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Report On

Geological, Geochemical, and Geophysical Surveys

JORDAN RIVER PROPERTY

Copeland Group Lat. 51° 07.5'N Long. 118° 24'W UTM 5664500N 401500E

Frisby Group Lat. 51° 08.5'N Long. 118° 17'W UTM 5666000N 40000E

Revelstoke Mining Division NTS 82M 1W

Owner and Operator: First Standard Mining Ltd. 802-6540 Burlington Ave. Burnaby, B.C. V5H 4G3

GEOLOGICAL BRANCH ASSESSMENT REPORT

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22,029

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Date: December 10, 1991

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1.0 Introduction

1.1 Summary

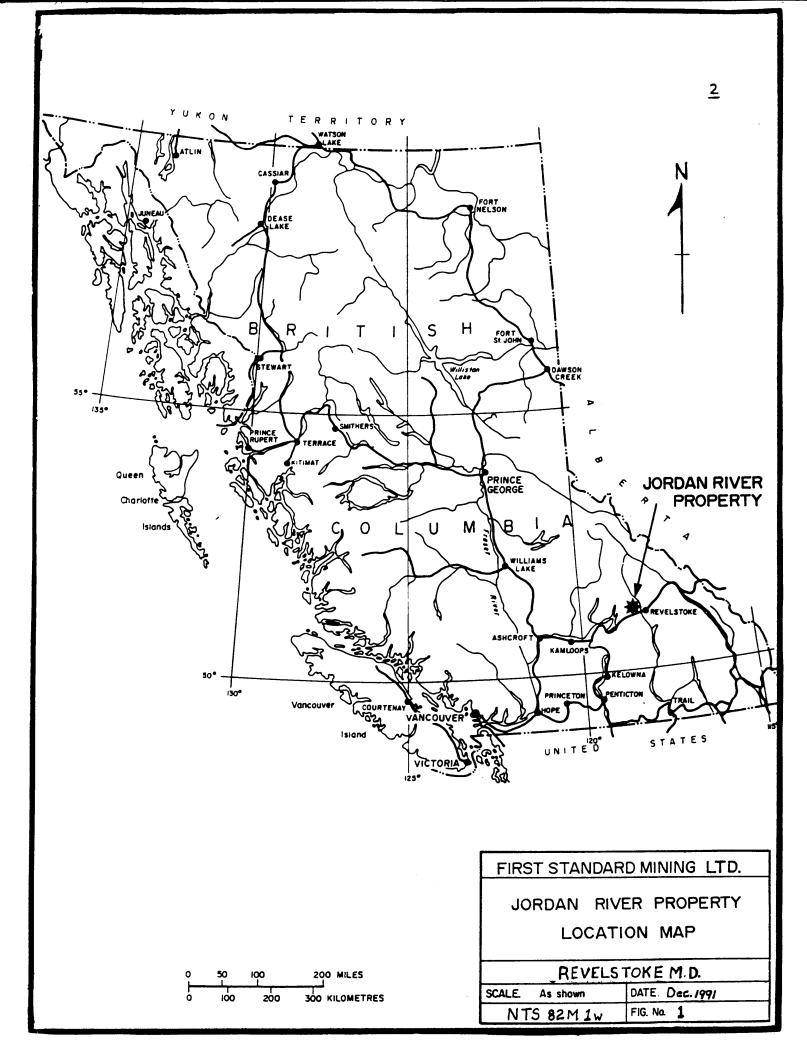
The Jordan River property, near Revelstoke, B.C., consists of the Copeland Group and the Frisby Group. The Copeland Group contains the King Fissure Pb-Zn-Ag-Ba massive sulphide deposit, while the Frisby Group covers additional exposures of the sulphide-bearing stratigraphy.

Work on the Jordan River property in 1991 consisted of linecutting, geological mapping, geophysical (magnetic and VLF-EM) surveys, a soil geochemical survey, prospecting, and rock chip sampling.

