4,300 4,300 - 130,900
Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000 1,128,000	Density (t/m3) 2.17 2.17 2.45 2.45 2.79 2.79 2.84	1.58 1.58 1.46 1.46 1.48 1.19	13,700 13,700 4,300 4,300 130,900 130,900 43,300	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured MHI Inferred Measured		Density	Grade	Content	sh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 269,000 269,000 - 92,000 92,000 - 2,742,000 2,742,000 1,128,000 -	Density (t/m3) 2.17 2.17 2.45 2.45 2.79 2.79 2.84	1.58 1.58 1.46 1.46 1.48 1.19	13,700 13,700 4,300 4,300 130,900 130,900 43,300

		Licence Area and Buf	fer Zone (M	ushroom)				But	ffer Zone(N	lushroom)			Lice	ence Area (N	/lushroom	n)
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
Oxide	Indicated	103,000	2.04	1.50	4,900	Oxide	Indicated					Oxide	Indicated	103,000	2.04	1.50	4,900
ő	M+I	103,000	2.04	1.50	4,900	ŏ	M+I					ő	M+I	103,000	2.04	1.50	4,900
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
<u> </u>	Measured	-			-	5	Measured					ڃ	Measured	-			-
sitic	Indicated	17,000	2.05	1.11	600	sitio	Indicated					sitic	Indicated	17,000	2.05	1.11	600
Transition	M+I	17,000	2.05	1.11	600	Transition	M+I					Transition	M+I	17,000	2.05	1.11	600
•	Inferred	-	-	-	<u>-</u>		Inferred					•	Inferred	-	-	=	-
	Measured	-			-		Measured						Measured	-			-
Fresh	Indicated	385,000	2.81	1.35	16,600	Fresh	Indicated	16,000	2.82	1.18	600	Fresh	Indicated	369,000	2.81	1.35	16,100
Œ	M+I	385,000	2.81	1.35	16,600	_ <u>_</u>	M+I	16,000	2.82	1.18	600	L.	M+I	369,000	2.81	1.35	16,100
	Inferred	-	-	-	-		Inferred	-	-	-	-		Inferred	-	-	-	-
	Measured	-	0.00	4.07	-		Measured	-	0.00	4.40	-		Measured	-	0.04	4.07	-
Total	Indicated M+I	505,000	2.63 2.63	1.37 1.37	22,200	Total	Indicated M+I	16,000	2.82 2.82	1.18 1.18	600	Total	Indicated M+I	489,000	2.64 2.64	1.37 1.37	21,600 21,600
_	Inferred	505,000	2.03	1.37	22,200		Inferred	16,000	2.02	1.10	600		Inferred	489,000	2.04	1.37	21,000
		Licence Area and Buffe	er Zone (So	uth Ridge)		·	_	Buff	ier Zone(Sc	outh Ridge	e)			Lice	nce Area (So	outh Rida	e)
		Licence Area and Buffe						Buff	fer Zone(So	Ū				Lice	nce Area (Se	_	
		Licence Area and Buffo	er Zone (So Density (t/m3)	uth Ridge) Grade (g/t)	Content (ozs)			Buff	fer Zone(So Density (t/m3)	Outh Ridge Grade (g/t)	Content (ozs)			Lice	Density (t/m3)	outh Ridg Grade (g/t)	e) Content (ozs)
	Measured		Density	Grade	Content		Measured		Density	Grade	Content		Measured		Density	Grade	Content
ide	,		Density	Grade	Content	ide	Measured Indicated		Density	Grade	Content	ide			Density	Grade	Content
Oxide	Measured	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)	Oxide			Density	Grade	Content	Oxide	Measured	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
Oxide	Measured Indicated	Tonnage (t) 423,000	Density (t/m3)	Grade (g/t)	Content (ozs)	Oxide	Indicated		Density	Grade	Content	Oxide	Measured Indicated	Tonnage (t) 423,000	Density (t/m3)	Grade (g/t)	Content (ozs) 15,600
	Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 -	Density (t/m3) 2.35 2.35	Grade (g/t) 1.15 1.15	Content (ozs) 15,600 15,600		Indicated M+I Inferred Measured		Density	Grade	Content		Measured Indicated M+I	Tonnage (t) 423,000 423,000 -	Density (t/m3) 2.35 2.35	Grade (g/t) 1.15 1.15	Content (ozs) 15,600 15,600
sition	Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 423,000 423,000 - 181,000	Density (t/m3) 2.35 2.35 - 2.68	Grade (g/t) 1.15 1.15 1.16	Content (ozs) 15,600 15,600	sition	Indicated M+I Inferred Measured Indicated		Density	Grade	Content	sition Oxide	Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 423,000 423,000 - 181,000	Density (t/m3) 2.35 2.35 - 2.68	Grade (g/t) 1.15 1.15 - 1.16	Content (ozs) 15,600 15,600 - 6,700
sition	Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 -	Density (t/m3) 2.35 2.35	Grade (g/t) 1.15 1.15	Content (ozs) 15,600 15,600	sition	Indicated M+I Inferred Measured Indicated M+I		Density	Grade	Content		Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 423,000 423,000 -	Density (t/m3) 2.35 2.35	Grade (g/t) 1.15 1.15	Content (ozs) 15,600 15,600
	Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 423,000 423,000 181,000 181,000	Density (t/m3) 2.35 2.35 - 2.68	Grade (g/t) 1.15 1.15 1.16	Content (ozs) 15,600 15,600		Indicated M+I Inferred Measured Indicated M+I Inferred		Density	Grade	Content	nsition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 423,000 423,000 181,000 181,000	Density (t/m3) 2.35 2.35 - 2.68	Grade (g/t) 1.15 1.15 - 1.16	Content (ozs) 15,600 15,600 - 6,700
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 - 181,000 181,000	Density (t/m3) 2.35 2.35 - 2.68 2.68	Grade (g/t) 1.15 1.15 1.16 1.16	Content (ozs) 15,600 15,600 - 6,700 6,700	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured		Density	Grade	Content	Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 181,000 181,000	Density (t/m3) 2.35 2.35 2.68 2.68	Grade (g/t) 1.15 1.15 1.16 1.16	Content (ozs) 15,600 15,600 - 6,700 6,700
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Tonnage (t) 423,000 423,000 181,000 1,402,000	Density (t/m3) 2.35 2.35 - 2.68 2.68 - 2.81	Grade (g/t) 1.15 1.15 1.16 1.16 1.03	Content (ozs) 15,600 15,600	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated		Density	Grade	Content	Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Tonnage (t) 423,000 423,000 181,000 181,000 - 1,402,000	Density (t/m3) 2.35 2.35 2.68 2.68 2.81	Grade (g/t) 1.15 1.15 1.16 1.16 1.03	Content (ozs) 15,600 15,600 6,700 6,700 - 46,400
sition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 423,000 423,000 181,000 181,000 - 1,402,000 1,402,000	Density (t/m3) 2.35 2.35 2.68 2.68 2.81 2.81	Grade (g/t) 1.15 1.15 1.16 1.16 1.03 1.03	Content (ozs) 15,600 15,600 - 6,700 6,700 - 46,400 46,400	sition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Measured Indicated M+I		Density	Grade	Content	nsition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 423,000 423,000 181,000 181,000 - 1,402,000 1,402,000	Density (t/m3) 2.35 2.35 - 2.68 2.68 - 2.81 2.81	Grade (g/t) 1.15 1.15 1.16 1.16 1.03 1.03	Content (ozs) 15,600 15,600 - 6,700 6,700 - 46,400 46,400
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated	Tonnage (t) 423,000 423,000 181,000 1,402,000 1,402,000 943,000	Density (t/m3) 2.35 2.35 - 2.68 2.68 - 2.81	Grade (g/t) 1.15 1.15 1.16 1.16 1.03	Content (ozs) 15,600 15,600 - 6,700 6,700 - 46,400 46,400 30,800	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated M+I Inferred		Density	Grade	Content	Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Measured Indicated M+I Inferred	Tonnage (t) 423,000 423,000 181,000 - 1,402,000 1,402,000 943,000	Density (t/m3) 2.35 2.35 2.68 2.68 2.81	Grade (g/t) 1.15 1.15 1.16 1.16 1.03	Content (ozs) 15,600 15,600 6,700 6,700 - 46,400
Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 181,000 181,000 - 1,402,000 1,402,000 943,000	2.35 2.35 2.68 2.68 2.81 2.81 2.82	Grade (g/t) 1.15 1.15 1.16 1.16 1.03 1.03 1.02	Content (ozs) 15,600 15,600 6,700 6,700 46,400 46,400 30,800	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred		Density	Grade	Content	Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 181,000 181,000 - 1,402,000 1,402,000 943,000 -	Density (t/m3) 2.35 2.35 2.68 2.68 2.81 2.81 2.82	Grade (g/t) 1.15 1.15 1.16 1.03 1.03 1.02	Content (ozs) 15,600 15,600 - 6,700 6,700 - 46,400 46,400 30,800
Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 423,000 423,000 181,000 181,000 1,402,000 1,402,000 943,000 - 2,005,000	2.35 2.35 2.68 2.68 2.81 2.81 2.82	Grade (g/t) 1.15 1.15 1.16 1.16 1.03 1.03 1.02	Content (ozs) 15,600 15,600	Fresh Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated Inferred Inferred		Density	Grade	Content	Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated M-I Inferred	Tonnage (t) 423,000 423,000 181,000 181,000 1,402,000 1,402,000 943,000 - 2,005,000	Density (t/m3) 2.35 2.35 2.68 2.68 2.81 2.81 2.82	Grade (g/t) 1.15 1.15 1.16 1.16 1.03 1.03 1.07	Content (ozs) 15,600 15,600 6,700 6,700 - 46,400 30,800 - 68,700
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 181,000 181,000 - 1,402,000 1,402,000 943,000	2.35 2.35 2.68 2.68 2.81 2.81 2.82	Grade (g/t) 1.15 1.15 1.16 1.16 1.03 1.03 1.02	Content (ozs) 15,600 15,600 6,700 6,700 46,400 46,400 30,800	Transition	Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred		Density	Grade	Content	Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 423,000 423,000 181,000 181,000 - 1,402,000 1,402,000 943,000 -	Density (t/m3) 2.35 2.35 2.68 2.68 2.81 2.81 2.82	Grade (g/t) 1.15 1.15 1.16 1.03 1.03 1.02	Content (ozs) 15,600 15,600 - 6,700 6,700 - 46,400 30,800

