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ASSESSMENT REPORT

ON 1997 DIAMOND DRILLING PROGRAM,

SNIP PROPERTY JIM 1 CLAIM AND MINING LEASE 37

LIARD MINING DIVISION, BRITISH COLUMBIA

NTS: 104B /11E LATITUDE: 56° 41' LONGITUDE: 131° 05'

OWNED BY: PRIME RESOURCES GROUP INC.

WORK PERFORMED BY: HOMESTAKE CANADA INC. P.O. Box 11115 1055 West Georgia St., Suite 1100 Vancouver, B.C. (604) 684-2345

FOR: PRIME RESOURCES GROUP INC.

> Submitted by: D.L. Kuran J.G. Moors B.L. Traub

November 14, 1997





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1.0 Introduction

In an effort to add to the reserves of the Snip deposit, Homestake Canada Inc., on behalf of Prime Resources Group Inc., initiated an assessment and evaluation of the Jim 1 Claim and Mining Lease 37 for Snip style mineralisation: mesothermal high grade (approx. 28 gpt) shear hosted quartz-carbonate-biotite-chlorite-sulphide vein systems.

798 metres of diamond drilling was targeted with results received from an earlier grass roots exploration program consisting of geological mapping, soil geochemistry, and follow-up trenching. The focus of exploration was directed at areas of the property have seen limited work to date.

1.1 Location and Access

The Snip property is located within the Liard Mining division on the 104B/11E NTS map sheet in northwestern BC (figure 1).

Access is from Smithers (320 km southeast), Terrace (280 km south- southeast) or Wrangell, Alaska (80 km west) by fixed-wing aircraft to the 1600 metre long Bronson Airstrip.

From the airstrip, the property is accessible by drill roads constructed between 1993 and 1996, and by helicopter to various drillpads and helipads constructed between 1991 and 1997.

1.2 Property Description

The Jim 1 claim consists of 20 contiguous units totaling 500 hectares (figure 2). Mining Lease 37 contains lots 7017 through 7020, the Snip 1 and 2 claims, and the Snip 4 and 5 claims totaling 482.02 hectares. The status of the claim and mining lease is summarised in Table 1.

TABLE 1: Claim Status

Record	Claim	<u>Units</u>	<u>Area</u>	Record	Expiry
Number	<u>Name</u>		<u>(ha)</u>	Date	Date
300552	Jim 1	20	500	1986.07.22	2002.07.22
226132	M.L. 37	~	482.02	1989.07.21	2019.07.21

1.3 Physiography and Climate

The majority of the Jim 1 Claim consists of flat swampy terrain surrounding Sky Creek and Cub Pond in the southern and north central portions of the property, respectively. Vegetation predominantly consists of mixed conifers, alder, and devil's club.

Annual precipitation is between 200 and 400 centimetres and winters see heavy

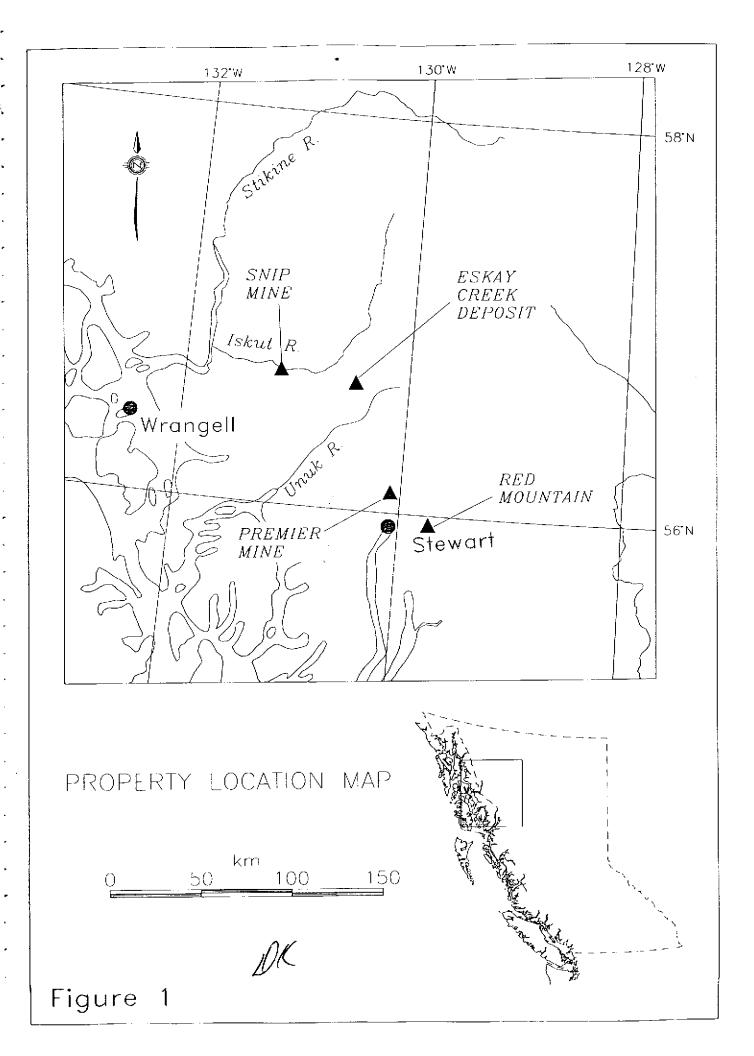
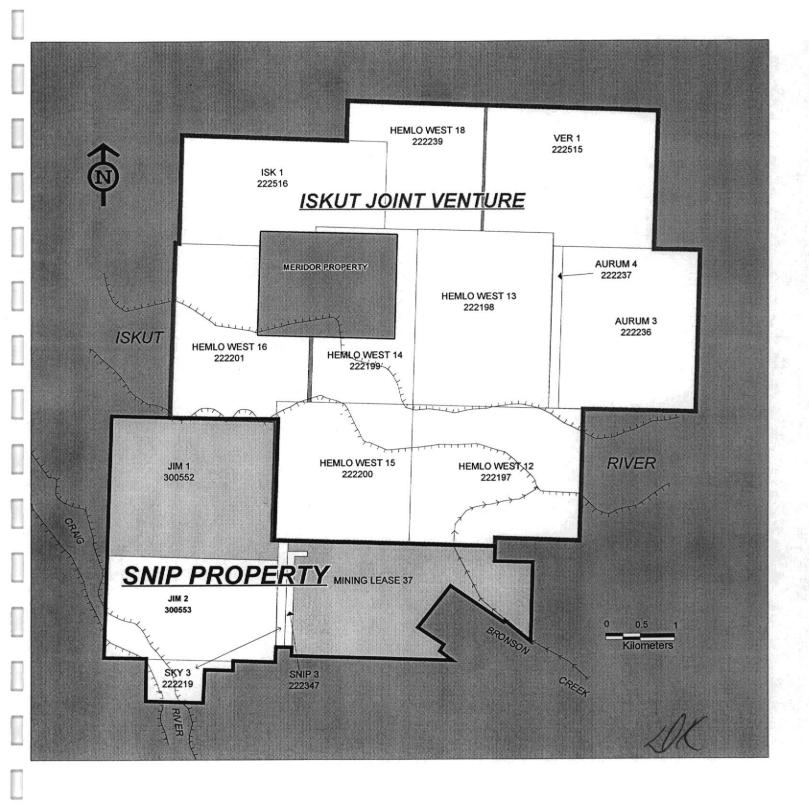


FIGURE 2: CLAIM LOCATION MAP



accumulations of snow attaining thicknesses of 2 metres. Temperatures are generally temperate.

1.4 Previous Work

<u>141 Jim1</u>

The Jim 1 claim was staked in 1986 by Cominco Ltd. in conjunction with the Snip deposit exploration.

In 1987 geologic mapping and 124 B horizon contour soil samples were taken.

No further work was performed on the property until 1991. During the 1991 field season a total of 43.54 kilometres of survey grid was cut. Following grid construction 57.2 km of UTEM (University of Toronto Electro-magnetic) surveying, 43.3 km of magnetics surveying, 7.9 km of HLEM (Horizontal Loop Electro-magnetic) surveying, and geologic mapping was conducted. The UTEM survey outlined three sub-parallel anomalous trends; two in the central portion of the claim and one in the southwest, all of which could be traced across survey lines for several hundred metres. A moderate magnetic high in the south eastern portion of the property was identified by the magnetic survey.

The 1992 program consisted of 5.4 kilometres of IP (Induced Polarization) surveying, geological mapping, and diamond drilling. Two BQTK diamond drill holes totaling 420.3 metres were drilled targeting IP geophysical anomalies.

The IP survey carried out in 1992 was extended both west and north along the existing cut grid (1991).

During the 1993 field season a total of 20.44 kilometres of IP survey was completed and three BQTK diamond drill holes totaling 892.7 metres were drilled.

No further work was performed on the property until 1996. The 1996 program consisted of a ten metre by one hundred metre soil geochemical survey, and 1:2,500 scale geological mapping. A total of 538 B horizon soil samples and 21 rock samples were collected.

The 1997 program focused on continuing the 1996 soil survey west and north. A total of 397 B horizon soil samples were collected.

1.4.2 Mining Lease 37 (Snip 1 & 2, Snip 4 & 5)

Prior to the 1997 field season little surface exploration was completed on the northeastern portion of the property north of Bronson Creek.

During the 1996 field season a total of six rock samples were collected during two days of prospecting on the Bronson North area.

During the 1997 field season a total of 5985 metres of survey grid was cut. Following grid construction, 1:2,500 geologic mapping, a ten metre by one hundred metre soil geochemical survey, hand trenching, and diamond drilling was completed. A total of thirteen rock samples and

479 B horizon soil samples were collected. Six BQTK diamond drill holes totaling 1979 metres were drilled from six helicopter accessible setups.

1.5 Geology

1.5.1 District and Local Geology

Adapted from Britton, Fletcher and Alldrick, 1990

The regional geological setting is within the Stikine Terrane, on the western edge of the Intermontane tectonic belt. Four tectonostratigraphic assemblages, bounded by unconformities are found in the 104B map area.

Three of the assemblages (excluding Bowser sediments) are represented in the area. Most strata are Upper Triassic to Lower Jurassic volcano-sedimentary arc-complex lithologies characterized by rapid facies changes. Strata have been cut by a variety of plutons representing at least four intrusive episodes spanning Late Triassic to Quaternary time. These included synvolcanic plugs, sills and stocks, minor dyke swarms, isolated dykes and sills, as well as the batholithic Coast plutonic complex. The stratigraphic sequence has been folded, faulted and metamorphosed mainly during Cretaceous time, but some Palaeozoic strata are polydeformed and probably record an earlier deformational event. Contacts between lithostratigraphic sequences within the area are not well exposed: commonly they are covered with moraine, disrupted by faults, or invaded by large intrusions such as the Lehto batholith and Coast plutonic complex.

Palaeozoic: The Palaeozoic Stikine assemblage is observed in outcrop west of the Craig River and northeast of Mount Verrett. Rocks tentatively assigned include abundant fine-grained, thinly layered, biotite-rich quartzofeldspathic gneiss, phyllite, metawacke, metatuff and thin recrystallised limestone (marble). The gneisses were probably derived from tuffaceous siltstones and sandstones, with minor ash and crystal tuffs, and are the most structurally complex in the area: two phases of penetrative deformation have been observed. The contact between Palaeozoic rocks and overlying Mesozoic strata is probably an unconformity, based on relative states of deformation.