On the King Fissure Deposit, geological mapping and geophysical surveys were successful in tracing the massive sulphide layer around the eastern closure of the Copeland synform. Evidence seen in drag folds, the general fold morphology as indicated by surface exposures and limited drilling, and in regional folding patterns, suggests that hinge zone thickening should occur in the eastern half of the synform. Preliminary examination of the West zone has demonstrated the presence of footwall sulphide stockworking, brecciated zones within the massive sulphides, and a massive sulphide-barite sequence exceeding 4m thickness. Brecciation and high-grade Pb-Zn-Ag stockworking is also seen in the Cliff Zone, where massive sulphides commonly exceed 2m thickness. These features indicate proximity to exhalative sulphide vent zones.

Prospecting and geological mapping on the Frisby Group have traced the favourable sulphide-host carbonate stratigraphy for approximately 2.5km. Within this sequence, the massive Zn-Pb sulphide layer has been traced for over 1.2km in the Big Slide zone, and remains unexplored to the south.

A light rare-earth bearing extrusive carbonatite layer was recognized in the King Fissure Deposit in 1990. This unit, occurring less than 50m stratigraphically below the massive sulphide horizon, was also successfully traced on the Frisby Group.



1.2 Location, Access, and Vegetation

The Jordan River property lies in mountainous terrain on both sides of the Jordan River, approximately 19 km northwest of the town of Revelstoke, B.C.. The Copeland and Frisby groups lie west and east, respectively, of the Jordan River. Elevations range from 620m in the river valley to 2560m at Mount Copeland. Access to the property is by a 15 minute helicopter flight from Revelstoke. A road accessing the former Mount Copeland molybdenum mine leads to within approximately 10 km of the King Fissure Deposit. An old pack trail leaves the road at Hiren Creek and proceeds up Jordan River and Copeland Creek to the King Fissure Deposit, a distance of 10km.

The Copeland Group is mostly rugged, rocky terrain, entirely in alpine. The upper part of Frisby Ridge on the Frisby Group lies mostly in sub-alpine, while the lower regions (generally below 1500m) are densely treed with fir, cedar, spruce, and pine. Open areas are covered with thick slide alder and scrub brush. Drainage from the Copeland Group is into Copeland and Hiren Creeks, which in turn drain easterly into the Jordan River. Drainage from the Frisby Group is both west into the Jordan River and east into Lake Revelstoke on the Columbia River.

1.3 Property Details

The Jordan River property is composed of two groups totalling 185 units as follows*:

	Frish	y Group	
Claim	Record #	Units	Expiry Date**
Jordan 2	248446	10	Nov. 20, 1993
Jordan 3	248447	20	Nov. 20, 1993
Jordan 4	248448	10	Nov. 20, 1993
Frisby 1	248440	12	Nov. 6, 1993
Frisby 2	248441	12	Nov. 20, 1993
Frisby 3	248442	12	Nov. 20, 1993
Frisby 4	248443	12	Nov. 20, 1993