		Licence Area and Buf	fer Zone (D	ouble 19)				Ви	uffer Zone(Double 19)			Lic	ence Area (I	Double 19)	
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
Oxide	Indicated	152,000	2.11	1.68	8,200	Oxide	Indicated					Oxide	Indicated	152,000	2.11	1.68	8,200
ő	M+I	152,000	2.11	1.68	8,200	ŏ	M+I					ŏ	M+I	152,000	2.11	1.68	8,200
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
u	Measured	-			-	_	Measured					_	Measured	-			-
sitio	Indicated	89,000	2.11	1.67	4,800	sitio	Indicated					sitio	Indicated	89,000	2.11	1.67	4,800
Transition	M+I	89,000	2.11	1.67	4,800	Transition	M+I					Transition	M+I	89,000	2.11	1.67	4,800
	Inferred	-	-	-		-	Inferred					_	Inferred	-	-	-	-
	Measured	-			-		Measured						Measured	-			-
Fresh	Indicated	1,343,000	2.71	1.33	57,400	Fresh	Indicated					Fresh	Indicated	1,343,000	2.71	1.33	57,400
F	M+I	1,343,000	2.71	1.33	57,400	E.	M+I					F	M+I	1,343,000	2.71	1.33	57,400
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
	Measured	-			-		Measured						Measured	-			-
		1,584,000	2.62	1.38	70,400	豆	Indicated					Total	Indicated	1,584,000	2.62	1.38	70,400
tal	Indicated											.0				4.00	70.400
Total	M+I	1,584,000	2.62	1.38	70,400	Total	M+I					-	M+I	1,584,000	2.62	1.38	70,400
Total		1,584,000		1.38	70,400	Ď	M+I Inferred					-	M+I Inferred	1,584,000	2.62 -	1.38	70,400
Total	M+I Inferred		-	-	<u> </u>	Ţ		Buffer Zo	one(Gold M Hous		nd Gate				-	-	-
Total	M+I Inferred	-	-	-	<u> </u>	Ď		Buffer Zo			nd Gate Content (ozs)			<u>-</u>	-	-	-
Total	M+I Inferred	Area and Buffer Zone (C	Gold Mounta	ain and Ga	te House)	Ď			Hous Density	se) Grade	Content	F		Licence Area	Gold Moun	tain and G	ate House)
	M+I Inferred Licence A	Area and Buffer Zone (C	Gold Mounta	ain and Ga	te House)		Inferred		Hous Density	se) Grade	Content		Inferred	Licence Area	Gold Moun	tain and G	ate House)
Oxide	M+I Inferred Licence A	Area and Buffer Zone (C	Gold Mounta	ain and Ga	te House)	Oxide	Inferred Measured		Hous Density	se) Grade	Content	Oxide	Inferred	Licence Area	Gold Moun	tain and G	ate House)
	M+I Inferred Licence A Measured Indicated	Area and Buffer Zone (C	Gold Mounta	ain and Ga	te House)		Inferred Measured Indicated		Hous Density	se) Grade	Content		Inferred Measured Indicated	Licence Area	Gold Moun	tain and G	ate House)
Oxide	M+I Inferred Licence A Measured Indicated M+I	- Area and Buffer Zone (C Tonnage (t)	Gold Mounta Density (t/m3)	- ain and Ga Grade (g/t)	te House) Content (ozs)	Oxide	Inferred Measured Indicated M+I		Hous Density	se) Grade	Content	Oxide	Inferred Measured Indicated M+I	Licence Area Tonnage (t)	Gold Moun Density (t/m3)	Grade (g/t)	Content (ozs)
Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred	- Area and Buffer Zone (C Tonnage (t)	Gold Mounta Density (t/m3)	- ain and Ga Grade (g/t)	te House) Content (ozs)	Oxide	Measured Indicated M+I Inferred		Hous Density	se) Grade	Content	Oxide	Measured Indicated M+I Inferred	Tonnage (t)	Gold Moun Density (t/m3)	Grade (g/t)	Content (ozs)
sition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured	- Area and Buffer Zone (C Tonnage (t)	Gold Mounta Density (t/m3)	- ain and Ga Grade (g/t)	te House) Content (ozs)	sition Oxide	Measured Indicated M+I Inferred Measured		Hous Density	se) Grade	Content	sition Oxide	Measured Indicated M+I Inferred Measured	Tonnage (t)	Gold Moun Density (t/m3)	Grade (g/t)	Content (ozs)
Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 9,000	Gold Mounta Density (t/m3)	- ain and Ga Grade (g/t)	te House) Content (ozs)	Oxide	Measured Indicated M+I Inferred Measured Indicated		Hous Density	se) Grade	Content	Oxide	Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 9,000	Gold Moun Density (t/m3)	Grade (g/t)	Content (ozs)
sition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 9,000	Density (t/m3) 1.50	Grade (g/t) 0.70	Content (ozs)	sition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I		Hous Density	se) Grade	Content	sition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I	- Licence Area Tonnage (t) 9,000	Cold Moun Density (t/m3) 1.50	Grade (g/t) 0.70	Content (ozs) 200
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred	9,000 	Density (t/m3) 1.50	Grade (g/t) 0.70	Content (ozs)	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred		Hous Density	se) Grade	Content	Transition Oxide	Measured Indicated M+I Inferred Indicated Indicated Indicated Indicated Indicated Inferred In	9,000 - - - 192,000	Cold Moun Density (t/m3) 1.50	Grade (g/t) 0.70	Content (ozs) 200
sition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured MHI Inferred Measured	9,000 	Density (t/m3)	0.70	Content (ozs)	sition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Measured Indicated M+I Inferred Measured Measured Measured Measured		Hous Density	se) Grade	Content	sition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Measured	9,000 - - - 192,000	Cold Moun Density (t/m3) 1.50	Grade (g/t) 0.70	Content (ozs) 200
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Inferred Inferred	9,000 	Density (t/m3) 1.50 - 2.40	0.70	Content (ozs)	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Inferred Measured Inferred		Hous Density	se) Grade	Content	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Miniferred Measured Inferred	9,000 - - - 192,000	- (Gold Moun Density (t/m3) 1.50 - 2.40	0.70	Content (ozs) 200
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	9,000 	1.50 - 2.40	0.70 - 0.73	200 	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I		Hous Density	se) Grade	Content	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	9,000 192,000 	- (Gold Moun Density (t/m3) 1.50	0.70 - 0.73	200
Fresh Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Measured Inferred Inferred Inferred Indicated Indicated Indicated	9,000	1.50 - 2.40	0.70 - 0.73	200 	Fresh Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred		Hous Density	se) Grade	Content	Fresh Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated Indicated Indicated	9,000 9,000 192,000 2,166,000	- (Gold Moun Density (t/m3) 1.50	0.70 - 0.73	200
Transition Oxide	M+I Inferred Licence A Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	9,000	1.50 - 2.40		200 	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured		Hous Density	se) Grade	Content	Transition Oxide	Measured Indicated M+I Inferred	9,000	- (Gold Moun Density (t/m3) 1.50	0.70	200