Mesozoic: Most of the stratified rocks in the area are Mesozoic. Strata form a thick (3 kilometres) sequence of mixed volcanic and sedimentary rocks. Facies changes, minor unconformities and the paucity of distinctive marker horizons make stratigraphic correlation difficult. Extrusive rocks are mostly volcaniclastic: pyroclastic units with derived epiclastic facies. Plagioclase, pyroxene and hornblende are common phenocrysts; distinctive coarse potassium feldspar is minor but important. Compositions range from basalt to rhyodacite, but most are andesite to dacite. Sedimentary rocks are volcanic-derived siltstone, wacke and conglomerate with minor amounts of limestone, either as relatively pure lenses or as calcareous mudstones. Limestone decreases upwards in the section and is rare in Hazelton strata.

Upper Triassic: Most of the volcanic rock in the Triassic succession is basaltic to andesitic with plagioclase and pyroxene as the principal phenocrysts, characteristic of the Stuhini Group. Pyroclastic units are more common than flows, but many outcrops are massive and difficult to classify. For example, a thick, monotonous sequence of finegrained, medium to dark green, feldspar porphyry andesite underlies the lower slopes of Mount Verrett and extends across the Iskut River to Bug Lake. These rocks are moderately to completely recrystallized north of the Iskut and could be either massive crystal tuffs or flows. There area some lapilli tuffs and tuff breccias around Bug Lake, but fragmental textures are generally absent.

Triassic sedimentary rocks are mostly siltstone with minor fine-grained wacke. Thin rhythmic bedding is common. In the north they are interbedded mudstone, lithic wacke, feldspathic wacke, minor conglomerate and limestone lenses, with locally abundant fine-grained volcaniclastic material;ash tuff or volcanic sandstone. These rocks host the Snip deposit and other prospects uphill from Bug Lake and on lower Bronson Creek. A sequence of light grey-green, waxy, dacitic pyroxene-plagioclase crystal and lapilli tuffs has been identified only on Winslow Ridge and appears to be conformable within the thick sedimentary sequence.

Lower Jurassic: Jurassic strata are mainly andesitic to dacitic fragmental volcanics with minor basaltic tuffs and lesser amounts of siltstone, wacke and conglomerate. Marked lateral facies changes, lithologic heterogeneity and variable rock colours (grey, green, maroon, and mottled combinations of these) are common.

On Johnny Mountain, the Jurassic strata consists of three main units. The lower unit is a plagioclase-phyric andesitic to dacitic crystal and ash tuff, lapilli tuff and agglomerate. In some of these rocks, the plagioclase phenocrysts are rounded, suggesting they have been reworked. The middle unit conformably overlies the lower unit and consists of grey and tan dacitic volcanic rocks. They include flow-banded and welded ash tuffs as well as well-bedded ash and lapilli tuffs with rhyodacite clasts. The upper unit comprises dark grey-green, glassy, well-foliated basaltic andesite ash tuffs with minor siltstone and wacke interbeds.

On Snippaker Mountain and extending southward, the Jurassic sequence includes at least 300 metres of matrix supported, polymictic pebble to cobble conglomerate with minor siltstone and wacke interbeds. The unit grades laterally and upwards into green volcanic conglomerate and lithic lapilli tuff. These conglomerates are locally overlain by thin-bedded, salt and pepper lithic arenite and siltstone with carbonized plant remains.

Quaternary: Pleistocene and recent basaltic lava flows, cones and tephra occupy the valleys of the Iskut River, Snippaker Creek and Lava Lakes. These olivine and plagioclase phyric, often strongly vesicular flows are part of the north-trending Stikine volcanic belt of Miocene to Quaternary eruptive centres.

Intrusive Rocks: The oldest intrusives in the area are sills, dykes and plugs of hornblende diorite that are contemporaneous with Triassic host rock volcanics. They are especially common in andesites located north of the Iskut River. There is a large hornblende diorite stock of this type on the south slope of Mount Verrett. The rock is texturally similar to the andesites it intrudes and consists of mesocratic medium to dark grey, fine grained, anhedral granular diorite with fine plagioclase phenocrysts. The diorite is largely recrystallised and pervasively propylitically altered. Near its contact with the Coast batholith it has pegmatitic zones up to 50 centimetres wide by 6 metres long consisting of coarse bladed intergrowths of hornblende and plagioclase with minor biotite. Against the batholith it is migmatitic with a swirled foliated fabric in the diorite that is cut by leucogranite dykes. Contacts with andesite are indistinct and may be in part gradational.

Jurassic intrusions include synvolcanic hypabbysal stocks as well as phaneritic plutons of considerable size. Synvolcanic intrusions are thought to be comagmatic and coeval with extrusive rocks. Examples include felsite stocks on Johnny Flats and the Inel property. These are leucocratic to holofelsic, cream to tan, porphyritic rocks with fine feldspar and quartz phenocrysts set in an aphanitic groundmass. Contacts are altered and sheared but the stocks appear to form sheet-like bodies that are crudely conformable with enclosing strata.

Phaneritic intrusions of probable early Jurassic age include the Lehto batholith, the Iskut River stock and smaller plugs and dykes such as the Red Bluff porphyry. A common feature of these intrusions is the presence of coarse (up to 5 centimetres) potassium feldspar phenocrysts. The Iskut River stock consists mainly of the coarse potassium feldspar phenocrysts set in a fine to medium-grained groundmass.

The largest intrusive mass in the map area is the Coast Mountains batholith which occupies the southern quarter and northwestern corner of the map area consisting

of medium-grained biotite and biotite-hornblende granite, granodiorite and rarely quartz diorite. Very little of it has been mapped. It is distinguished from Jurassic plutons by its fresh appearance, lack of foliation and shearing, minimal saussuritization and abundance of quartz. Biotite is either the sole mafic mineral or else is much more common than hornblende. There is little or no hydrothermal alteration of skarn developed along the intrusive contacts despite the presence of limestone units in the Palaeozoic country rocks. The age of these rocks is probably middle Eocene based on potassium-argon dating near Stewart.

Isolated dykes and minor dyke swarms occur locally in the area. In addition to local feeder dykes associated with the overlying volcanics, widespread biotite and hornblende lamprophyre dykes cut all other rock types including the Coast Mountains batholith. They are typically isolated and narrow (up to 2 metres wide). The age of these dykes is probably Oligocene.

Structure: Palaeozoic rocks exhibit the strongest deformation. Folds range from crenulations through upright chevrons to recumbent isoclines with fold amplitudes of 100 metres. The largest folds plunge gently east-northeast. Crenulations and contorted open folds are also developed adjacent to faults in fine-grained sediments and tuffs of any age. These structures die out within a few metres of the fault zones.

At a regional scale the Mesozoic lithostratigraphic sequences form a flat-lying package, but Triassic and Jurassic strata show mesoscopic folds. Some of these are primarily depositional features such as convolute layering in welded tuffs, flow banding and soft-sediment slumps.

Many rocks, but especially fine-grained sediments, mafic tuffs and limestones, show intense foliation, boudinage and transposition of primary layering. Rock composition, especially mica content, largely determines the amount of foliation developed.

There is widespread sub-horizontal cleavage in most Triassic and some Jurassic rocks. Locally this is expressed in sub-horizontal faults between blocks of differing competence. An example of this is the contact between Jurassic volcaniclastic and Triassic sediments on Johnny Mountain. The underlying siltstone exhibits folding, shearing and recrystallisation that decreases in intensity away from the fault. Overlying dacitic volcaniclastic rocks which act as a competent unit also show increased strain near the fault but deformation is much weaker amounting to minor shearing and recrystallisation.

High-angle faults are common in the area and appear to post-date flat faults. Some form well-defined lineaments, traceable for kilometres and visible in radar images and air photographs. Most have small displacements on mappable faults like those seen on Johnny and Snippaker Mountains is in the order of a few hundred metres. Most faults strike northeasterly or northwesterly.

Metamorphism: Rank is generally low (ie. lower greenschist), although recrystallisation is complete. Contact metamorphism occurs within 1 to 2 kilometres of the Coast Mountains batholith. The main effects are recrystallisation with coarsening of grain size and replacement of a mafic minerals by metamorphic biotite.

1.5.2 Mineral Occurrences

Much of the area surrounding and including the Snip property has been subject to intense geological investigation due to the greater amount of mining and associated development in the area. The Snip and Stonehouse gold deposits on the lower and upper slopes of Johnny Mountain

respectively have provided for detailed studies on mineralisation in the area; eg. Rhys, 1995. This information is applicable to the abundant mineralisation that occurs over the Snip property as the target and it's associated geology are similar.

Potentially economic gold +/- silver, copper, zinc, and lead mineralisation in the Iskut region is genetically classified as: 1. Mesothermal/transitional quartz-sulphide veins (eg. Snip Twin zone, Johnny Mountain, Sulphurets West zone, Silbak-Premier); 2. Stratabound/form VMS (eg. Eskay 21B, Granduc, Big Missouri, Black Dog, SMC zones),and; 3. Alkaline Porphyry systems (eg. Galore Creek, Copper Canyon, Kerr, Sulphurets)

There is evidence that the Snip property is prospective for two of these styles of gold mineralisation, low grade Gold porphyry and mesothermal veins, however the most attractive economic target is the Snip vein style. This deposit type has dictated the exploration methodology since the realization of Snip's merits in 1986.

The following summary of the Snip mine "Twin zone" is taken from Rhys, 1995.

The Twin zone shear-vein system strikes 120° and dips 30° to 60° southwest. It is the largest of many shear veins in the Snip mine. Thickness ranges up to a maximum of 13 m, but averages approximately 2.5 m. In the eastern and lowest parts of the mine, the zone becomes a series of discontinuous pyrite veins and veinlets. Several smaller en echelon shear veins occur below its lower termination. The Red Bluff porphyry occurs north-east of , and has a parallel strike to, the Twin zone. Distance to the porphyry varies with elevation, but averages approximately 600 m.

The Twin zone has a pronounced internal layering of several vein types. Veins of calcite-chlorite-biotite, which typically contain 15 g/t Au to 40 g/t Au, comprise approximately 60% of the zone. They are commonly compositionally layered with alternating laminae of schistose chlorite-biotite and calcite. Dilatant pyrite-pyrrhotite and quartz veins, typically grading > 60 g/t Au, form discrete foliation-parallel veins, and occur independently of or within a matrix of the other ore types. Chlorite-biotite and carbonate ore types display progressive alteration sequences suggesting that they were formed by a combination of both replacement and dilation. Alteration envelopes of black biotite 0.5 cm to 2 cm wide surround many veins, internal to an outer bleached K-feldspar-calcite-quartz-sericite envelope. Biotite-rich veins and sulphide veins, common in the lowermost and eastern parts of the zone, have elevated copper grades (0.15% to 0.5% Cu). Chloritic veins are most abundant in the western and uppermost portions of the orebody and are associated with the highest molybdenum grades (0.01% to 0.05% Mo). Coarse visible gold commonly occurs with molybdenite in chlorite-rich veins.

Structures internal to the Twin zone suggest it formed as a dilatant shear zone with a predominantly normal sense of movement. Down-dip verging folds, sheath folds, synthetic shear bands, asymmetric augen, rotated quartz and pyrite porphyroclasts, and oblique subhorizontal foliations are common to the Twin zone and other shear zones in the Snip workings. They indicate an oblique normally-directed shear sense parallel to an oblique southwesterly plunging lineation developed on foliation surfaces. This slip direction is parallel to the shear sense indicated on shear zones in the Red Bluff porphyry. Deformation is localised and is confined to the southwest-dipping phyllitic and schistose foliation within the shear zones. The fabrics suggest that deformation was accomplished without loss of cohesion at a mesoscopic scale and predominantly by semi-brittle processes. The occurrence of both deformed and undeformed quartz veins suggest several generations of syntectonic quartz and sulphide veining during formation of the Twin zone.