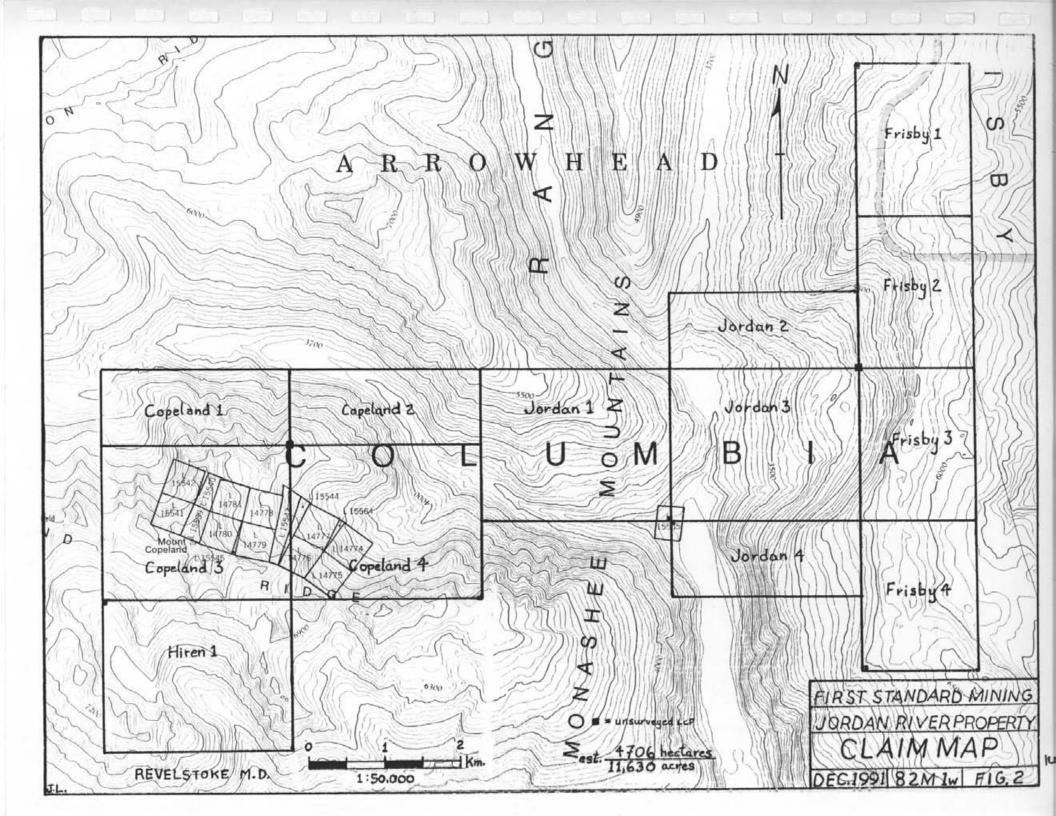
Total size of the Frisby Group is 88 units.

	Copela	nd Group	
Claim	Record #	Units	Expiry Date**
Copeland 1 Copeland 2 Copeland 3 Copeland 4 Jordan 1 L. 14774 L. 14775 L. 14775 L. 14776 L. 14777 L. 14778 L. 14779	248318 248319 248321 248320 248445 	10 10 20 20 20 	Nov. 22, 1992 Nov. 22, 1992 Nov. 22, 1993 Nov. 22, 1993 Nov. 22, 1993 Nov. 20, 1992
L. 14780 L. 14781			
L. 15539 L. 15540			
L. 15541 L. 15542 L. 15543			
L. 15545 L. 15544 L. 15545			
L. 15564			
L. 15565			

Total size of the Copeland Group is 97 units. (* The 20 unit Hiren 1 claim, also part of the Jordan River property, staked in October, 1991, is not included in this report.)

(** Assuming acceptance of this report)

All claims, including crown-granted claims which cover the King Fissure Deposit, are 100% owned by First Standard Mining Ltd.. The Copeland 1-4 claims were staked to protect ground surrounding the deposit. The Jordan 1-4 and Frisby 1-4 claims were staked to cover favourable stratigraphy east of the King Fissure Deposit.



1.4 History

The King Fissure Deposit was discovered in the late 1800's by early prospectors who cut a trail into the property and drove three short tunnels. No other systematic exploration was carried out on the property until the late 1950's when American Standard Mines Ltd. and Bunker Hill Mines Ltd. conducted extensive surface sampling and metallurgical testing. Bralorne Pioneer Mines Ltd. drilled five diamond drill holes in 1963 and another five in 1965. A summary report, drill logs, and preliminary mine plans prepared subsequent to this drilling are not available.

In 1970, Dr. J. T. Fyles of the B.C. Department of Mines published a report on geology and mineral deposits of the Jordan River area. The report included detailed maps and preliminary cross sections of the King Fissure Deposit.

In the fall of 1990, a re-examination of the property was conducted by J. Laird and R. MacGillivray. Significant results of this work included identification and sampling of several high-grade Pb-Zn-Ag-Ba zones within the King Fissure Deposit, and the recognition of a light rare-earth bearing extrusive carbonatite layer stratigraphically below the massive sulphides.

1.5 Summary of Work

Work on the Jordan River property was conducted between August 23 and September 30, 1991.

Copeland Group:

All work on the Copeland Group was performed on the Copeland 3-4 claims.