		Licence Area and Bu	ffer Zone (R	load Cut)				Ви	ıffer Zone(l	Road Cut)				Lio	cence Area (Road Cut)	
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
Oxide	Indicated					Oxide	Indicated					Oxide	Indicated				
ő	M+I					ŏ	M+I					ŏ	M+I				
	Inferred						Inferred						Inferred				
<u></u>	Measured					<u> </u>	Measured					<u> </u>	Measured				
Transition	Indicated	12,000	2.50	0.74	300	Transition	Indicated					Transition	Indicated	12,000	2.50	0.74	300
lran	M+I	12,000	2.50	0.74	300	l an	M+I					la l	M+I	12,000	2.50	0.74	300
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
	Measured	-	-	-	-		Measured						Measured	-	-	-	-
Fresh	Indicated	213,000	2.82	0.85	5,800	Fresh	Indicated					Fresh	Indicated	213,000	2.82	0.85	5,800
표	M+I	213,000	2.82	0.85	5,800	E E	M+I					Ę	M+I	213,000	2.82	0.85	5,800
	Inferred	-	-	-	-		Inferred						Inferred	-	-	-	-
	Measured	-	-	-	-		Measured						Measured	-	-	-	-
	Indicated	225,000	2.80	0.85	6,100	Total	Indicated					Total	Indicated	225,000	2.80	0.85	6,100
tal	Indicated											₽	M+I	225,000	2.80	0.85	6,100
Total	M+I	225,000	2.80	0.85	6,100	=	M+I						IVITI				*, * * *
Total		225,000	2.80	0.85	6,100	Ĕ	M+I Inferred						Inferred	-	-	-	-
Total	M+I Inferred	·	-	-	<u> </u>	Ĕ		Buff	fer Zone (B	Boomeranç	1)			-	ence Area (B	-	-
Total	M+I Inferred	-	-	-	<u> </u>	Ĕ		Buff Tonnage (t)	fer Zone (B Density (t/m3)	Grade (g/t)	Content (ozs)			-		-	-
Total	M+I Inferred	Licence Area and Buff	er Zone (Bo	- oomerang) Grade	Content				Density	Grade	Content			Lice	ence Area (B	- coomerang) Content
	M+I Inferred	Licence Area and Buff	er Zone (Bo	- oomerang) Grade	Content		Inferred		Density	Grade	Content		Inferred	Lice	ence Area (B	- coomerang) Content
Oxide	M+I Inferred	Licence Area and Buff	er Zone (Bo Density (t/m3)	- comerang) Grade (g/t)	Content (ozs)	Oxide	Inferred	Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)	Oxide	Inferred	Lice Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	M+I Inferred Measured Indicated	Licence Area and Buff Tonnage (t)	er Zone (Bo Density (t/m3)	Grade (g/t)	Content (ozs)		Inferred Measured Indicated	Tonnage (t) 164,000	Density (t/m3)	Grade (g/t) 0.94	Content (ozs)		Inferred Measured Indicated	- Lice Tonnage (t) 314,000	Density (t/m3)	Grade (g/t)	Content (ozs)
Oxide	M+I Inferred Measured Indicated M+I	Licence Area and Buff Tonnage (t)	er Zone (Bo Density (t/m3)	- comerang) Grade (g/t) 1.01 1.01	Content (ozs)	Oxide	Inferred Measured Indicated M+I	Tonnage (t) 164,000 164,000	Density (t/m3) 1.87	Grade (g/t) 0.94 0.94	Content (ozs) 5,000 5,000	Oxide	Inferred Measured Indicated M+I	Tonnage (t) 314,000 314,000	Density (t/m3) 2.01 2.01	Grade (g/t) 1.05	Content (ozs)
Oxide	M+I Inferred Measured Indicated M+I Inferred	Licence Area and Buff Tonnage (t)	er Zone (Bo Density (t/m3)	- comerang) Grade (g/t) 1.01 1.01	Content (ozs)	Oxide	Measured Indicated M+I Inferred	Tonnage (t) 164,000 164,000	Density (t/m3)	Grade (g/t) 0.94 0.94	Content (ozs) 5,000 5,000	Oxide	Measured Indicated M+I Inferred	Tonnage (t) 314,000 314,000	Density (t/m3) 2.01 2.01	- Goomerang Grade (g/t) 1.05 1.05	Content (ozs)
sition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured	- Licence Area and Buff Tonnage (t) 478,000	- Density (t/m3) 1.96	- comerang) Grade (g/t) 1.01 1.01	Content (ozs) 15,500 15,500	sition Oxide	Measured Indicated M+I Inferred Measured	Tonnage (t) 164,000 164,000	Density (t/m3) 1.87 1.87	Grade (g/t) 0.94 0.94	Content (ozs) 5,000 5,000	sition Oxide	Measured Indicated M+I Inferred Measured	Tonnage (t) 314,000	Density (t/m3) 2.01 2.01 -	- Grade (g/t) 1.05	Content (ozs) 10,600 10,600
Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 478,000 478,000 242,000	- Per Zone (Both Density (t/m3) 1.96 1.96 - 2.35	- comerang) Grade (g/t) 1.01 1.01 - 1.02	Content (ozs) 15,500 15,500 - 7,900	Oxide	Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 164,000 164,000 79,000	Density (t/m3) 1.87 1.87	Grade (g/t) 0.94 0.94 - 0.98	Content (ozs) 5,000 5,000 - 2,500	Oxide	Measured Indicated M+I Inferred Measured Indicated	- Lice Tonnage (t) 314,000 - 162,000	2.01 2.01 2.35	- Grade (g/t) 1.05 1.05 - 1.04	Content (ozs) 10,600 10,600 - 5,400
sition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 478,000 478,000	- Per Zone (Both Density (t/m3) 1.96 1.96 - 2.35	- comerang) Grade (g/t) 1.01 1.01 - 1.02 1.02	Content (ozs) 15,500 15,500 - 7,900	sition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 164,000 164,000 79,000	Density (t/m3) 1.87 1.87 2.36 2.36	Grade (g/t) 0.94 0.94 - 0.98	Content (ozs) 5,000 5,000 - 2,500	sition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I	- Lice Tonnage (t) 314,000 - 162,000	2.01 2.01 2.35	- Grade (g/t) 1.05 1.05 - 1.04	Content (ozs) 10,600 10,600 - 5,400
Transition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	- Licence Area and Buff Tonnage (t) 478,000 478,000 - 242,000 242,000	Density (t/m3) 1.96 1.96 2.35 2.35	- pomerang) Grade (g/t) 1.01 1.01 - 1.02 1.02	Content (ozs) 15,500 15,500 - 7,900	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 164,000 164,000 79,000 79,000	Density (t/m3) 1.87 1.87 - 2.36 2.36	Grade (g/t) 0.94 0.94 0.98 0.98	Content (ozs) 5,000 5,000 - 2,500 2,500	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	- Lice Tonnage (t) 314,000 314,000 - 162,000 162,000	2.01 2.01 2.35 2.35	Grade (g/t) 1.05 1.05 - 1.04 1.04	Content (ozs) 10,600 10,600 - 5,400
sition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Measured	Tonnage (t) 478,000 478,000	- Per Zone (Book Density (t/m3) 1.96 1.96 2.35 2.35	- pomerang) Grade (g/t) 1.01 1.01 - 1.02 1.02	7,900 7,900	sition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Measured MHI Inferred Measured	Tonnage (t) 164,000 164,000 79,000 79,000	Density (t/m3) 1.87 1.87 - 2.36 2.36	Grade (g/t) 0.94 0.94 - 0.98 0.98	Content (ozs) 5,000 5,000 - 2,500 2,500 -	sition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured MHI Inferred	- Lice Tonnage (t) 314,000 - 162,000 162,000	2.01 2.01 2.35 2.35	Grade (g/t) 1.05 1.04 1.04	Content (ozs) 10,600 10,600 5,400 -
Transition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Inferred	Licence Area and Buff Tonnage (t) 478,000 478,000	- Per Zone (Both Density (t/m3) 1.96 1.96 - 2.35 2.35 - 2.89	- Domerang) Grade (g/t) 1.01 1.01 - 1.02 1.02 - 0.99	7,900 7,900 400,800	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Inferred Measured Inferred	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000	Density (t/m3) 1.87 1.87 2.36 2.36 2.88	Grade (g/t) 0.94 0.94 0.98 0.98 - 1.22	Content (ozs) 5,000 5,000 - 2,500 2,500 - 136,600	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Inferred	- Lice Tonnage (t) 314,000 314,000 - 162,000 162,000 - 9,083,000	2.01 2.01 2.35 2.35 2.89	Grade (g/t) 1.05 1.05 - 1.04 1.04 - 0.90	Content (ozs) 10,600 10,600 - 5,400 5,400 - 264,200
Transition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Licence Area and Buff Tonnage (t) 478,000 478,000	- Per Zone (Book Density (t/m3) 1.96 1.96 1.96 1.95 1.95 1.95 1.95 1.95 1.95 1.95 1.95	- pomerang) Grade (g/t) 1.01 1.01 - 1.02 1.02 - 0.99 0.99	7,900 7,900 400,800	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000	Density (t/m3) 1.87 1.87 2.36 2.36 2.88 2.88	Grade (g/t) 0.94 0.94 - 0.98 0.98 - 1.22	Content (ozs) 5,000 5,000 - 2,500 2,500 - 136,600	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Measured Indicated M+I	- Lice Tonnage (t) 314,000 314,000 - 162,000 162,000 - 9,083,000	2.01 2.01 2.35 2.35 2.89 2.89	Grade (g/t) 1.05 1.04 1.04 1.04 1.09 0.90	Content (ozs) 10,600 10,600 - 5,400 5,400 - 264,200
Fresh Transition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred Inferred Inferred Indicated Indicated Indicated	Licence Area and Buff Tonnage (t) 478,000 478,000 - 242,000 - 12,561,000 12,561,000 -	- Per Zone (Both Per	- pomerang) Grade (g/t) 1.01 1.01 - 1.02 1.02 - 0.99 0.99	7,900 7,900 400,800	Fresh Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated Indicated Indicated	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000 3,478,000	Density (t/m3) 1.87 1.87	Grade (g/t) 0.94 0.94 0.98 0.98 1.22 1.22	Content (ozs) 5,000 5,000 - 2,500 2,500 - 136,600 136,600	Fresh Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Indicated Indicated Indicated Indicated Indicated	- Lice Tonnage (t) 314,000 314,000 - 162,000 162,000 - 9,083,000 9,083,000	2.01 2.01 2.35 2.35 2.89 2.89	Grade (g/t) 1.05 1.05 - 1.04 1.04 - 0.90 0.90	Content (ozs) 10,600 10,600 - 5,400 5,400 - 264,200
Transition Oxide	M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Licence Area and Buff Tonnage (t) 478,000 478,000	- Per Zone (Both Density (t/m3) 1.96 1.96 - 2.35 2.35 - 2.89 2.89	- pomerang) Grade (g/t) 1.01 1.01 - 1.02 1.02 - 0.99 0.99 -	7,900 7,900 - 400,800 400,800	Transition Oxide	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured M+I Inferred Measured Measured Measured Measured	Tonnage (t) 164,000 164,000 79,000 79,000 - 3,478,000 3,478,000	Density (t/m3) 1.87 1.87 - 2.36 2.36 2.88 2.88	Grade (g/t) 0.94 0.94 - 0.98 0.98 - 1.22 1.22	Content (ozs) 5,000 5,000 2,500 2,500 - 136,600 136,600	Transition Oxide	Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated	- Lice Tonnage (t) 314,000 314,000 - 162,000 162,000 - 9,083,000 9,083,000	2.01 2.01 2.35 2.35 2.89 2.89		Content (ozs) 10,600 10,600 - 5,400 5,400 - 264,200 264,200