1.5.3 Property Geology

The Jim 1 claim and Bronson North is composed predominantly of metasediments of volcaniclastic origin consisting of arkosic greywacke, siltstone, and local fine grained siliceous rocks, believed to be tuffaceous (Holroyd, 1993).

On the Jim 1 claim outcrop is best exposed on topographic highs in the eastern and northwestern portions of the property. The northern portion of the property is underlain by fine to medium grained massive greywacke with interbedded laminated siltstone. A moderate foliation oriented (130° / 65°) is best defined within the greywacke north of Boundary Pond. The northeastern portion of the property contains a K-feldspar megacrystic porphyry, which is part of the large dioritic Bronson Creek Stock to the east.

Three distinct alteration assemblages are found across both the Jim 1 claim and Bronson North. These are common across the Johnny Mountain/Bronson Creek area and are interpreted as primarily a direct result of porphyry emplacement. The first is a silica dominated quartz-sericite-pyrite alteration (QSP) which overprints lithological and structural features. Stockwork quartz veinlets with up to 2% disseminated pyrite are commonly associated with this alteration type. The second type is a pervasive fine grained chlorite alteration. The final alteration unit that occurs is pervasive very fine grained medium to deep brown biotite. Chlorite also occurs intimately with biotite. The adjacent Red Bluff Porphyry to Bronson North results in a more intense alteration assemblage as compared to those on the Jim 1 claim.

Mineralisation on the Jim 1 claim consists of fine grained disseminated pyrite that is commonly found in trace amounts over most of the property. Stockwork pyrite, with <u>+</u> molybdenite and magnetite is associated with the stockwork quartz veining within the QSP altered area north of Sky Creek on the Jim 1 claim and adjacent to the Red Bluff Porphyry on Bronson North. Narrow (10 to 25 centimetre) quartz-carbonate veins containing sphalerite, galena, pyrite, and chalcopyrite in varying amounts, and chlorite-magnetite-carbonate veining occur within areas of intense chlorite alteration north of Bronson Creek.

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2.0 1997 Exploration Program

2.1. Summary

Four BQTK diamond drill holes totaling 798 metres were drilled from four helicopter accessible setups. Three holes (597 metres) from three setups were drilled on the Jim 1 claim northwest of Boundary Pond (map 1), and one hole (201 metres) from one setup was drilled on Mining Lease 37 (Snip 1 & 2, Snip 4 & 5 claims) north of Bronson Creek (map 2). Plan and section views of holes on each set-up are provided in maps 3 and 4 respectively. Drill hole collar and attitude information are provided in Table 2. Drill logs are provided in Appendix A. Assay Results are provided in Appendix B. Assay Certificates are provided in Appendix C.

A total of 237 samples were sent to Snip Mine laboratory for five element analysis.

2.2 Diamond Drilling - North Bronson

2.2.1 Review

Hole S-275 was collared 135 metres grid west (300°) of diamond drill hole S-269 which contains a 1.2 metre biotite-carbonate-pyrite shear vein containing 6 g/t gold.

2.2.2 Results - DDH S-275

2.2.2.1 Lithology and Structure

A medium to coarse grained massive greywacke comprises hole S-275. Coarser sections of the greywacke contain sub-rounded chert fragments between 7.3 and 50.5 metres. A fifty centimetre weakly magnetic lamprophyre dyke cross cuts the greywacke at eleven metres down hole.

A weak foliation at 45 to 50° to core axis occurs from 109.5 metres to the end of hole at 201.2 metres.

2.2.2.2 Alteration

Three distinct alteration assemblages compose hole S-275: quartz-sericite-pyrite (QSP); pervasive fine grained chlorite with trace biotite, and; selective epidote alteration of iron-carbonate clasts and veins.

The upper 50.5 metres consists of moderate QSP alteration with trace pyrite. Moderate chlorite alteration overprints QSP and rarely exceeds five metres in true thickness. From 50.5 to 109.5 metres alteration occurs locally as either chlorite with minor biotite or as patchy epidote. The remainder of the hole contains moderate pervasive chlorite alteration, which is overprinted by

Table 2

Jim 1 Claim and Mining Lease 37 Diamond Drilling, 1997 Hole Collar Summary Information

TARGET	HOLE	EASTING 1 (mine grid)		ELEVATION (m)	<u>AZIMUTH</u> (true)	DIP	<u>LENGTH</u> (ft)	(m)	<u>Target Total</u> (m)
North Bronson	S- 275	5465	3350	190	30	-45	660	201	201
Jim West	S- 276 S- 277 S- 278	1630 1470 1425	2510 2040 1820	105 110 105	50 50 10	-45 -45 -45	500 960 500	152 293 152	597

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moderate to strong epidote alteration. Epidote altered sections are restricted to areas containing iron-carbonate clasts and veins, the largest section occurring between 121.5 metres and 150 metres.

2.2.2.3 Veining and Mineralisation

Trace disseminated pyrite occurs throughout QSP altered sections and locally up to 7% in narrow quartz-carbonate-chlorite shear veins. Trace to 1% pyrrhotite occurs intimately with pyrite in shear veining. A total of 65 samples of prospective veining, alteration, and/or sulphide mineralisation were split and sent to Snip Mine Lab for analysis. No significant gold values were intersected by hole S-275.

2.3 Diamond Drilling - Jim West

2.3.1 Review

Drilling was prompted by both the results from geological mapping and soil geochemistry and the presence of extensive clay cover over the area hindering standard advanced surface exploration methods. A total of three holes S-276, S-277, and S-278 were drilled from three helicopter setups.

2.3.2 Results: DDH S-276

2.3.2.1 Lithology and Structure

A diorite porphyry with one centimetre by two centimetre K-feldspar phenocrysts was intersected from 6 metres to 49.7 metres. Beyond the porphyry is a fine to medium grained massive arkosic greywacke with 13.7 metres of interbedded siltstone commencing at 65.5 metres.

A moderate to strong foliation occurs throughout the diorite porphyry increasing from 55° to core axis at 15.5 metres to 80° to core axis at 30.5 metres and returning to 55° to core axis at the lower contact. The lower 102.7 metres of greywacke below the diorite porphyry contains a weak to moderate foliation at 50° to core axis increasing slightly downhole to 80° to core axis.

2.3.2.2 Alteration

Two distinct alteration assemblages occur throughout hole S-276. The first assemblage consists of quartz-sericite-pyrite alteration (QSP) and the second is a pervasive fine grained biotite-chlorite alteration.

The QSP alteration occurs throughout the entire length of the diorite porphyry and as local pods between 65.5 metres and 79.2 metres and between 129 metres and 138 metres. The remainder of the hole contains moderate pervasive biotite-chlorite alteration.

2.3.2.3 Veining and Mineralisation

An eight metre shear zone at 107 metres contains twenty to thirty percent quartzcarbonate veins at 80° to core axis, and 2 to 4% pyrite, 3 to 4% sphalerite, and 1% chalcopyrite between 113.2 metres and 114 metres.

No significant gold values were intersected by hole S-276 from 29 samples of prospective veining, alteration, and/or sulphide mineralisation.

2.3.3 Results - DDH S-277

2.3.3.1 Lithology and Structure

Underlying 24 metres of overburden, hole S-277 intersected fine to medium grained massive greywacke similar to that in hole S-276. Three weakly magnetic lamprophyre dykes less than 0.5 metres thick, cross cut the greywacke at 95.4 metres, 229.3 metres, and 231.7 metres. Intermediate dykes of a similar thickness were intersected at 99 metres and 100.5 metres respectively.

A moderate foliation throughout the greywacke at 40 to 55° to core axis increases to 80° to 90° in a strongly foliated section between 110.6 and 125 metres.

2.3.3.2 Alteration

Pervasive fine grained chlorite alteration with local patches of epidote and pervasive fine grained biotite alteration was intersected by hole S-277. The upper 110.6 metres of the hole consists of propylitic porphyry associated alteration consisting of pervasive chlorite with moderate epidote alteration between 30 and 76 metres. The lower 182 metres contains pervasive biotite alteration with minor chlorite.

2.3.3.3 Veining and Mineralisation

97 samples of prospective veining, alteration, and/or sulphide mineralisation from hole S-277 returned 3.05 g/t gold over 0.3 metres at 252.7 metres and 3.45 g/t gold over 0.5 metres at 258 metres. These intervals contain trace to 3% disseminated pyrite parallel to foliation.

2.3.4 Results - DDH S-278

2.3.4.1 Lithology and Structure

A medium grained massive greywacke and lamprophyre dykes comprises hole S-278. Weakly magnetic and strong chlorite altered lamprophyre dykes with chilled margins vary from 0.5 to 2 metres thick and occur throughout the hole.

Moderately developed foliation within the greywacke varies from 45° to 60° to core axis

throughout the hole. A strong foliation is associated with sections of intense fine grained pervasive chlorite and biotite alteration.

2.3.4.2 Alteration

Four alteration assemblages were intersected by hole S-278: pervasive fine grained biotite; pervasive chlorite; patchy quartz-sericite-pyrite (QSP), and; selective epidote alteration of iron-carbonate clasts and veinlets.

Pervasive biotite alteration in the uppermost 36.3 metres continues to 75 metres as biotite-chlorite alteration. Both QSP and epidote alteration obliterates the biotite-chlorite alteration from 54 to 57 metres and from 62 to 70 metres respectively. The lower 77.4 metres contains moderate pervasive chlorite alteration with local patches of QSP, biotite, and epidote alteration.

2.3.4.3 Veining and Mineralisation

Minor trace to 3% disseminated pyrite occurs locally within areas of moderate to intense shear veining. Fourty-two samples from hole S-278 produced no significant gold values.

3.0 Recommendations

3.1 Jim West

The program initiated during the 1997 field season adequately tested this portion of the Jim 1 claim and failed to intersect gold bearing mineralised shear veins of considerable width. No further drilling is recommended at this time.

3.2 North Bronson

North Bronson drilling targeted the small portion of Mining Lease 37 located north of Bronson Creek. Hole S-275 adequately tested the area between Bronson Creek and the softmern property boundary. Insignificant gold values were intersected and no further drilling Browner Province D. L. D. L. D. L.