- Grid establishment A total of 7.8km of line were established on two grids on the King Fissure Deposit.
- Geological mapping 7.8km's of grid were mapped at a scale of 1:2,500.
- 3. Geochemical sampling A total of 29 rock chip samples were collected from the Copeland Group; all samples were variously analyzed by ICP or whole rock methods; 14 samples were assayed for base and precious metals, and 4 extrusive carbonatite samples were analyzed for light rare-earth element content.
- Geophysical surveys 7.4km's of grid were surveyed for magnetics; 4.35km's of grid were surveyed by VLF-EM methods.

Frisby Group:

- 1. Grid establishment 15.6km's of grid were established on the Frisby 1-3 claims.
- Geological mapping 12.7km's of the grid were mapped at a scale of 1:2,500.
- 3. Geochemical sampling 173 soil samples were collected from grid on the Frisby 2 claim and analyzed for 35 elements by ICP methods; A total of 32 rock chip samples were collected from the Frisby 1-4 and Jordan 2-4 claims; all samples were variously analyzed by ICP or whole rock methods. In addition, 12 samples were assayed for base and precious metals, and 4 extrusive carbonatite samples were analyzed for light rare-earth element content.
- Geophysical surveys 0.49km's of grid on the Frisby 3 claim were surveyed for magnetics; 3.0km's of grid on the Frisby 2 claim were surveyed by VLF-EM methods.

2.0 Geology

2.1 Regional Geology (Figure 3)

The Jordan River property is underlain by Monashee Complex metamorphic rocks which lie within the Paleozoic and older Shuswap metamorphic complex. The Monashee Complex consists of a series of granitic gneiss domes of probable Aphebian age enveloped by metasedimentary gneisses and schists (Hoy, 1987). The Jordan River map area lies on the southeastern flank of the northernmost of these domes, the Frenchman Cap gneiss dome.

Compilation work conducted in the winter of 1990-1991 has resulted in stratigraphic and structural interpretations substantially different to those in previous publications (ex. Fyles, 1970, Hoy and Brown, 1980). This re-interpretation, made possible largely by the recognition of an extrusive carbonatite layer in the King Fissure Deposit, has enabled correlations to be made with stratigraphy elsewhere in the Monashee Complex, such as the Cottonbelt area on the northwest flank of the Frenchman Cap gneiss dome (see Hoy, 1987). Units referred to in this report are keyed to those on the Jordan River area geology map (Figure 3).

The Frenchman Cap gneiss dome consists largely of medium-to dark-grey, medium-grained, granitic biotite-feldspar gneiss. Within the granitic gneiss are inclusions of biotite-hornblende gneiss and light grey granitic gneiss. Folding within the gneiss intensifies towards the unconformably overlying metasediments (Fyles, 1970). Previously referred to as mixed gneiss (Wheeler, 1965; Fyles, 1970), it is herein referred to as Unit 1 granitic gneiss.

Overlying the core gneisses are Unit 2 quartz-pebble conglomerates, white quartzites, and less commonly, quartz-mica schists. In most places the conglomerates and quartzites are between 15 and 60m thick, but in hinge zones of folds they can exceed 300m thickness. Cross-bedding has been noted in the transition zone between the lower conglomerate and overlying quartzites (Fyles, 1970).

Above Unit 2 lies Unit 3, a package of green calc-silicate gneiss, calcareous schist, marble, biotite schist, quartzite, and Tremolite-rich, locally dolomitic marble occurs as discontinuous layers and lenses. Where quartzites achieve appreciable thicknesses they are recognized separately as Unit 3q. Amphibolite sills(?) are also locally significant. Fenite has been noted in correlative stratigraphy in the Mt. Grace area on the northwest flank of the Frenchman Cap gneiss dome (Hoy, 1987). Unit 3 has been described as being a few hundred feet thick, pinching out west of the Jordan River, south of Hiren Creek (Fyles, 1970).

Overlying Unit 3 is grey-green coloured calc-silicate gneiss of Unit 4. Amphibolites intercalated with the calc-silicate gneiss, generally less than 2m thick, are thought to be sills due their pinching and swelling nature. Quartzites also occur within this unit, and where significantly thick are mapped as Unit 4q.

Above Unit 4 lies Unit 5, a predominantly carbonate sequence hosting the massive Pb-Zn-Ag-Ba sulphide layer. Lithologies within this unit are continuous over large areas, and are directly correlatable with massive sulphide-bearing stratigraphy described in the Mt. Grace area (Hoy, 1987).