		Licence Area and But	ffer Zone (T	win Zone)				Buf	fer Zone (T	win Zone)				Lice	ence Area (1	win Zone)	
		Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)			Tonnage (t)	Density (t/m3)	Grade (g/t)	Content (ozs)
	Measured						Measured						Measured				
Oxide	Indicated	47,000	2.41	0.65	1,000	Oxide	Indicated	7,000	1.86	0.54	100	Oxide	Indicated	39,000	2.51	0.67	800
ő	M+I	47,000	2.41	0.65	1,000	ŏ	M+I	7,000	1.86	0.54	100	ő	M+I	39,000	2.51	0.67	800
	Inferred	-	-	-			Inferred	-	-	-	-		Inferred	-	-	-	-
چ	Measured	-	-	-	-	ڃ	Measured	-	-	-	-	چ	Measured	-	-	-	-
sitio	Indicated	2,000	2.43	0.74	100	sitio	Indicated	60	2.75	0.54	1	sitio	Indicated	2,000	2.43	0.74	100
Transition	M+I	2,000	2.43	0.74	100	Transition	M+I	60	2.75	0.54	1	Transition	M+I	2,000	2.43	0.74	100
	Inferred	-	-	-	<u>-</u>		Inferred	-	-	-	<u>-</u>		Inferred	-	-	-	-
	Measured	-	-	-	-		Measured	-	-	-	-		Measured	-	-	-	-
Fresh	Indicated	309,000	2.88	0.93	9,200	Fresh	Indicated	23,000	2.90	0.72	500	Fresh	Indicated	286,000	2.88	0.95	8,700
Ē	M+I	309,000	2.88	0.93	9,200	Ē	M+I	23,000	2.90	0.72	500	Ē	M+I	286,000	2.88	0.95	8,700
	Inferred	-	-	-	-		Inferred	-	-	-	<u>-</u>		Inferred	-	-	-	-
	Measured	-	-	-	-		Measured	-	-	-	-		Measured	-	-	-	-
Total	Indicated	358,000	2.81	0.89	10,300	Total	Indicated	31,000	2.65	0.68	700	Total	Indicated	327,000	2.83	0.91	9,600
ř	M+I	358,000	2.81	0.89	10,300	Ĕ	M+I	31,000	2.65	0.68	700	Ĕ	M+I	327,000	2.83	0.91	9,600
	Inferred	-	-	-	-		Inferred	-	-	-	-		Inferred	-	-	-	-
	Inferred	Licence Area an			-		Inferred	-	Buffer 2		-		Inferred	-	Licence		-
	Inferred				Content (ozs)		Inferred	Tonnage (t)			Content (ozs)		Inferred	Tonnage (t)			Content (ozs)
	Inferred Measured	Licence Area ar	nd Buffer Zo	one Grade			Inferred		Buffer 2	Zone Grade			Inferred Measured		Licence Density	Area Grade	
ide		Licence Area ar	nd Buffer Zo	one Grade		ide			Buffer 2	Zone Grade		epi			Licence Density	Area Grade	
Oxide	Measured	Licence Area an	Density (t/m3)	Grade (g/t)	(ozs)	Oxide	Measured	Tonnage (t)	Buffer 2 Density (t/m3)	Zone Grade (g/t)	(ozs)	Oxide	Measured	Tonnage (t)	Licence Density (t/m3)	Area Grade (g/t)	(ozs)
Oxide	Measured Indicated	Licence Area an Tonnage (t) 1,755,000	Density (t/m3)	Grade (g/t)	(ozs) 77,700	Oxide	Measured Indicated	Tonnage (t) 172,000	Buffer 2 Density (t/m3)	Grade (g/t)	(ozs) 5,100	Oxide	Measured Indicated	Tonnage (t) 1,583,000	Density (t/m3)	Area Grade (g/t)	(ozs) 72,600
	Measured Indicated M+I	Licence Area an Tonnage (t) 1,755,000 1,755,000	Density (t/m3)	Grade (g/t) 1.38	77,700 77,700		Measured Indicated	Tonnage (t) 172,000 172,000	Buffer 2 Density (t/m3)	Zone Grade (g/t) 0.92 0.92	(ozs) 5,100		Measured Indicated M+I	Tonnage (t) 1,583,000 1,583,000	Licence Density (t/m3) 2.15	Area Grade (g/t) 1.43 1.43	72,600 72,600
sition	Measured Indicated M+I Inferred	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000	Density (t/m3) 2.12 2.12 1.50	Grade (g/t) 1.38 1.38 0.70	77,700 77,700 200		Measured Indicated M+I Inferred	Tonnage (t) 172,000 172,000	Buffer 2 Density (t/m3)	Grade (g/t) 0.92 0.92	(ozs) 5,100	sition	Measured Indicated M+I Inferred	Tonnage (t) 1,583,000 1,583,000	Density (t/m3) 2.15 2.15 1.50	Area Grade (g/t) 1.43 1.43 0.70	72,600 72,600
sition	Measured Indicated M+I Inferred Measured	Licence Area an Tonnage (t) 1,755,000 1,755,000 9,000	Density (t/m3) 2.12 2.12 1.50	Grade (g/t) 1.38 1.38 0.70	77,700 77,700 200	ısition	Measured Indicated M+I Inferred Measured	Tonnage (t) 172,000 172,000	Buffer 2 Density (t/m3) 1.87 1.87	Grade (g/t) 0.92 0.92	5,100 5,100 -	sition	Measured Indicated M+I Inferred Measured	Tonnage (t) 1,583,000 1,583,000 9,000	Density (t/m3) 2.15 2.15 1.50	Area Grade (g/t) 1.43 1.43 0.70	72,600 72,600 200
	Measured Indicated M+I Inferred Measured Indicated	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000	Density (t/m3) 2.12 2.12 1.50 - 2.42	1.38 1.38 0.70	77,700 77,700 200 - 32,600		Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 172,000 172,000 - 79,000	Density (t/m3) 1.87 1.87	Zone Grade (g/t) 0.92 0.92 - 0.98	5,100 5,100 2,500		Measured Indicated M+I Inferred Measured Indicated	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000	Licence 2.15 2.15 2.15 1.50 - 2.43	Area Grade (g/t) 1.43 1.43 0.70 - 1.39	72,600 72,600 200 - 30,000
sition	Measured Indicated M+I Inferred Measured Indicated M+I	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000 750,000	Density (t/m3) 2.12 2.12 1.50 - 2.42 2.42	Grade (g/t) 1.38 1.38 0.70 - 1.35 1.35	77,700 77,700 200 - 32,600 32,600	ısition	Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 172,000 172,000 - 79,000	Buffer 2 Density (t/m3) 1.87 1.87 - 2.36 2.36	Cone Grade (g/t) 0.92 0.92 - 0.98 0.98	5,100 5,100 2,500	sition	Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000 670,000	Licence Density (t/m3) 2.15 2.15 1.50 - 2.43 2.43	Area Grade (g/t) 1.43 1.43 0.70 - 1.39 1.39	72,600 72,600 200 - 30,000 30,000