Prime Resources Group Inc./Homestake Canada Inc. Jim 1 Claim and Mining Lease 37 Exploration Program 1997 Statement of Costs July 23 to August 30, 1997

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Personnel				<u>July (days)</u>	A	ugust (days)		
Huggins	Geologist			7.3		- 14		
Moors	Project Geologist			2.75		9		
Traub	Geologist			5.3		14		
Tutt	Labourer			0		9		
	Mandays/ mo.			15.35		46	•	
	·····		\$	3,133.00	\$	9,264.00	\$	12,397.00
Line Cutting								
		Crew Days		Rate				
MFH Contracting		10	\$	575.00			\$	5,750.00
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Snip Mine Assay Lab:	core samples	Samples		Cost				
Analysis		237	\$	11.00			\$	2,607.00
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		hours		wage		sub-total		
Sample bucker 1		3	\$	23.55	\$	70.65		
Sample bucker 2		2	\$	26.48	\$	52.96	\$	123.61
Cample Backer 2		-	•	20.10	•		<u> </u>	
Helicopter Support		hours		Rate				
Northern Mountain Heli.		31.4	\$	800.00			\$	25,120.00
			•					
<u>Airfare/freight</u>		<u>Trips</u>		Rate				
Smithers-Bronson	Charter (drillers)	1	\$	1,200.00				
Vancouver-Smithers	Commercial	2	\$	371.00			\$	1,942.00
Field Supplies								
Various							\$	450.00
Diamond Drilling				Rate				
Drilling	metres	798	\$	41.50			\$	33,117.00
Moving, Standby,testing		88	\$	30.00			\$	2,640.00
	drill hours		\$	23.60			\$	295.00
Materials	rods	10	\$	60.50			\$	605.00
Mob/Demob	1000		\$	2,000.00			\$	2,000.00
ATV (1) rental	days	1	\$	70.00			\$	70.00
Core Boxes	44,0	142	\$	6.50			\$	923.00
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Accommodation	Mandays	<u>cost/day</u>	
National Caterers	96.7	30.92	\$ 2,988,96
(Snip Mine site)			PROVIDE

Total: \$ SCIEN

Statement of Qualifications

I, David L. Kuran of 25630 Bosonworth Avenue, in the Municipality of Maple Ridge, British Columbia, do hereby certify that:

- 1. I am a graduate of the University of Manitoba (1978) and hold a B.Sc. in Geology.
- 2. I am a fellow of the Geological Association of Canada.
- 3. 1 am a member in good standing of the Association of Professional Engineers and Geoscientists of the Province of British Columbia.
- 4. I have been employed in my profession as an Exploration Geologist in Canada, USA, and Mexico since graduation.
- 5. I am presently employed by Homestake Canada Inc. of 1100-1055 West Georgia Street, Vancouver, British Columbia as Senior Geologist.
- 6. I supervised the planning and implementation of the work described in this report, was in communication with the geologists on site, conducted periodic site visits and was involved in the data interpretation and the editing of this 1997 Jim 1 Claim, Mining Lease 37 Assessment Report.
- I consent to the use of this report concerning the 1997 exploration program on the Jim 1 Claim, Mining Lease 37, owned by Prime Resources Group Inc. in the Liard Mining Division, NTS 104B 11W, for all corporate purposes relating to Homestake Canada Inc. and Prime Resources Group Inc.

Signed at Vancouver, British Columbia, on this, the 15th day of November, 1997.



Statement of Qualifications

I, James Gregory Moors of 3375 Ontario St., Vancouver, British Columbia state that:

1. I am a 1988 graduate of the University of Waterloo, Waterloo, Ontario with a B.Sc (Hons) in Earth Sciences.

2. I have been employed in mineral exploration prior to my graduation and have been practicing my profession since 1988.

3. I am presently on contract as a Project Geologist with Homestake Canada Inc., 1100-1055 West Georgia, Vancouver, British Columbia.

4. I supervised and performed planning, implementation, and interpretation of the work described in this 1997 Jim 1 Claim and Mining Lease 37 Assessment Report.

5. I consent to the use of this report concerning the 1997 exploration program carried out on the Jim 1 mineral claim and Mining Lease 37 owned by Prime Resources Group Inc. in the Liard Mining Division, NTS 104B/11E for all corporate purposes relating to Homestake Canada Inc. and Prime Resources Group Inc..

Signed at Vancouver, British Columbia, on this the 14th day of November, 1997.

James G. Moors, B.Sc (Hons)

Statement of Qualifications

I, Brian L. Traub of 4212 Whitemud Road, Edmonton, Alberta, do hereby certify that:

- 1. I graduated in 1996 with a Bachelor of Science Degree, Honours in Geology from the University of Alberta, Edmonton, Alberta.
- 2. I have been employed in my profession as an Exploration Geologist in Canada since graduation.
- 3. I am presently employed by Homestake Canada Inc. of 1100-1055 West Georgia Street, Vancouver, British Columbia as Geologist.
- 4. That I personally performed or supervised the work referenced in this report and was on the property between July 20th to September 6th, 1997.
- 5. I consent to the use of this report concerning the 1997 exploration program carried out on the Jim 1 mineral claim and Mining Lease 37 owned by Prime Resources Group Inc., in the Liard Mining Division, NTS 104B 10E, for all corporate purposes relating to Homestake Canada Inc. and Prime Resources Group Inc..

Signed at Vancouver, British Columbia, on this, the 7th day of November, 1997.

BiZTA

Brian L. Traub, B.Sc. (Hons)

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APPENDIX A

Diamond Drill Core Logs

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CARD NO .: CORE SIZE LENGTH: DATE: COMMENCED RECOVERY: LOG BY: HOLE # ²/2 GORK 5-277 COMPLETED CBH CODED7 GEORE57 CORRECTED DIPS: ASSAY? COLLAR DIP: AREA: COLLAR ELEVATION: STICK UP INGE 450 MINERALS CONTAL CONT 1 SHP DIP METERS SUB ALTER, REMARKS CONTAC VENS METERS MAJOR SNIP MINE: DRILL HOLE RECORD MAJOR HE FROM то ю FROM FORATION IN FINE AGAINST SETTS C 20 - - PY 55 3 FZ. C.W -81.85 95.4 76 1873 DBY STEND FRE C 35-46 + 58 1200 OC Min where the AL AV ALTOS : 14/ 15 FR 84.4 73 3491 - MARTING FRIDATE WITH FILLATIN DECOTING OWNER F 91954 MASSIVE MIND. COR OLARGO GN. 954-95-7 LAMP OYKE -HEARLY MEG. 95.4 157 **** 15.7-10 CREAMAGE is MASSING BOT MERCHY BID MUTCHED & MUCH MAD 6 Ja 55.7 111 99-57? OYKE Ton your colore of tem chill MADECKY, 1-22- 01AG2 BANDS 92 975 Ωγ<u>k</u>r` 1005-1009 - SAME OVER MS ADADE DYRE ALL F 99. 54005 KETLIEN DYRES -STRONGLY FRATURED WITH OL PILL 1039-110.4 MADDINE - MOD CAR PURCHED FINT TO MED. GREYMALETS. 114 101 110H-1106 MARK PARLE - 70 an ma buse of Pault Parts C- 75 10.0 110-6-31 INTEREDUCTLY BEAN SAMET IS INTERSELY STATERED SILTS PONT WITH STRENH ENTE SHEN 10.6 125 UP TO 30-447. OL WINS " PREAA CORD. MYO OC ALT'N cm_ 110.4 MOST OF THE FILMTION IS VIENT TONIALLOR "SHIRLY POUL" ANT ALSO MOR 27.5 57K/ SULPHIPE MULESAN POTTON IN Sit 11.6 THE COLOR IS VORIDELE FROM DATER GATY/OLTON TO PRIMA THIS ALOUR MEANING die 102 725-5 1) ASJUMED NO RECEPTION OFFECTION ALTERATION - 16 CHL - AN SN 124 1575 mapping of engran 11 (-20-91 Out same MADS come as a 1377155.5 6/1 <u>In AL 50</u> б*і*я 139.2 141 1170-119 KOLE LIPE TO LA. (0) Fa + 450 49.2 MS-1275 LESS STREWELV TITERED + PRIMARE THAN REST OF ALEA 134 m 90 6 11 SAT FA 94 BES NEAT THEO N (M) 54 6 -157 2 FO. 970- 0- -0" 143-152 "Febra Cas" special 1-3mm de veins. ~/ Fa. 960" 15/1-1526 LIGHT ORITE MARLY STRERTT MID STR / RID ATO Ser 15% ية، ص ا جسی 11.10

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APPENDIX B

Assay Results, Diamond Drill Core

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From	То	Sample	Width	Gold	Gold	Silver	Copper	Lead	Zinc
		number		(Au)	(Au)	(A <u>g</u>)	(Cu)	(Pb)	(Zn)
(m)	(m)		<u>(m)</u>	(ppb)	g per T	(ppm)	(ppm)	(ppm)	(ppm)
8.8	9.3	35257	0.5	398	~	0.6	411	144	16
16.7	17.5	35258	0.8	121	~	0.1	152	76	6
21.0	21.8	35259	0.8	87	~	0.3	211	75	8
21.8	22.4	35260	0.6	144	~	1	798	79	4
22.35	22.8	35261	0.45	82	~	0.5	225	49	4
22.8	23.5	35262	0.7	65	~	0.4	314	72	5
29.1	29.6	35263	0.5	134	~	0.3	68	78	4
35.2	35.8	35264	0.6	423	~	0.4	131	61	5
35.8	36.5	35265	0.7	126	~	0.3	143	59	4
36.5	37.0	35266	0.5	42	~	0.4	163	62	_5
41.5	42.0	35267	0.5	51	~	0.6	136	73	6
42.0	42.5	35268	0.5	317	~	19.2	209	109	102
44.2	44.8	35269	0.6	344	~	4.7	379	274	105
44.8	45.5	35270	0.7	66	~	0.5	83	115	6
56.2	56.8	35271	0.6	25	~	1.4	58	56	6
56.8	57.2	35272	0.4	165	~	0.7	81	102	5
57.2	57.9	35273	0.7	13	~	0.2	66	54	5
66.5	67.1	35274	0.6	73	~	0.3	163	87	7
68.4	69.1	35275	0.7	50	~	0.8	124	68	10
77.3	77.9	35276	0.6	158	~	0.4	107	113	8
77.9	78.5	35277	0.6	107	~	0.4	105	108	9
78.5	79.0	35278	0.5	138	~	0.2	127	93	11
79.0	79.5	35279	0.5	193		0.5	74	62	6
89.5	90.2	35280	0.7	80	~	1.4	135	63	11
90.2	90.5	35281	0.3	208		1.1	503	61	7
90.5	91.0	35282	0.5	220	~	1.2	380	50	6
91.0	91.5	35283	0.5	96	~	0.3	211	48	6
97.6	98.3	35284	0.7	51	<u> </u>	01	80	52	6
98.3	98.7	35285	0.4	38	~	0.1	58	99	4
98.7	99.7	35286	1.0	83	<u>~</u>	0.4	48	58	4
99.7	100.2	35287	0.5	48		0.4	72	134	5
100.2	100.8	35288	0.6	32	~	0.1	67	86	5
103.0	103.5	35289	0.5		~	0.3	246	38	6
111.0	111.8	35290	0.8	32	<u> </u>	0.5	90	52	7
117.0	117.5	35291	0.5	87	~	0.7	149	113	9
118.2	118.8	35292	0.6	115	~	1.2	334	97	9
121.5	122.0	35293	0.5	228	~	3.4	1472	111	4
122.0	122.5	35294	0.5	126		1.4	565	71	6
122.5	123.0	35295	0.5	118		0.9	369	82	
123.0	123.6	35296	0.6	92	~	0.8	342	151	42
134.0	134.4	35297	0.4	46	~	0.4	98	55	4