In the Jordan River area the base of the Unit 5 sequence is indicated by a 0.5-1.0m thick gneissic-textured marble layer, informally named the basal marble. This marble consists of white to light-grey calcite, with brown biotite laminations occurring near the upper contact. This upper contact with the extrusive carbonatite layer is gradational over approximately 15cm, with the base of the carbonatite containing laminations of light grey marble. The continuity and contact relationships of the basal marble suggest it may also have an exhalative origin.

The extrusive carbonatite (Unit 5c) is medium to dark brown in colour, commonly over 5m thick, and ranges from non- to highly fragmental in nature. Mineralogy of the matrix consists primarily of calcite and phlogopite, with lesser fluorapatite and pyrochlore, while the light grey breccia fragments consist almost entirely of albite and phlogopite (Hoy, 1987). Fragment size ranges from less than 1cm to over 20cm. The largest fragments occur in the most intensely brecciated sections, and are interpreted to be near vent zones. More detailed descriptions of the carbonatite can be found in Hoy (1987). Above the carbonatite lies interlayered fine grained mica schist and calcsilicate gneiss and schist, in turn overlain by a regionally continuous white marble layer, informally named the marker marble The carbonatite-marker marble contact appears to be (Unit 5m). gradational, with brown phlogopite-biotite layers occurring near the base of the marble; these micaceous intercalations do not appear to be appreciably anomalous in rare earth element content (T. Hoy, verb. comm.). Thickness of the marble is commonly 3-Mineralogy is almost entirely calcite, although accessory 10m. scapolite occurs on Frisby Ridge. Above the marker marble lies relatively non-descript, grey, fine-grained mica schist and calcsilicate gneiss 5-30m thick. This is overlain by the massive sulphide sequence (Unit 5s). The sulphide layer, while not ubiquitous, is locally well-developed (ex. King Fissure and Cottonbelt deposits). Sulphides consist primarily of fine to coarse grained pyrrhotite, sphalerite, galena, and pyrite, commonly within a siliceous or calcareous matrix. Barite ranging in occurrence from discrete crystals to massive layers is intimately associated with the sulphides. More detailed descriptions of the massive sulphide layer are presented in property geology sections.

Above the sulphide layer lie quartzites and quartz-biotite schists of Unit 5q grade upwards into Unit 6. Amphibolite sills(?) displaying local contact metamorphism are common in the 5q metasediments.

Unit 6 medium-grained biotite-sillimanite schists and quartzites commonly form rusty weathering cliffs, best exposed on the King Fissure Deposit, on the north side of lower Copeland Creek, and on the western slope of Frisby Ridge. The schists often have a knotted appearance, and are migmatitic in the centre of the King Fissure Deposit. Thin (<1m), irregular marble layers occur within Unit 6 on Frisby Ridge and Mount Copeland.

Intruding the metasedimentary sequence are gneissic nepheline syenites (Unit N). These grey, medium-grained feldspar-biotite gneisses have moderately-well defined foliations and locally pitted weathering surfaces. Nepheline amounts to as much as 20 percent; accessory minerals include calcite, zircon, sphene, fluorite, and magnetite. Concentrations of molybdenite occurring in the border phases of the nepheline syenite have been mined at the Mount Copeland molybdenum mine. Lack of quartz and effervescence of some samples with acid distinguish the syenite from biotite-quartz-feldspar gneisses (Fyles, 1970). Zircons extracted from the nepheline syenite have been dated at 740 +/-36 Ma (Parrish and Scammell, 1988). Regional mapping indicates that the syenites' preferred level of intrusion was in the upper regions of unit 3, and that it is folded by the earliest recognized deformation. This is best displayed in the area southwest of Mount Copeland.

The youngest rocks recognized in the Jordan River area are Tertiary lamprophyre dykes. Ranging from <1m to over 3m thickness and often occurring in swarms, these dykes tend to fill northerly trending faults and fractures. Rarely the lamprophyre forms sills. In the King Fissure Deposit area, fault-hosted and manto style Pb-Zn-Ag mineralization is associated with the dykes and structures. On the north side of Copeland Creek, on the Wild Goose property, similar structures host Au-bearing Pb-Zn-Ag-Cu veins.