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000 750,000 192,000	Density (t/m3) 2.12 2.12 1.50 - 2.42 2.42 2.40	1.38 1.38 0.70 - 1.35 1.35	77,700 77,700 200 - 32,600 32,600 4,500	Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 172,000 172,000 - 79,000 79,000	Buffer 2 Density (t/m3) 1.87 1.87 - 2.36 2.36	Zone Grade (g/t) 0.92 0.92 - 0.98 0.98	5,100 5,100 2,500 2,500	sition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000 670,000 192,000	Licence Density (t/m3) 2.15 2.15 1.50 - 2.43 2.43 2.40	Area Grade (g/t) 1.43 1.43 0.70 - 1.39 1.39 0.73	72,600 72,600 200 - 30,000 30,000
sition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000 750,000 192,000	2.12 2.12 1.50 - 2.42 2.40	1.38 1.38 0.70 - 1.35 1.35	77,700 77,700 200 - 32,600 32,600 4,500	esh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 172,000 172,000 79,000 79,000	Buffer 2 Density (t/m3) 1.87 1.87 - 2.36 2.36	Grade (g/t) 0.92 0.92 - 0.98 0.98	5,100 5,100 2,500 2,500	sition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000 192,000	Licence Density (t/m3) 2.15 2.15 1.50 - 2.43 2.43 2.40 -	Area Grade (g/t) 1.43 1.43 0.70 - 1.39 1.39 0.73	72,600 72,600 200 - 30,000 4,500 -
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000 750,000 192,000 - 25,027,000	Density (t/m3) 2.12 2.12 1.50 - 2.42 2.42 2.40 - 2.84	1.38 1.38 0.70 - 1.35 1.35 0.73 - 1.18	77,700 77,700 200 - 32,600 32,600 4,500 - 947,900	Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Tonnage (t) 172,000 172,000 - 79,000 79,000 - 3,517,000	Buffer 2 Density (t/m3) 1.87 1.87 - 2.36 2.36 - 2.88	Zone Grade (g/t) 0.92 0.92 - 0.98 0.98 - 1.221	5,100 5,100 2,500 2,500 - 137,700	esh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Inferred	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000 670,000 192,000 - 21,510,000	Licence Density (t/m3) 2.15 2.15 1.50 - 2.43 2.43 2.40 - 2.84	Area Grade (g/t) 1.43 1.43 0.70 - 1.39 1.39 0.73 - 1.17	72,600 72,600 200 - 30,000 30,000 - 810,200
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Measured Indicated	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000 192,000 - 25,027,000 25,027,000	2.12 2.12 1.50 - 2.42 2.42 2.40 - 2.84 2.84	Grade (g/t) 1.38 1.38 0.70 - 1.35 1.35 1.18 1.18	77,700 77,700 200 - 32,600 32,600 4,500 - 947,900	Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 172,000 172,000 79,000 79,000 - 3,517,000 3,517,000	Buffer 2 Density (t/m3) 1.87 1.87 - 2.36 2.36 - 2.88 2.88	Cone Grade (g/t) 0.92 0.92 - 0.98 0.98 - 1.221 1.22	5,100 5,100 2,500 2,500 - 137,700	esh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000 192,000 - 21,510,000 21,510,000	Licence Density (t/m3) 2.15 2.15 1.50 - 2.43 2.43 2.40 - 2.84 2.84	Area Grade (g/t) 1.43 1.43 0.70 - 1.39 1.39 0.73 - 1.17	72,600 72,600 200 - 30,000 4,500 - 810,200
Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000 750,000 192,000 - 25,027,000 25,027,000 5,706,000	Density (t/m3) 2.12 2.12 1.50 - 2.42 2.42 2.40 - 2.84 2.84 2.71	1.38 1.38 0.70 - 1.35 1.35 0.73 - 1.18 1.18	77,700 77,700 200 200 32,600 4,500 - 947,900 947,900 176,100	Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Indicated M-I Inferred	Tonnage (t) 172,000 172,000 - 79,000 79,000 - 3,517,000 3,517,000	Buffer 2 Density (t/m3) 1.87 1.87 - 2.36 2.36 - 2.88 2.88	Grade (g/t) 0.92 0.92 - 0.98 0.98 - 1.221 1.22	(ozs) 5,100 5,100 - 2,500 2,500 - 137,700 137,700	al Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated Indicated M-I Inferred	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000 192,000 - 21,510,000 21,510,000 5,493,000	Licence Density (t/m3) 2.15 2.15 1.50 - 2.43 2.43 2.40 - 2.84 2.84 2.82	Area Grade (g/t) 1.43 1.43 0.70 - 1.39 1.39 0.73 - 1.17 1.00	72,600 72,600 200 - 30,000 4,500 - 810,200
Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured	Licence Area and Tonnage (t) 1,755,000 1,755,000 9,000 - 750,000 192,000 - 25,027,000 25,027,000 5,706,000	2.12 2.12 1.50 - 2.42 2.42 2.40 - 2.84 2.71	1.38 1.38 0.70 - 1.35 1.35 0.73 - 1.18 1.18	77,700 77,700 200 200 - 32,600 4,500 - 947,900 947,900 176,100	Fresh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 172,000 172,000 - 79,000 79,000 - 3,517,000 3,517,000	Buffer 2 Density (t/m3) 1.87 1.87 - 2.36 2.36 - 2.88 2.88	Zone Grade (g/t) 0.92 0.92 - 0.98 0.98 - 1.221 1.22	5,100 5,100 - 2,500 2,500 - 137,700 137,700	esh Transition	Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred Measured Indicated M+I Inferred	Tonnage (t) 1,583,000 1,583,000 9,000 - 670,000 192,000 - 21,510,000 21,510,000 5,493,000	Licence Density (t/m3) 2.15 2.15 1.50 - 2.43 2.43 2.40 - 2.84 2.84 2.82	Area Grade (g/t) 1.43 1.43 0.70 - 1.39 0.73 - 1.17 1.17 1.00	72,600 72,600 200 - 30,000 4,500 - 810,200 810,200 - 176,100