From	То	Sample	Width	Gold	Gold	Silver	Copper	Lead	Zinc
		number		(Au)	(Au)	(Ag)	(Cu)	(Pb)	(Zn)
(m)	(m)		(m)	(ppb)	g per ⊤	(ppm)	(ppm)	(ppm)	(ppm)
138.0	138.6	35298	0.6	10	~	0.5	61	23	3
138.6	139.1	35299	0.5	93	~	0.3	156	57	5
139.1	139.8	35300	0.7	29	~	0.5	111	54	4
142.0	142.5	35301	0.5	190	~	1.6	612	49	8
147.8	148.5	35302	0.7	219	~	1	150	79	12
149.3	150.2	35303	0.9	139	~	1.6	310	27 9	317
150.2	150.5	35304	0.3	4711	1.5	3.7	219	170	654
150.5	151.0	35305	0.5	80	~	0.7	143	64	24
153.6	154.1	35306	0.5	1020	0.4	3.5	734	153	30
154.1	154.8	35307	0.7	150	~	2.3	266	123	2028
154.8	155.5	35308	0.7	52	~	0.5	125	78	11
162.5	163.1	35309	0.6	108	~	0.1	180	111	11
164.0	164.5	35310	0.5	159	~	1	433	36	7
164.5	164.9	35311	0.4	105	~	0.5	154	104_	6
164.9	165.7	35312	0.8	992	~	0.9	490	78	6
176.0	176.6	35313	0.6	101	~	0.1	221	47	4
176.6	177.1	35314	0.5	97	~	0.6	397	51	5
177.1	177.6	35315	0.5	51	~	0.2	266	57	4
179.5	180.3	35316	0.8	53	~	0.6	132	39	6
181.1	181.8	35317	0.7	78	~	0.8	149	94	5
184.1	184.6	35318	0.5	56	~	0.3	180	87	9
184.6	185.2	35319	0.6	190	~	1.5	414	233	43
185.2	185.9	35320	0.7	80	~	0.1	192	84	10
197.9	198.4	35321	0.5	57	~	0.4	251	651	98

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From	Τo	Sample	Width	Gold	Gold	Silver	Copper	Lead	Zinc
		number		(Au)	(Au)	(Ag)	(Cu)	(Pb)	(Zn)
(m)	(m)		(m)	(ppb)	g per T	(ppm)	(ppm)	(ppm)	(ppm)
40.6	41.3	35322	0.7	151.0	~	1	185	241	350
41.3	42.0	35323	0.7	170.0	~	1	108	221	108
42.0	42.7	35324	0.7	93.0	~	1	130	509	167
42.7	43.3	35325	0.6	46.0	~	1	114	210	33
43.3	44.8	35326	1.5	188.0	~	2	524	2540	44
44.8	45.4	35327	0.6	167.0	~	1	121	76	18
49.5	49.9	35328	0.4	146.0	~	0	188	154	50
49.9	50.3	35329	0.4	151.0	~	0	138	70	109
64.0	65.0	35330	1.0	124.0	~	0	109	52	5
76.2	77.2	35331	1.0	95.0	~	0	114	35	12
77.2	77.8	35332	0.6	102.0	~	24	1234	29650	5138
77.8	78.8	35333	1.0	104.0	~	0	86	63	36
88.0	89.1	35334	1.1	101.0	~	0	92	45	29
94.3	94.8	35335	0.5	96.0	~	0	78	34	10
102.3	103.2	35336	0.9	170.0	~	5	440	7225	1494
106.1	106.7	35337	0.6	115.0	~	0	94	96	23
108.6	109.6	35338	1.0	82.0	~	0	146	256	234
112.3	113.2	35339	0.9	102.0	~	0	116	78	20
113.2	114.0	35340	0.8	149.0	~	3	390	9850	1001
114.0	114.5	35341	0.5	147.0	~	0	132	75	21
117.5	118.0	35342	0.5	123.0	~	1	154	30	9
122.7	123.2	35343	0.5	229.0	~	1	257	71	8
140.3	141.3	35344	1.0	100.0	~	1	154	61	12
142.5	143.1	35345	0.6	135.0	~	1	93	87	18
144.0	144.5	35346	0.5	128.0	~	1	132	50	10
144.5	145.0	35347	0.5	123.0	~	0	119	30	6
145.0	145.5	35348	0.5	148.0	~	1	115	33	6
148.3	148.9	35349	0.6	142.0	~	0	112	36	7
149.9	150.5	35350	0.6	132.0	~	1	146	22	4

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From	То	Sample	Width	Gold	Gold	Silver	Copper	Lead	Zinc
Į		number		(Au)	(Au)	(Ag)	(Cu)	(Pb)	(Zn)
(m)	(m)		(m)	(ppb)	g per ⊤	(ppm)	(ppm)	(ppm)	(ppm)
24.4	25.2	35351	0.8	113	~	0.6	164	114	7
27.9	28.6	35352	0.7	202	~	5.3	637	1307	66
28.6	29.2	35353	0.6	141	~	0.9	359	89	7
29.2	30.0	35354	0.8	113	~	1.1	286	121	5
30.0	30.6	35355	0.6	119	~	0.2	114	50	4
34.0	34.4	35356	0.4	116	~	0.3	91	41	5
34.4	34.8	35357	0.4	1074	0.1	1.1	212	62	4
34.8	35.5	35358	0.7	131	~	0.5	39	39	5
35.5	35.8	35359	0.3	138	~	1.6	562	68	6
35.8	36.7	35360	0.9	125	~	1.2	354	59	3
42.3	43.0	35361	0.7	129	~	0.8	201	64	6
43.0	43.6	35362	0.6	106	~	0.1	250	90	7
43.6	44.2	35363	0.6	78	~	0.1	235	63	5
44.2	45.0	35364	0.8	436	~~	0.1	324	252	89
48.7	49.3	35365	0.6	67	~	0.1	376	94	4
49.3	49.9	35366	0.6	66	~	0.7	257	85	2
49.9	50.5	35367	0.6	68	~	1.4	445	58	3
50.8	51.6	35368	0.8	94	~	0.2	267	105	5
51.6	52.1	35369	0.5	54	~	0.9	435	136	5
52.1	53.0	35370	0.9	57	~	0.2	365	164	9
55.0	55.7	35371	0.7	39		0.4	320	60	7
55.7	56.5	35372	0.8	59	~	0.4	313	66	1
56.5	57.0	35373	0.5	54	~	0.3	537	57	1
65.5	66.1	35374	0.6	53	~	0.6	238	154	3
66.1	66.7	35375	0.6	65	~	0.4	349	138	18
66.7	67.1	35376	0.4	58	~	0.6	365	116	3
67.1	68.0	35377	0.9	77	~	0.8	470	120	4
71.3	71.8	35378	0.5	52	~	0.7	417	80	3
71.8	72.4	35379	0.6	288	~	0.9	607	209	45
72.4	73.2	35380	0.8	67	~	0.7	453	145	5
73.2	73.7	35381	0.5	75	~	0.9	958	77	4
73.7	74.5	35382	0.8	84	~	0.7	588	93	
74.5	74.8	35383	0.3	72	~	0.8	822	94	4
76.8	77.3	35384	0.5	83	~	0.7	372	100	6
77.3	77.8	35385	0.5	76	~	0.6	414	71	6
77.8	78.6	35386	0.8	79	~	0.6	402	98	5
78.6	79.2	35387	0.6	73	~	0.3	309	96	7
80.5	81.0	35388	0.5	70	~	0.7	303	442	219
81.0	81.5	35389	0.5	302	~	5	723	1351	850
81.5	82.0	35390	0.5	68	~	0.2	259	92	7
82.0	82.7	35391	0.7	92	~	0.1	250	75	4

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From	То	Sample	Width	Gold	Gold	Silver	Copper	Lead (Pb)	Zinc (Zn)
	·	number		(Au)	(Au)	(Ag)	(C <u>u)</u>		
(m)	(m)		(m)	(ppb)	g per ⊤	(ppm)	(ppm)	(ppm)	(ppm)
86.6	87.3	35392	0.7	74	~	0.4	331	70	4
87.3	87.8	35393	0.5	86	~	0.7	495	63	5
87.8	88.4	35394	0.6	114	~	0.5	271	78	5
88.4	89.0	35395	0.6	79	~	1.1	471	89	7
93.2	94.2	35396	1.0	78	~	0.7	514	69	8
99.3	100.3	35397	1.0	54	~	0.9	354	107	12
106.7	107.5	35398	0.8	58	~	0.7	299	108	8
109.7	110.4	35399	0.7	83	~	0.4	221	107	4
117.5	118.0	35400	0.5	41	~	0.1	118	50	10
118.0	118.7	35401	0.7	68	~	1.2	125	59	5
120.3	120.9	35402	0.6	65	~	0.9	149	60	12
136.6	137.6	35403	1.0	63	~	0.5	118	112	9
138.3	138.8	35404	0.5	32	~	0.7	132	82	6
138.8	139.6	35405	0.8	66	~	1.2	143	123	6
139.6	140.2	35406	0.6	52	~	1.7	182	166	20
140.2	140.6	35407	0.4	57	~	0.6	128	88	6
144.0	145.0	35408	1.0	10	~	1	166	105	4
153.7	154.5	35409	0.8	62	~	0.8	118	84	6
154.5	155.3	35410	0.8	54	~	2.5	107	1253	282
155.3	156.0	35411	0.7	73	~	1.9	101	597	168
156.0	156.7	35412	0.7	47	~	4	102	724	627
166.3	166.9	35413	0.6	21	~	1	45	232	34
166.9	167.4	35414	0.5	52	~	0.4	25	57	7
167.4	168.0	35415	0.6	28	~	1.3	36	90	7
172.6	173.4	35416	0.8	47	~	0.1	69	77	6
178.0	178.5	35417	0.5	47	~	0.6	193	80	5
178.5	179.1	35418	0.6	69	~	0.5	152	84	6
179.1	179.8	35419	0.7	59	~	0.2	147	96	6
184.6	185.2	35420	0.6	86	~	0.4	157	55	11
201.3	201.8	35421	0.5	44	~	16	126	395	133
207.6	208.0	35422	0.4	10	~	5.1	171	214	39
208.0	208.6	35423	0.6	10	~	2.5	97	103	15
208.6	209.3	35424	0.7	10	~	3.6	99	107	56
209.3	210.0	35425	0.7	21	~	2.3	66	74	15
222.5	223.5	35426	1.0	112	~	2.1	101	267	54
235.7	236.3	35427	0.6	50	~	1.5	130	160	119
248.0	249.0	35428	1.0	120	~	1.4	214	392	16
249.0	249.8	35429	0.8	230	~	3	689	151	15
249.8	250.4	35430	0.6	851		2.4	154	85	14
250.4	251.0	35431	0.6	134	~	0.8	69	80	22
251.0	252.0	35432	1.0	1094	1.15	2.2	325	245	38
252.0	252.7	35433	0.7	1241	0.9	1.6	250	37	8
252.7	253.0	35434	0.3	2394	3.05	2.4	265	140	92
255.1	255.9	35435	0.8	139	~	1.9	131	182	57
255.9	256.4	35436	0.5	229	~	2.1	138	285	53

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From	То	Sample	Width	Gold	Gold	Silver	Copper	Lead	Zinc
		number		(Au)	(Au)	(Ag)	(Cu)	(Pb)	(Zn)
(m)	(m)		(m)	(ppb)	g per T	(ppm)	(ppm)	(ppm)	(ppm)
256.4	256.5	35437	0.1	138	~	1.3	151	153	46
256.5	257.0	35438	0.5	115	~	2.2	313	1636	26
257.0	257.5	35439	0.5	1403	1.55	1.4	315	34	5
257.5	258.0	35440	0.5	934	~	1.2	229	14	2
258.0	258.5	35441	0.5	2496	3.45	1.7	212	65	9
258.5	259.0	35442	0.5	272	~	1	191	156	13
259.0	259.5	35443	0.5	58	~	0.4	59	58	7
259.5	260.0	35444	0.5	528	~	0.9	192	115	9
260.0	261.0	35445	1.0	188	~	0.9	116	97	25
271.3	272.0	35446	0.7	39	~	2	226	468	57
275.0	275.5	35447	0.5	65	~	3.2	384	143	51
275.5	276.0	35448	0.5	49	~ ~	2.2	358	271	43
276.5	277.2	35449	0.7	67	~	0.2	122	70	12
278.0	278.5	35450	0.5	74	~	0.4	139	44	6
289.5	290.0	35451	0.5	10	~ ~	0.7	203	77	4