15 ADJACENT PROPERTIES

Although the Kibi area is blanketed by mining concessions, very little systematic exploration work for bedrock gold deposits has been conducted over the years in the Kibi Greenstone Belt. This reflects the fact that the Kibi area has traditionally been recognized as an alluvial gold district, and the surrounding concessions have been held since the mid-1980s to early 1990s specifically for their alluvial gold potential.

1	6	OTHER	RFI F	TNAV	$D\Delta T\Delta$		INFORMATION	J
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None.

17 INTERPRETATION AND CONCLUSIONS

The work undertaken has confirmed the presence of a number of auriferous bodies (i.e., Big Bend, East Dyke, Mushroom, Road Cut, South Ridge, Double 19, Gate House and Gold Mountain, Twin Zone and Boomerang) within the concession, as well as providing a structural model that explains the paragenesis of the mineralized bodies. The geological continuity has been demonstrated.

The structural information has been utilised in the mineral resource estimate completed and presented in Table 17.1.

<u>Cautionary Note:</u> The Mineral Resource Estimate calculation includes the area termed the "Buffer Zone", which encompasses the "open ground" between the existing Apapam mining lease boundary up to the Forest Reserve boundary. Xtra-Gold applied for this Buffer Zone in the original application of extension/renewal of the Apapam mining lease on June 17th, 2015, and also in the updated extension/renewal dated June 28th, 2019. These applications were approved by the Technical Committee of the Minerals Commission and certified to be in conformity with the official cadastral system introduced under Ghana Mineral Law. Xtra-Gold is currently awaiting formal approval of the extension/renewal. Refer to the Apapam mining lease Status Opinion prepared by REM Law Consultancy (Appendix B) for further details on the lease renewal / extension, including the Buffer Zone area.

Table 17.1: Mineral Resource Declaration - September 2024

Declared in terms of the CIM Standards Cut-off: Au 0.5 g/t

	Apapam License and Buffer Areas				Buffer Zo	ne between L Reserv		Forest	Apapam License Area					
	Indicated					Indicat	ed			Indicated				
	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs		Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	
Big Bend	6,472,000	2.78	1.48	307,400	-	-	ı	ı		6,472,000	2.78	1.48	307,400	
East Dyke	3,102,000	2.72	1.49	148,800	-	-	ı	ı		3,102,000	2.72	1.49	148,800	
Mushroom	505,000	2.63	1.37	22,200	16,000	2.82	1.18	600		489,000	2.64	1.37	21,600	
South Ridge	2,005,000	2.70	1.07	68,700	-	=		ı		2,005,000	2.70	1.07	68,700	
Double 19	1,584,000	3	1	70,400	-	-	-	-		-	-	-	-	
GH and GM	-	-	-	-	-	-	-	-		-	-	-	-	
Road Cut	225,000	3	1	6,100	-	-	-	-		225,000	2.80	0.85	6,100	
Boomerang	13,281,000	2.84	0.99	424,300	3,722,000	2.83	1.20	144,100		9,559,000	2.85	0.91	280,200	
Twin Zone	358,000	2.81	0.89	10,300	31,000	2.65	0.68	700		327,000	2.83	0.91	9,600	
	27,532,000	2.79	1.20	1,058,200	3,769,000	2.83	1.20	145,400		22,179,000	2.79	1.18	842,400	
		Inferr	ed			Inferre	ed		•		Inferr	ed		
	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	•	Tonnage (t)	Density (t/m3)	Grade Au (g/t)	Ozs	
Big Bend	1,257,000	2.82	1.03	41,400	-	-	-	-		1,257,000	2.82	1.03	41,400	
East Dyke	1,128,000	2.84	1.19	43,300	-	-	ı	ı		1,128,000	2.84	1.19	43,300	
Mushroom	-	-	=	-	-	=	ı	-		-		-	-	
South Ridge	943,000	2.82	1.02	30,800	-	-	-	-		943,000	2.82	1.02	30,800	
Double 19	-	-	=	-	-	-	-	-		-	=	-	-	
GH and GM	2,366,000	2.76	0.79	65,200	-	-	-	-		2,366,000	2.76	0.79	65,200	
Road Cut	-	-	-	-	-	-	-	-		-	-	-	-	
Boomerang	-	-	-	-	-	-	-	-		-	-	-	-	
Twin Zone	-	-	-	-	-	-	-	-		-	-	-	-	
	5,694,000	2.80	0.99	180,700	-			-		5,694,000	2.80	0.99	180,700	

18 RECOMMENDATIONS

Based on the results of the 2024 Mineral Resource Estimate (MRE) and exploration results on early-stage targets across the project area, Pivot and TECT recommend a two-phase exploration program to further advance the Kibi Gold Project.

Phase 1 is aimed of advancing the Kibi Gold Project by further delineation of existing mineral resources and identification of additional resource bodies within the MRE footprint area, continued advancement of early-stage targets across the Apapam concession, and property-scale target generation exploration work.

Phase 2, designed to support the continued advancement of the project, includes additional drilling to further define mineral resources, an updated mineral resource estimate, completion of a Preliminary Economic Assessment (PEA), metallurgical test work, and collection of additional data to support future scoping studies. With the implementation of Phase 2 being contingent upon the success of Phase 1.

18.1 Phase 1 Exploration

The recommended Phase 1 work program is detailed below:

18.1.1 Mineral Resource Estimate Footprint Area Work Program

Additional resource body exploration drilling targeting the down-plunge extensions of the respective 3D litho-structural deposit models and/or the lateral extensions of the resource bodies. With drillhole design aimed at intersecting the dominant vein set geometries.

Infill drilling to upgrade Inferred Mineral Resources to the Indicated category criteria. Conversion drilling should initially focus on the South Ridge and Gatehouse and Gold Mountain mineral resource bodies based on the relatively shallow depth of the target mineralization. With these bodies also offering potential for resource growth.

Continued exploration drilling of the early-stage Lone Tree (Zone 1) and Orange No. 5 (Zone 4) targets, with the positioning of these prospects along the same northwestern limb of the F2 anticline on which Gatehouse and Gold Mountain are situated and in proximity to a folded 2nd – order syncline along the hanging wall of the D2 shear, respectively, offering potential for shallow, near surface resource growth.

Systematic scout drilling following the 1st-order F2 folded stratigraphy, along strike from known deposits, to identify additional gold-prospective litho-structural settings that offer potential for resource growth.

Continued re-logging of historical drill core geared towards the clarification / standardization of lithological units to facilitate detailed 3D litho-structural modelling.

Continued collection of more comprehensive / representative structural measurements utilizing the IQL-Logger structural logging laser-device from historical drill core across relevant mineralization zones to further determine geometries of dominant vein sets.

A comparative study on the locality and geometry of measured veins and gold grades, from which spatial and statistical relationships can be derived and related to the extent of gold mineralization.

Continued mechanical trenching to permit detailed geological / structural mapping, establish litho-structural controls of early-stage mineralization zones to help guide drilling efforts, test priority gold-in-soil anomalies, and ground proof 3D geophysical and geological / structural modelling targets.

Completion of the drone LiDAR topographical survey over the MRE footprint area (Zone 1 & Zone 2), combined with the professional Differential GNSS surveying of the MRE footprint drill collars, to enhance the confidence level of drill collar positioning in preparation for the Phase 2 PEA study.

18.2 Property-Scale Work Program

Additional follow up drilling on the Cobra Creek (Zone 5) auriferous shear system will be prioritized. With drillhole design aimed at intersecting the sub-horizontal, en-echelon, high-grade auriferous vein arrays and shallow-plunging stockwork "shoots". Drilling efforts will also include scout drilling targeting possible dilational zone-hosted gold mineralization along the sheared QFP body contacts.

Further exploration drilling will also be conducted on the Akwadum South (Zone 7) target to follow up on the exploration significant gold mineralization traced by the 2019 scout drilling program over an approximately 1,000 m strike-length, and to a vertical depth exceeding 100 m, along a NE-trending volcaniclastic rock package exhibiting widespread silica-sericite-pyrite alteration with associated quartz veining.

The three well-defined auriferous rock float fields identified by the Zone 4 grab sampling will be subjected to systematic scout trenching to establish the in situ source and economic potential of the auriferous silicified / pyritized siltstone material.

The area to the southwest of the Gatehouse and Gold Mountain (Zone 1) Mineral Resource bodies, along the strike-extension of the northwestern limb of the F2 anticline on which Gatehouse and Gold Mountain are situated, will be subjected to reconnaissance geology, grab sampling and scout trenching targeting potential limb-parallel shear zones. Similarly, Zone 9 which straddles the hinge of the 1st-order F2 anticlinorium and the western limb of the fold will also be subjected to additional ground proofing.

Based on encouraging scout trenching results to date over an approximately 400 m segment of the Hillcrest Shear (Zone 6), it is recommended that the remaining 800 m of the approximately 1,200 m long anomalous gold-in-soil trend spatially associated with the Hillcrest Shear target,

located on the Akim Apapam reconnaissance licence application, be subjected to systematic scout trenching.

Continued high-resolution 3D implicit geological and structural modelling to guide ground proofing of geophysical interpretations and help define licence- to target-scale geological settings and structural geometries.

Continued scout trenching to test priority gold-in-soil anomalies and to ground proof 3D geophysical and geological / structural modelling targets.