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From	То	Sample	Width	Gold	Gold	Silver	Copper	Lead	Zinc
		number		(Au)	(Au)	(Ag)	(Cu)	(Pb)	(Zn)
(m)	(m)		(m)	(ppb)	g per T	(ppm)	(ppm)	(ppm)	(ppm)
19.5	19.9	35452	0.4	517.0	~	0	62	34	4
19.9	20.5	35453	0.6	90.0	~	1	358	55	6
20.5	21.0	35454	0.5	83.0	~	1	428	53	6
21.0	21.5	35455	0.5	29.0	~	1	222	53	4
34.4	35.0	35456	0.6	86.0	~	1	349	74	5
35.0	35.6	35457	0.6	182.0	~	1	278	107	4
35.6	36.3	35458	0.7	151.0	~	1	427	82	5
38.3	38.9	35459	0.6	59.0	~	0	108	120	3
38.9	39.3	35460	0.4	60.0	~	0	133	112	3
39.3	40.0	35461	0.7	101.0	~	0	81	134	3
40.0	40.7	35462	0.7	71.0	~	1	134	112	4
47.7	48.4	35463	0.7	68.0	~	0	124	127	5
48.4	48.8	35464	0.4	56.0	~	1	136	193	4
48.8	49.3	35465	0.5	101.0	~	1	139	167	5
49.3	50.0	35466	0.7	72.0	~	1	178	124	5
50.0	50.5	35467	0.5	89.0		0	106	145	10
50.5	51.0	35468	0.5	56.0	~	0	128	122	0
66.0	66.5	35469	0.5	53.0	~	3	297	81	18
66.5	67.0	35470	0.5	103.0	~	4	361	445	54
68.5	69.0	35471	0.5	58.0	~		345	66	2
69.0	69.5	35472	0.5	107.0	~	0	456	70	4
74.0	74.5	35473	0.5	77.0	~	0	267	119	4
74.5	75.0	35474	0.5	138.0	~	1	234	114	2
88.0	88.4	35475	0.4	38.0	~	0	148	370	
88.4	88.6	35476	0.2	165.0	~	1	592	130	7
88.6	89.3	35477	0.7	61.0	~	1	265	162	1
89.3	89.9	35478	0.6	30.0		1	320	158	3
89.9	90.5	35479	0.6	33.0	~	0	125	241	
90.5	91.2	35480	0.7	59.0	~	0	92	220	3
107.8	108.3	35481	0.5	69	~	0.3	122	398	1
108.3	108.9	35482	0.6	48	~	0.4	157	245	1
119.8	120.8	35483	1.0	71	~	0.4	139	135	6
120.8	121.3	35484	0.5	55	~	0.7	275	121	3
121.3	121.7	35485	0.4	656	~	1.8	1412	57	10
121.7	122.3	35486	0.6	101	~	0.3	238	266	5
122.3	123.0	35487	0.7	82	~	0.2	87	503	5
133.0	134.0	35488	1.0	63	~	0.8	156	177	3
134.0	135.0	35489	1.0	57	~	0.3	165	187	5
140.0	140.6	35490	0.6	75	~	0.5	156	140	8
140.6	141.1	35491	0.5	570	~	8.8	1059	977	
141.1	141.6	35492	0.5	79	~	0.8	232	108	12

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From	То	Sample number	Width	Gold (Au)	Gold (Au)	Silver (Ag)	Copper (Cu)	Lead (Pb)	Zinc (Zn)
(m)	(<u>m</u>)		(m)	(ppb)	g per ⊤	(ppm)	(ppm)	(ppm)	(ppm)
141.6	142.4	35493	0.8	127	~	1	194	101	6

APPENDIX C

Assay Certificates, Diamond Drill Core

SNIP GOLD MINE CERTIFICATE OF ASSAY

Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
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97621	35257	398.0	~	0.6	411.0	144	16
97621	35258	121.0	~	0.1	152.0	76	6
97621	35259	87.0	· ~	0.3	211.0	75	8
97621	35260	144.0	~	1.0	798.0	79	4
97621	35261	82.0	~	0.5	225.0	49	4
97621	35262	65.0	~	0.4	314.0	72	5
97621	35263	134.0	~	0.3	68.0	78	4
97621	35264	423.0	~	0.4	131.0	61	5
97621	35265	126.0	~	0.3	143.0	59	4
97621	35266	42.0	~	0.4	163.0	62	5
97621	35267	51.0	~	0.6	136.0	73	6
97621	35268	317.0	~	19.2	209.0	109	102
97621	35269	344.0	~	4.7	379.0	274	105
97621	35270	66.0	~	0.5	83.0	115	6
97621	35271	25.0	~	1.4	58.0	56	6
97621	35272	165.0	~	0.7	81.0	102	5
97621	35273	13.0	~	0.2	66.0	54	5
97621	35274	73.0	~	0.3	163.0	87	7
97621	35275	50.0	~	0.8	124.0	68	10
97621	35276	158.0		0.4	107.0	113	8
97621	35277	107.0	~	0.4	105.0	108	9
97621	35278	138.0	~	0.2	127.0	93	11

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SNIP GOLD MINE CERTIFICATE OF ASSAY

Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
97621	35279	193.0	~	0.5	74.0	62	6
97621	35280	80.0	~	1.4	135.0	63	11
97621	35281	208.0	~	1.1	503.0	61	7
97621	35282	220.0	~	1.2	380.0	50	6
97621	35283	96.0	~	0.3	211.0	48	6
97621	35284	51.0	~	0.1	80.0	52	6
97621	35285	38.0	~	0.1	58.0	99	4
97621	35286	83.0	~	0.4	48.0	58	4
97621	35287	48.0	~	0.4	72,0	134	5
97621	35288	32.0	~	0.1	67.0	86	5
97621	35289	344.0	~	0.3	246.0	38	6
97621	35290	32.0	~	0.5	90.0	52	7
97621	35291	87.0	~	0.7	149.0	113	9
97622	35292	115.0	~	1.2	334.0	97	9
97622	35293	228.0	~	3.4	1472.0	111	4
97622	35294	126.0	· ~	1.4	565.0	71	6
97622	35295	118.0	~	0.9	369.0	82	7
97622	35296	92.0	~	0.8	342.0	151	42
97622	35297	46.0	~	0.4	98.0	55	4
97622	35298	10.0	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	0.5	61.0	23	3
97622	35299	93.0	~	0.3	156.0	57	5
97622	35300	29.0	~	0.5	111.0	54	4
97622	35301	190.0	~	1.6	612.0	49	8
97622	35302	219.0	~	1.0	150.0	79	12
97622	35303	139.0	~	1.6	310.0	279	317
97622	35304	4711.0	1.5	3.7	219.0	170	654
97622	35305	80.0	~	0.7	143.0	64	24
97622	35306	1020.0	0.4	3.5	734.0	153	30
97622	35307	150.0	~	2.3	266.0	123	2028
97622	35308	52.0	~	0.5	125.0	78	11
97622	35309	108.0	~	0.1	180.0	111	11
97622	35310	159.0	~	1.0	433.0	36	7
97622	35311	105.0	~	0.5	154.0	104	6
97622	35312	992.0	~	0.9	490.0	78	6
97622	35313	101.0	~	0.1	221.0	47	4

CERTIFIED BY; Vink A Shindel

SNIP GOLD MINE CERTIFICATE OF ASSAY

Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
97622	35314	97.0	~	0.6	397.0	51	5
97622	35315	51.0	۱ ۲	0.2	266.0	57	4
97622	35316	53.0	~	0.6	132.0	39	- 6
97622	35317	78.0	~	0.8	149.0	94	5
97622	35318	56.0	2	0.3	180.0	87	9
97622	35319	190.0	~	1.5	414.0	233	43
97622	35320	80.0	~	0.1	192.0	84	10
97622	35321	57.0	~	0.4	251.0	651	98
97622	35322	151.0	~	1.4	185.0	241	350
97622	35323	170.0	*	0.7	108.0	221	108
97622	35324	93.0	~	0.9	130.0	509	167
97622	35325	46.0	~	0.8	114.0	210	33
97622	35326	188.0	~	1.6	524.0	2540	44
97624	35327	167.0	~	0.7	121.0	76	18
97624	35328	146.0	~	0.4	188.0	154	50
97624	35329	151.0	~	0.4	138.0	70	109
97624	35330	124.0	~	0.1	109.0	52	5
97624	35331	95.0	~	0.1	114.0	35	12
97624	35332	102.0	~	24.2	1234.0	29650	5138
97624	35333	104.0	~	0.2	86.0	63	36
97624	35334	101.0	~~	0.1	92.0	45	29
97624	35335	96.0	~	0.1	78.0	34	10
97624	35336	170.0	~	5.2	440.0	7225	1494
97624	35337	115.0	~	0.1	94.0	96	23
97624	35338	82.0	~	0.1	146.0	256	234
97624	35339	102.0	~	0.1	116.0	78	20
97624	35340	149.0	~~	3.3	390.0	9850	1001
97624	35341	147.0	~~	0.1	132.0	75	21
97624	35342	123.0	~^	1.1	154.0	30	9
97624	35343	229.0	~	1 .1	257.0	71	8
97624	35344	100.0	~~	0.8	154.0	61	12
97624	35345	135.0	~	0.7	93.0	87	18
97624	35346	128.0	~	0.8	132.0	50	10
97624	35347	123.0	~	0.1	119.0		6
97624	35348	148.0	~	0.5	<u>1</u> 15.0	33	6

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SNIP GOLD MINE CERTIFICATE OF ASSAY

Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
97624	35349	142.0	~	0.3	112.0	36	7
97624	35350	132.0	~	0.6	146.0	22	4
97624	35351	113.0	~	0.6	164.0	114	7
97624	35352	202.0	~	5.3	637.0	1307	66
97624	35353	141.0	~	0.9	359.0	89	7
97624	35354	113.0	~	1.1	286.0	<u>1</u> 21	5
97624	35355	119.0	~	0.2	114.0	50	4
97624	35356	116.0	~	0.3	91.0	41	5
97624	35357	1074.0	0.05	1.1	212.0	62	4
97624	35358	131.0	~	0.5	39.0	39	5
97624	35359	138.0	~	1.6	562.0	68	6
97624	35360	125.0	~	1.2	354.0	59	3
97624	35361	129.0	~	0.8	201.0	64	6
97625	35362	106.0	~	0.1	250.0	90	7
97625	35363	78.0	~	0.1	235.0	63	5
97625	35364	436.0	~	0.1	324.0	252	89
97625	35365	67.0	~	0.1	376.0	94	4
97625	35366	66.0	~	0.7	257.0	85	2
97625	35367	68.0	~	1.4	445.0	58	3
97625	35368	94.0	~	0.2	267.0	105	5
97625	35369	54.0	~	0.9	435.0	136	5
97625	35370	57.0	~	0.2	365.0	164	9
97625	35371	39.0	~	0.4	320.0	60	7
97625	35372	59.0	~	0.4	313.0	66	1
97625	35373	54.0	~	0.3	537.0	57	1
97625	35374	53.0	~	0.6	238.0	154	3
97625	35375	65.0	~	0.4	349.0	138	18
97625	35376	58.0	~	0.6		116	3
97625	35377	77.0	~	0.8	470.0	120	4
97625	35378	52.0	~	0.7	417.0	80	3
97625	35379	288.0	~	0.9	607.0	209	45
97625	35380	67.0	~	0.7	453.0	145	5
97625	35381	75.0	~	0.9	958.0	77	4
97625	35382	84.0	~	0.7	588.0	93	4
97625	35383	72.0	~	0.8	822.0	94	4

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SNIP GOLD MINE CERTIFICATE OF ASSAY

Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
97625	35384	83.0	~	0.7	372.0	100	6
97625	35385	76.0	~	0.6	414.0	71	6
97625	35386	79.0	~	0.6	402.0	98	5
97625	35387	73.0	~	0.3	309.0	96	7
97625	35388	70.0	~	0.7	303.0	442	219
97625	35389	302.0	~	5.0	723.0	1351	850
97625	35390	68.0	~	0.2	259.0	92	7
97625	35391	92.0	~	0.1	250.0	75	4
97625	35392	74.0	~	0.4	331.0	70	4
97625	35393	86.0	~	0.7	495.0	63	5
97625	35394	114.0	~	0.5	271.0	78	5
97625	35395	79.0	~	1.1	471.0	89	7
97625	35396	78.0	~	0.7	514.0	69	8
97628	35397	54.0	~	0.9	354.0	107	12
97628	35398	58.0	~	0.7	299.0	108	8
97628	35399	83.0	· ~	0.4	221.0	107	4
97628	35400	41.0	~	0.1	118.0	50	10
97628	35401	68.0	~	1.2	125.0	59	5
97628	35402	65.0	~	0.9	149.0	60	12
97628	35403	63.0	~	0.5	118.0	112	9
97628	35404	32.0	~	0.7	132.0	82	6
97628	35405	66.0	~	1.2	143.0	123	6
97628	35406	52.0	~	1.7	182.0	166	20
97628	35407	57.0	~	0.6	128.0	88	6
97628	35408	10.0	~	1.0	166.0	105	4
97628	35409	62.0	~	0.8	118.0	84	6
97628	35410	54.0	~	2.5	107.0	1253	282
97628	35411	73.0	2	1.9	101.0	597	168
97628	35412	47.0	~	4.0	102.0	724	627
97628	35413	21.0	~	1.0	45.0	232	34
97628	35414	52.0	~	0.4	25.0	57	7
97628	35415	28.0	~	1.3	36.0	90	7
97628	35416	47.0	4	0.1	69.0	77	6
97628	35417	47.0	~	0.6	193.0	80	5
97628	35418	69.0	ş	0.5	152.0	84	6

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SNIP GOLD MINE CERTIFICATE OF ASSAY

Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
97628	35419	59.0	~	0.2	147.0	96	6
97628	35420	86.0	~	0.4	157.0	55	11
97628	35421	44.0	~	16.0	126.0	395	133
97628	35422	10.0	~	5.1	171.0	214	39
97628	35423	10.0	~	2.5	97.0	103	15
97628	35424	10.0	~	3.6	99.0	107	56
97628	35425	21.0	~	2.3	66.0	74	15
97628	35426	112.0	~	2.1	101.0	267	54
97628	35427	50.0	~	1.5	130.0	160	119
97628	35428	120.0	~	1.4	214.0	392	16
97628	35429	230.0	~	3.0	689.0	151	15
97628	35430	851.0	~	2.4	154.0	85	14
97628	35431	134.0	~	0.8	69.0	80	22
97631	35432	1094.0	1,15	2.2	325.0	245	38
97631	35433	1241.0	0.9	1.6	250.0	37	8
97631	35434	2394.0	3.05	2.4	265.0	140	92
97631	35435	139.0	~	1.9	131.0	182	57
97631	35436	229.0	~	2.1	138.0	285	53
97631	35437	138.0	~	1.3	1 51.0	153	46
97631	35438	115.0	~	2.2	313.0	1636	26
97631	35439	1403.0	1.55	1.4	315.0	34	5
97631	35440	934.0	~	1.2	229.0	14	2
97631	35441	2496.0	3.45	1.7	212.0	65	9
97631	35442	272.0	~	1.0	191.0	156	13
97631	35443	58.0	~	0.4	<u> </u>	58	7
97631	35444	528.0	~	0.9	192.0	115	
97631	35445	188.0	~	0.9	116.0	97	
97631	35446	39.0	~	2.0	226.0	468	
97631	35447	65.0	~	3.2	384.0		
97631	35448	49.0	~	2.2	358.0	271	
97631	35449	67.0	~	0.2	122.0		
97631	35450	74.0		0.4		44	·
97631	35451	10.0	~	0.7	203.0	77	
97631	35452	517.0	~	0.2			
97631	35453	90.0	~	0.8	358.0	55	6

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SNIP GOLD MINE CERTIFICATE OF ASSAY

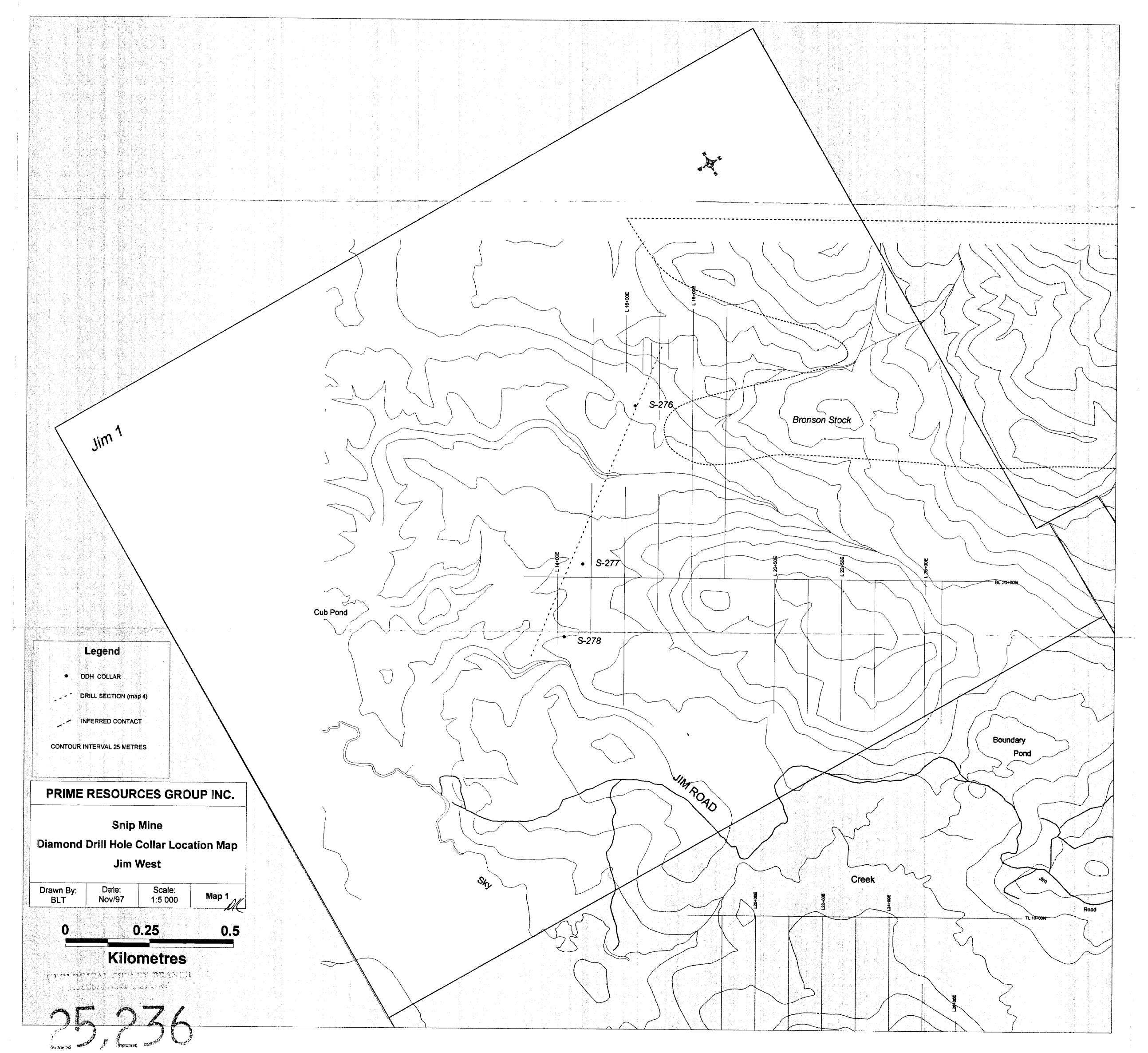
Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
97631	35454	83.0	~	0.8	428.0	53	6
97631	35455	29.0	~	0.6	222.0	53	4
97631	35456	86.0	~	0.9	349.0	74	5
97631	35457	182.0	~	0.7	278.0	107	4
97631	35458	151.0	~	0.8	427.0	82	5
97631	35459	59.0	~	0.2	108.0	120	3
97631	35460	60.0	2	0.1	133.0	112	3
97631	35461	101.0	~	0.4	81.0	134	3
97631	35462	71.0	~	0.5	134.0	112	4
97631	35463	68.0	~	0.4	124.0	127	5
97631	35464	56.0	~	0.8	136.0	193	4
97631	35465	101.0	~	1.0	139.0	167	5
97631	35466	72.0	~	1.1	178.0	124	5
97632	35467	89.0	~	0.1	106.0	145	10
97632	35468	56.0	~	0.2	128.0	122	0
97632	35469	53.0	· ~	3.0	297.0	81	18
97632	35470	103.0	~	3.5	361.0	445	54
97632	35471	58.0	~	0.7	345.0	66	2
97632	35472	107.0	~	0.4	456.0	70	4
97632	35473	77.0	~	0.1	267.0	119	4
97632	35474	138.0	~	0.7	234.0	114	2
97632	35475	38.0	~	0.1	148.0	370	1
97632	35476	165.0	~	1.0	592.0	130	7
97632	35477	61.0	~~	0.6	265.0	162	1
97632	35478	30.0	~	0.7	320.0	158	3
97632	35479	33.0	~	0.4	125.0	241	1
97632	35480	59.0	~	0.3	92.0	220	3
97632	35481	69.0	~~	0.3	122.0	398	
97632	35482	48.0	~~	0.4	157.0	245	
97632	35483	71.0	~	0.4	139.0	135	6
97632	35484	55.0	~	0.7	275.0	121	3
97632	35485	656.0	~	1.8	1412.0	57	10
97632	35486	101.0	~~	0.3	238.0	266	
97632	35487	82.0	~	0.2	87.0	503	
97632	35488	63.0	~~~	0.8	156.0	177	3