18.3 Cost Estimate

A cost estimate for the recommended two-phase work program is provided in Table 18.1 to serve as a guideline. The estimated drilling expenditures are based on all-inclusive drilling costs utilizing Xtra-Gold's in-house operated diamond core drill rigs. Total expenditures are estimated at USD 5,185,000, including: USD 3,585,000 for Phase 1; and USD 1,600,000 for Phase 2. With the implementation of Phase 2 being contingent upon the success of Phase 1.

PHASE 1: WORK PROGRAM	BUDGET COS
MRE Footprint Work Program	
Exploration / Conversion Drilling (35,000 m)	\$2,275,000
Historical Core Re-Logging & Structural Logging / Comparative Vein Study	\$25,000
Continued 3D Litho-Structural Modelling / Target Generation	\$35,000
Mechanical Trenching (2,500 m)	\$100,000
Property-Scale Work Program	
Exploration Drilling (15,000 m)	\$975,000
Scout Manual / Mechanical Trenching (2,500 m)	\$100,000
Completion of drone LiDAR topographical survey (Zone 1 & Zone 2)	\$40,000
Professional Differential GNSS Survey of MRE Footprint Drill Collars	\$35,000
Phase 1 subtotal	\$3,585,000
PHASE 2: Work Program	BUDGET COS
Exploration Drilling (15,000 m)	\$975,000
MRE Update & PEA	\$250,000
Metallurgical Test Work	\$150,000
Environmental Baseline Study	\$150,000
Community and Stakeholder Engagement	\$75,0000
	\$1,600,000

19 REFERENCES

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Appendix A

Authors Certificate

Authors Certificate Ken Lomberg (Director Geology and Resources)

As the Lead Qualified Person and Complier of the report entitled "**Xtra-Gold Resources Corporation Kibi Gold Project**" with an effective date of 30 September 2024, I hereby state:

- 1. My name is Kenneth Graham Lomberg and I am the Director (Geology and Resources) for Pivot Mining Consultants (Pty) Ltd, Lower Ground Floor, Island House, Constantia Office Park, Corner 14th Avenue and Hendrik Potgieter Road, Weltevreden park, 1709, Roodepoort, South Africa.
- 2. I am a practicing geologist and a registered with the South African Council for Natural Professionals.
- 3. I have a B.Sc. (Hons) (Geology), B.Com. (Economics and Statistics) and an M.Eng. (Mining Engineering). I have been 36 years mining industry experience (especially in platinum and gold). I have practiced my profession continuously since 1985.
- 4. I have over 5 years of relevant experience, having completed Mineral Resource estimations on various gold properties hosting Magmatic orogenic-style mineralization. I have the relevant experience of the type of deposit and of the resource estimation that is the subject of this report.
- I have performed consulting work on various projects. These assignments have ranged from listings documents, CPRs, ITRs, feasibility studies, NI43-101 compliant resource estimations and valuations.
- 6. I am a 'Qualified Person' that term is defined in and for the purposes of the National Instrument 43-101 (Standards of Disclosure for Mineral Projects) (the Instrument).
- 7. I have visited the Kibi Project for personal inspection on 29 November to 5 December 2020.
- 8. I have prepared all sections of this report and am responsible for the Report and the Mineral Resource declaration.
- I am not aware of any material fact or material change with respect to the subject matter of the Report that is not reflected in the Report, the omission of which would make the Report misleading.
- 10. I declare that this Report appropriately reflects the Qualified Person's/author's view.
- 11. I do not have nor do I expect to receive a direct or indirect interest in the Mineral Properties of Xtra-Gold Resources Corporation, and I do not beneficially own, directly or indirectly, any securities of Xtra-Gold Resources Corporation or any associate or affiliate of such company.
- 12. I am independent of Xtra-Gold Resources Corporation.
- 13. I have read the Instrument and Form 43-101F1 (the Form) and the Report has been prepared in compliance with the Instrument and the Form.
- 14. At the effective date of the Report, to the best of my knowledge, information and belief, the Report contains all scientific and technical information that is required to be disclosed to make the Report not, mjsleading.

Dated at Johannesburg and 30 September 2024.

Mr Kenneth Lomberg Director (Geology and Resources)

Authors Certificate Corné Koegelenberg (Principal Geoscientist)

As the Lead Qualified Person and Complier of the report entitled "**Xtra-Gold Resources Corporation Kibi Gold Project**" with an effective date of 30 September 2024, I hereby state:

- 1. My name is Corné Koegelenberg and I am the Principal Geoscientist for TECT Geological Consultants (CC), Unit 3, Metrohm House, 20 Gardner Williams Avenue, Paardevlei, Somerset West, 7130, Western Cape, South Africa.
- 2. I am a practicing geologist and am registered with the South African Council for Natural Professionals (SACNASP 114569).
- 3. I have a B.Sc. (Hons) (Applied Geology), M.Sc. (Economic Geology) and Ph.D. (Structural Geology and Tectonics). I have been 9 years mining industry experience (especially in gold, copper, iron, PGE). I have practiced my profession continuously since 2012.
- 4. I have over 8 years of relevant experience, having completed structural investigations and the construction of 3D geological models on various gold properties hosting either magmatic or orogenic style mineralization. I have the relevant experience of the type of deposit, its structural setting(s) and best-practise 3D modelling methodologies that are the subject of this report.
- 5. I have performed consulting work on various projects. These assignments have ranged from structural investigations to 3D geological modelling for exploration, mining resource management and geotechnical engineering, and NI43-101 compliant resource estimations and valuations.
- 6. I am a 'Qualified Person' as defined in and for the purposes of the National Instrument 43-101 (Standards of Disclosure for Mineral Projects) (the Instrument).
- 7. I have visited the Kibi Project for personal inspection on 29 November to 10 December 2020.
- 8. I prepared 1, 2, 7, 17 and 18 sections of this report and am responsible for the Report. I am responsible for the Geology (Lithology, Structure, Mineralization) and 3D Deposit Modelling.
- I am not aware of any material fact or material change with respect to the subject matter of the Report that is not reflected in the Report, the omission of which would make the Report misleading.
- 10. I declare that this Report appropriately reflects the Qualified Person's/author's view.
- 11. I do not have nor do I expect to receive a direct or indirect interest in the Mineral Properties of Xtra-Gold Resources Corporation, and I do not beneficially own, directly or indirectly, any securities of Xtra-Gold Resources Corporation or any associate or affiliate of such company.
- 12. I have read the Instrument and Form 43-101F1 (the Form) and the Report has been prepared in compliance with the Instrument and the Form.
- 13. At the effective date of the Report, to the best of my knowledge, information and belief, the Report contains all scientific and technical information that is required to be disclosed to make the Report not misleading.

Dr Corné Koegelenberg Principal Geoscientist (Structural and Economic Geology)

Authors Certificate Ian Basson (Principal Geoscientist)

As the Qualified Person and Complier of the report entitled "Xtra-Gold Resources Corporation Kibi Gold Project" with an effective date of 30 September 2024, I hereby state:

- 1. My name is Ian James Basson and I am the Principal Geoscientist for TECT Geological Consulting (CC), Unit 3, Metrohm House, 20 Gardner Williams Avenue, Paardevlei, Somerset West, 7130, Western Cape, South Africa.
- 2. I am a practicing geologist and a registered with the South African Council for Natural Professionals (SACNASP 400006/04).
- 3. I have a B.Sc. (Hons) (Geology and Applied Geology) and Ph.D. (Structural Geology). I have 23 years mining industry experience (especially in gold, copper, iron and PGE). I have practiced my profession continuously since 1997.
- 4. I have over 15 years of relevant experience, having completed structural investigations and the construction of 3D geological models on various gold properties hosting either magmatic, orogenic, greenstone or Birimian-style mineralization. I have the relevant experience of the type of deposit, its structural setting(s) and best-practise 3D modelling methodologies that are the subject of this report.
- 5. I have performed consulting work on various projects. These assignments have ranged from listings documents, structural investigations and 3D geological modelling for exploration, mining resource management and geotechnical engineering, and NI43-101 compliant resource estimations and valuations.
- 6. I am a 'Qualified Person' as defined in and for the purposes of the National Instrument 43-101 (Standards of Disclosure for Mineral Projects) (the Instrument).
- 7. I have not visited the Kibi Project for personal inspection.
- 8. I prepared Sections 1, 2, 7, 17 and 18 of this report .I am responsible for reviewing the Geology (Lithology, Structure, Mineralization) and 3D Deposit Modelling.
- 9. I am not aware of any material fact or material change with respect to the subject matter of the Report that is not reflected in the Report, the omission of which would make the Report misleading.
- 10. I declare that this Report appropriately reflects the Qualified Person's/author's view.
- 11. I do not have nor do I expect to receive a direct or indirect interest in the Mineral Properties of Xtra-Gold Resources Corporation, and I do not beneficially own, directly or indirectly, any securities of Xtra-Gold Resources Corporation or any associate or affiliate of such company.
- 12. I have read the Instrument and Form 43-101F1 (the "Form) and the Report has been prepared in compliance with the Instrument and the Form.
- 13. At the effective date of the Report, to the best of my knowledge, information and belief, the Report contains all scientific and technical information that is required to be disclosed to make the Report not misleading.