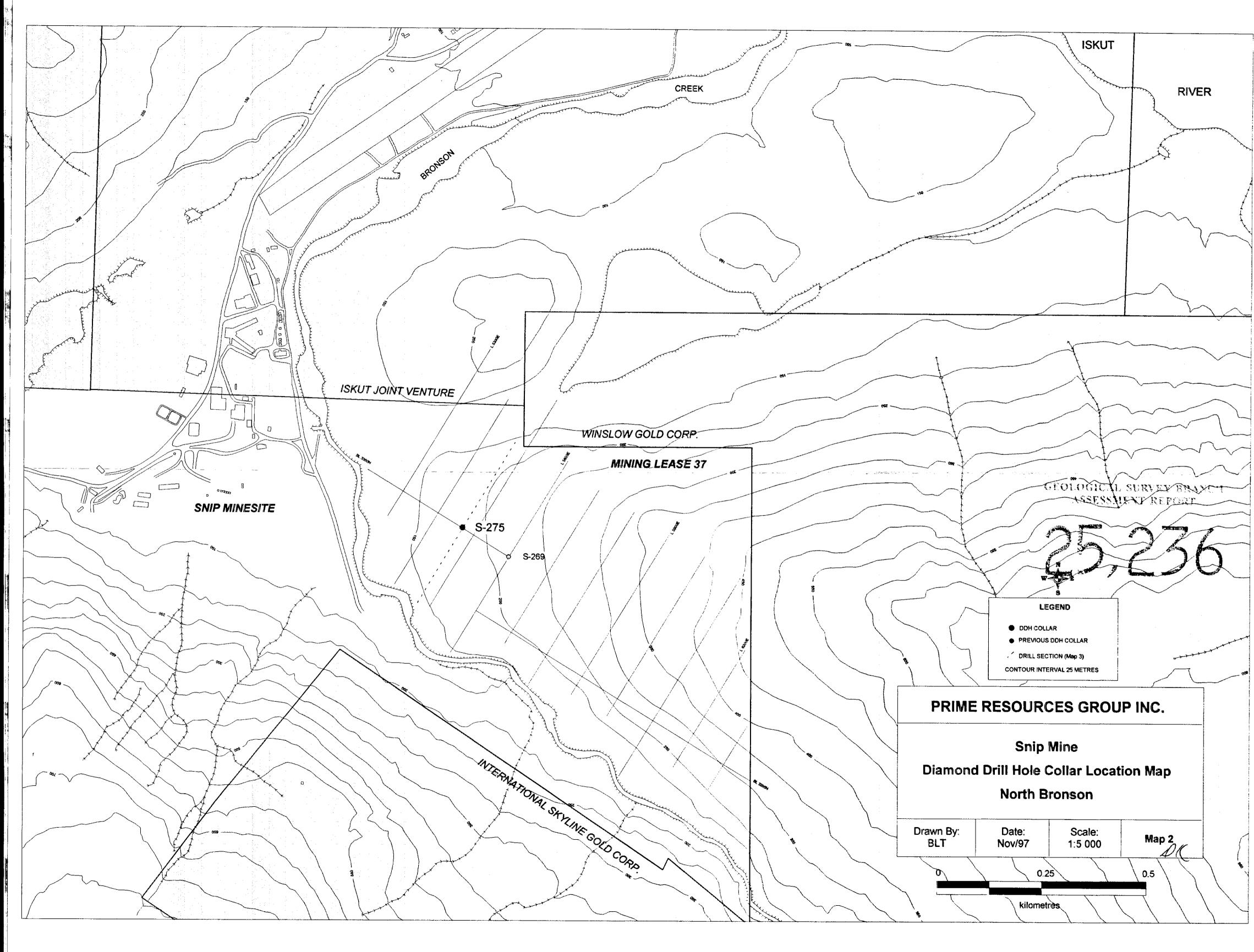
CERTIFIED BY; June A. Standiel

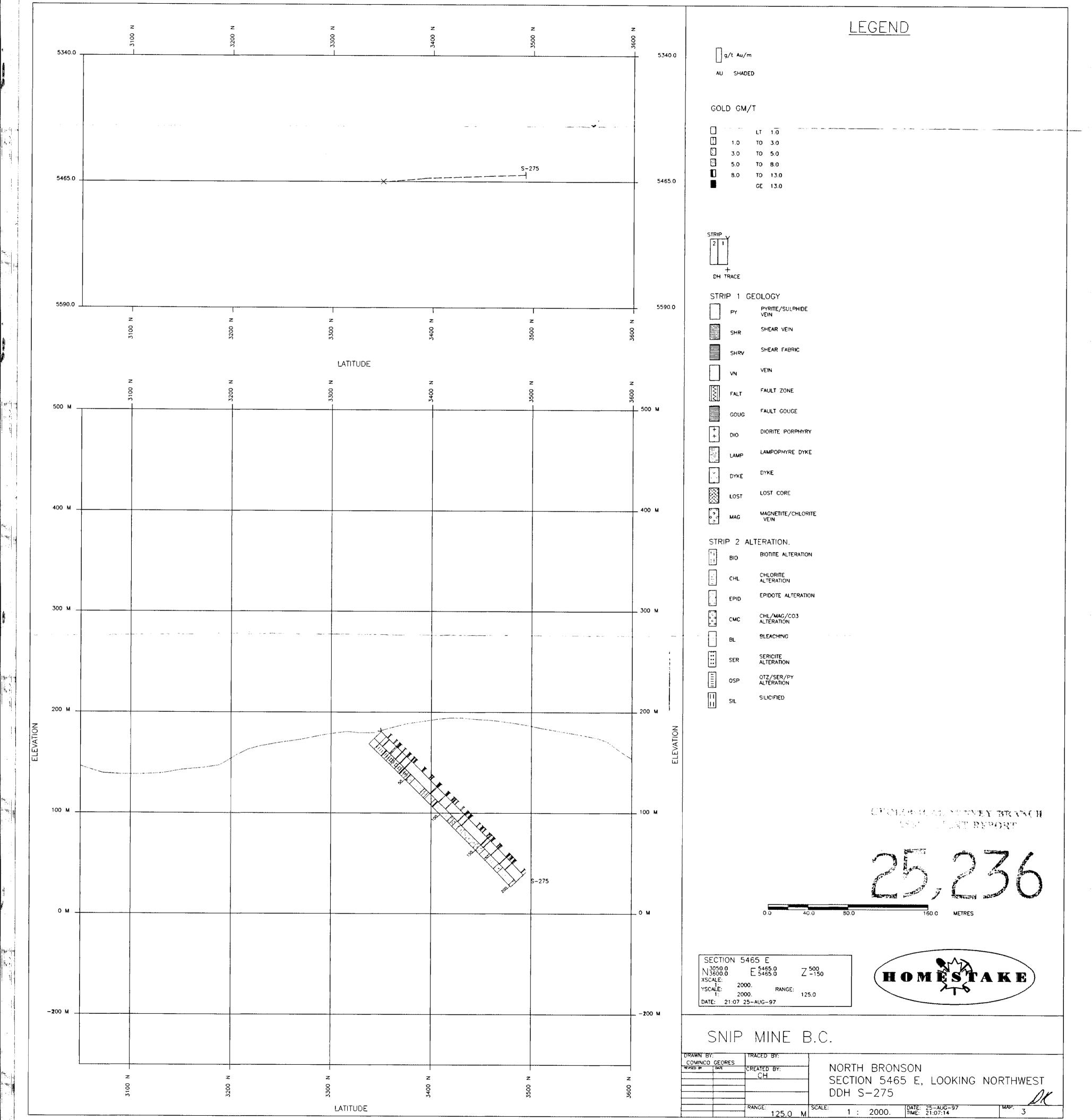
SNIP GOLD MINE CERTIFICATE OF ASSAY

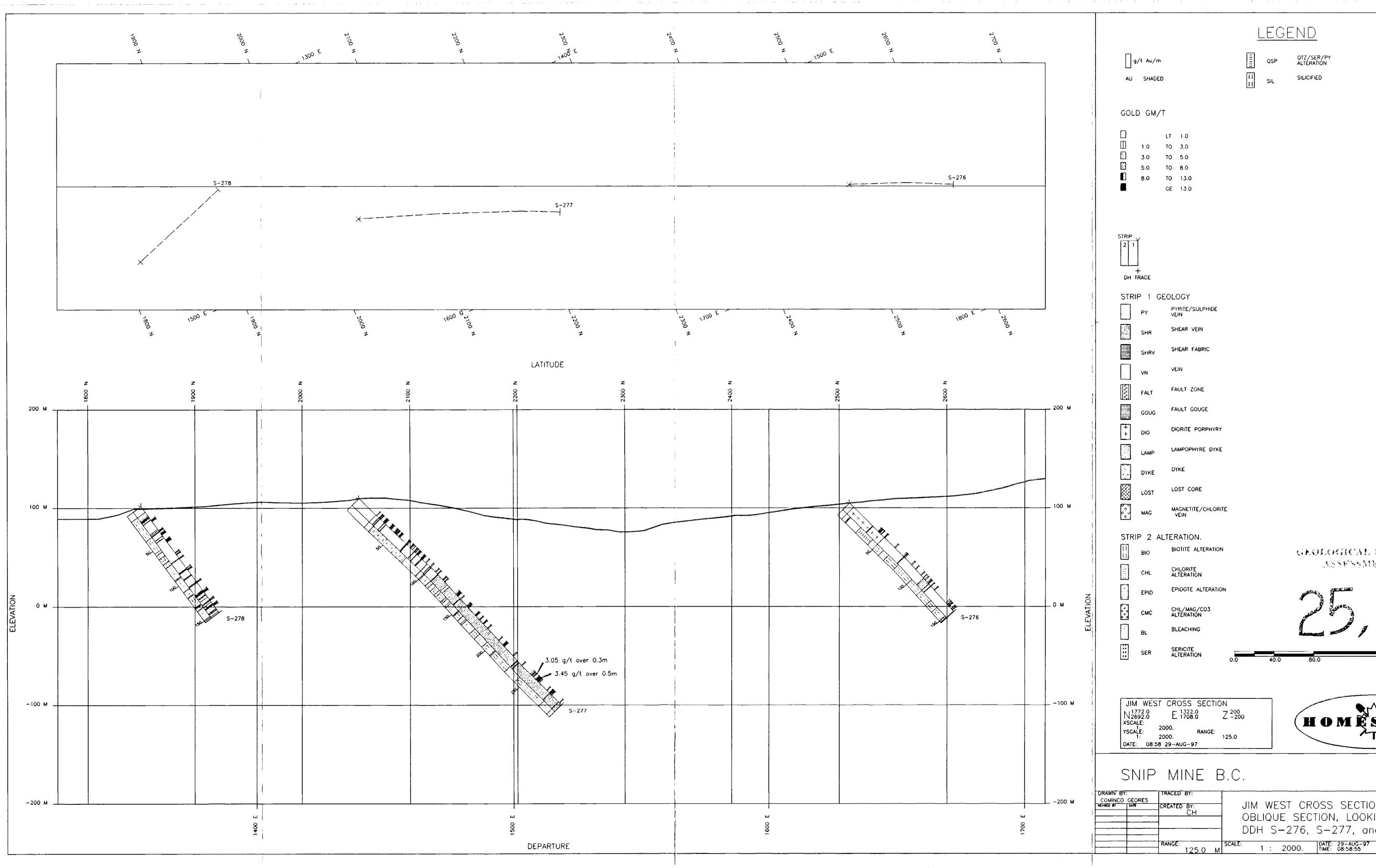
Job#	Sample#	Au ppb	F.A.	Ag g/t	Cu g/t	Zn g/t	Pb g/t
97632	35489	57.0	~	0.3	165.0	187	5
97632	35490	75.0	~	0.5	156.0	140	8
97632	35491	570.0	~	8.8	1059.0	977	22
97632	35492	79.0	~	0.8	232.0	108	12
97632	35493	127.0	~	1.0	194.0	101	6

CERTIFIED BY; Jack A Stundely









LEGEND QTZ/SER/PY ALTERATION QSP SILICIFIED SIL GEOLOGICAL SURVEY BRANCH 化合金的合金的复数 化自己合合化的 60.0 METRES HOMESTAKE JIM WEST CROSS SECTION OBLIQUE SECTION, LOOKING NORTHWEST DDH S-276, S-277, and S-278

MAP: 4