Dated at Somerset West 30 September 2024.

Dr lan James Basson Principal Geoscientist (Structural and

Economic Geology)

Appendix B

Apapam Mining Lease Status Opinion prepared by REM Law Consultancy



REM Law Consultancy

Legal and Investment Advisory Services, Notaries Public No. 28, 4th Street West, North Legon Residential Area, Accra. Digital Postal Address: Gm – 048 - 8863, Accra - Ghana Tel: (233-30) 2522 658 Mobile (233-20) 145 87 92 Email: remlaw@africaonline.com.gh; info@remlawgh.com

Website: www.remlawgh.com

Innocent Akwayena: LLB. Hons; LLM Emmanuel Sekor: LLB. Hons; LLM

Enyonam Dedey: LLB. Hons; LLM Winnifred Odoi-Quarshie LLB Hons; BL

Our Ref: REM24/X1/NNR

August 8, 2024

Xtra-Gold Resources Corp. Village Road Shopping Plaza Suite#2150, P.O. Box AP 59217 Nassau, Bahamas

Xtra-Gold Mining Limited H/No. 2 Extra Gold Street Kwabeng, Eastern Region Ghana

Dear Sirs.

Re: Xtra-Gold Resources Corp. ("Xtra-Gold")

We are Ghanaian counsel to Xtra-Gold Resources Corp. ("Xtra-Gold") and its Ghanaian subsidiary Xtra-Gold Mining Limited ("Xtra-Gold Ghana") in respect of mining law matters in Ghana. We have been requested by Xtra-Gold to provide an opinion as to the current status of the Apapam mining lease over the Apapam mining concession (the "Apapam Mining Lease") held by Xtra-Gold Ghana, which opinion is required of Xtra-Gold pursuant to the reporting requirements under Canadian guidelines for NI 43-IDI.

Capitalized terms not expressly defined herein shall have the meanings ascribed to them in the Ghana Minerals and Mining Act 2006, Act 703 and Regulations made under the Ghana Minerals and Mining (Licensing Regulations 2012 (LI 2176) (the "Ghana Mining Laws").

In this regard, we have examined the following:-

Apapam Mining Lease which is the mining lease for gold and associated substances dated 18th
 December 2008 and registered at the Lands Commission as No. 24/2009 for a term of seven (7)

years, subject to renewal, over an area covering 33.65 square kilometres approximately, comprising the Apapam gold mine which said lease was made between the Government and Xtra-Gold Ghana.

- Minerals and Mining Act 2006 Act 703 (the "Ghana Mining Law").
- 3. Minerals and Mining Licensing Regulations 2012 L.I. 2176 (the "Licensing Regulations").
- 4. Letter dated June 17, 2015 by Xtra-Gold Ghana to the Minerals Commission applying for extension/renewal of the Apapam Mining Lease over a portion of the Lease identified as area (A) and (B) consisting of two blocks totaling 8.19 square kilometers; and a Prospecting Licence over the remaining Lease area totaling 25.41 square kilometers.
- 5. Follow up letter dated February 27, 2018 by Xtra-Gold Ghana to the Minerals Commission requesting further action on the Xtra-Gold Ghana application for extension/ renewal of the Apapam Mining Lease.
- **6.** Updated letter dated June 28, 2019 by Xtra-Gold Ghana to the Minerals Commission applying for extension/renewal of the Apapam Mining Lease amending area applied for to 31.22 square kilometers (or 147 cadastral units), which area includes the "Buffer Zone" area between the current Lease boundary up to the edge of the Forest Reserve.

We have also examined the original, or copies certified or otherwise, identified to our satisfaction, of such corporate records of Xtra-Gold Ghana as we have considered necessary for the purpose of this opinion. As to questions of fact material to this opinion we have, to the extent that such facts were not independently determined by us, relied upon other agreements, instruments, documents and certificates, including certificates of public officials and written confirmations of Xtra-Gold Ghana as of the date hereof. In all such examinations we have assumed the genuineness of all signatures, the authenticity of all documents submitted to us as originals and the conformity to the originals of all documents submitted to us as copies or facsimiles thereof. We have also considered such matters of Ghanaian law as we have considered necessary for the purpose of this opinion.

We are qualified to practice law only in Ghana. We have not made any independent examination of the laws of a jurisdiction other than Ghana and we do not express or imply any opinion as to the laws of

any other jurisdiction. To the extent that certificates upon which we have relied are based upon any assumption or subject to any limitation or qualification, our opinions rendered in reliance thereon are also based upon such assumptions and subject to such limitations and qualifications.

Based upon our review of the above referenced documents we note that:

- Pursuant to section 44(3) of the Ghana Mining Law, Xtra-Gold Ghana as holder of the Apapam Mining Lease is legally entitled to apply in the prescribed manner to the Government for extension/renewal of each mining lease for such further period up to maximum thirty (30) years in respect of any or all of the minerals that were the subject of the initial term of the said mining lease.
- Once such application for renewal of the Apapam Mining Lease was duly made by Xtra-Gold Ghana and Xtra-Gold Ghana had materially complied with all the obligations imposed by the lease and the Ghana Mining Laws, the Government is obliged to renew the lease.
- 3. Once Xtra-Gold Ghana had made an application to the Government for renewal/extension of the Apapam Mining Lease and the terms of the lease expired whilst the application for its renewal remained pending, section 44(4) of the Ghana Mining Law stipulates that the mining lease shall continue in force in respect of all the land that is subject to the application until the application has been duly determined by the Government.
- 4. The Xtra-Gold Ghana's application to the Minerals Commission for renewal of the Apapam Mining Lease had been reviewed by the Mineral Commission's Technical Committee on Mineral Rights and certified to be in conformity with the Official Cadastral System introduced under the Ghana Mining Laws to govern the grant of Mineral Rights. Upon issuance of the renewal/extension lease by Mineral Commission, it will include the "Buffer Zone" area between the current Lease boundary up to the edge of the Forest Reserve.
- The Xtra-Gold Ghana's application for renewal of the Apapam Mining Lease submitted to the Minerals Commission has materially complied with all the conditions required under the Ghana Mining Law for renewal/extension of the Apapam Mining Lease.

For the purposes of this opinion, the following term have the following meanings:

"Apapam Mineral Rights" means the Apapam Mining Lease and its related surface rights as described in the lease.

"Minerals Commission" means the Ghana Minerals Commission established under the Minerals Commission Act 1993 (Act 450) as a body corporate with perpetual succession and vested with power and authority to regulate and manage the mineral resources of Ghana.

"Minister" means the Ghanaian government minister responsible for Lands and Natural Resources.

Based and relying upon, and subject to the foregoing, we are of the opinion that:

- Xtra-Gold Ghana is the registered lessee of the Apapam Mining Lease, with a good leasehold title thereto subject to the reversionary interest of the Government.
- 2. Xtra-Gold Ghana is the registered and beneficial holder of the Apapam Mineral Rights.
- 3. The Apapam Mineral Rights are in good standing.
- 4. Xtra-Gold Ghana's interest as the holder of the Apapam Mineral Rights is subject only to those statutory rights and options conferred on the Government in the Ghana Mining Law.
- 5. The renewal/extension application for the Apapam Mineral Rights of Xtra-Gold Ghana that have expired or are due for renewal/extension have been duly submitted in accordance with the applicable Ghana Mining Laws and therefore the renewed Apapam Mining Lease will in all probability be granted to cover the updated lease map submitted by Xtra-Gold Ghana that has been certified by the Technical Committee of the Minerals Commission as conforming to the Cadastral System and which map is attached hereto as Appendix "A".

Consequently, upon issuance of the renewal/extension lease by the Government of Ghana acting through the Minister the lease area will most probably include the "Buffer Zone" area between the current Lease boundary up to the edge of the Forest Reserve.

This opinion relates exclusively to the Apapam mining concession and is for the sole use and benefits of the parties to whom it is addressed and for the purpose referred to above. Accordingly, it cannot be relied upon by any other party or used in any other transaction without our express written consent.

Yours truly,

Innocent Akwayena

Managing Consultant

REM Law Consultancy

Encl.

APPENDIX "A"