Three phases of folding are recognized in the Jordan River area (Fyles, 1970). Phase 1 folds, having warped axial planes dipping primarily to the southwest, are isoclinal with highly attenuated limbs and thickened hinge zones. Thrust faulting and local shearing parallel to the foliation accompanies Phase 1 folding. Phase 2 folds are generally overturned, with axial planes dipping at low to moderate angles to the south and southwest. Although most Phase 2 folds are of a concentric style, thickened hinge zones have been noted, particularly near the gneiss dome. One large Phase 3 antiform has been mapped straddling the Jordan River valley. The axis of this fold plunges moderately to the south, dipping steeply to the east (Fyles, 1970).

2.2 Copeland Group Geology (King Fissure Deposit - Figure 4)

The King Fissure Deposit lies within a southeasterly trending, southwesterly dipping syncline with an overturned southern limb, known as the Copeland synform (Fyles, 1970). Folding is open and concentric at the western end, but tightens considerably towards the east. The synform has approximate dimensions of 2.5km long by 0.8km wide. Stratiform massive sulphides are seen on both limbs of the fold. Several zones within the deposit have been established by Riley (1961); the West, Cliff, and East zones as well as the newly named Northeast and Lake zones were examined during the 1991 fieldwork.

2.2.1 Lithology

Rock units 4, 5, and 6 are present in the King Fissure Deposit area. At the bottom of the sequence, Unit 4 grey-green gneiss, quartzites and quartz-biotite schists, form virtually inaccessible cliffs along the overturned southern limb of the deposit. Commonly weathering to grey and black, these rocks are unusually rusty above the Cliff Zone.

Above Unit 4, the Unit 5 basal marble is commonly less than 1m thick. In gradational contact with the basal marble is the extrusive carbonatite (Unit 5c). Best exposures of the carbonatite occur in the Cliff and Northeast zones. In the Cliff zone the carbonatite is approximately 5m thick and almost entirely tuffaceous in nature. Rare fragments less than 2cm in size tend to occur along discrete horizons. Repetitive centimetre-scale interlayering of fine and medium grain sizes indicates several episodes of deposition. In the Northeast zone, the carbonatite is highly fragmental and reaches 10m in thickness. Poorly sorted, matrix-supported fragments up to 25cm in size form approximately 20% of the volume, and are interpreted to be indicative of a proximal source vent. Light rare-earth element content is markedly higher in the Northeast zone samples than in the Cliff zone samples, particularly with respect to Ce, La, and Nd.

Discontinuous medium to coarse grained amphibolite layers are often present within the immediate carbonatite stratigraphy, and probably represent metamorphosed basic volcanics and related intrusives (Hoy, 1987). Amphibolite samples KF-1 and KF-2 from the King Fissure deposit are chemically similar to basic metavolcanic rocks near Blais Creek in the Cottonbelt area (Hoy, 1987, p. 19).

The marker marble, Unit 5m, ranges from 3-10m in thickness, is composed almost entirely of coarse-grained white calcite, and may also be of exhalative origin.

Above the marker marble lies feldspar-porphyroblastic grey mica schist with lesser calc-silicate schist. This unit is uniformly non-descript, notable only in that it directly underlies the massive sulphides.

The massive sulphide horizon (Unit 5s) can be traced throughout the entire King Fissure deposit with the exception of talus and snow covered intervals. Greatest known primary massive sulphide thicknesses occur in the West and Cliff zones. Mineralogy consists mostly of fine to coarse grained pyrrhotite, sphalerite, galena and pyrite, often within a siliceous or calcareous matrix. Massive barite occurs with sulphides in the Northeast and West zones. More detailed descriptions of the massive sulphides and barite are presented in section 2.2.3.

Directly overlying the sulphide horizon are more grey mica schists and calc-silicate gneisses, in turn overlain by interlayered quartzites and mica schists (Unit 5q). The quartzites are generally white to tan coloured and have welldeveloped micaceous partings. Most of the mica is muscovite, although a green (fuchsite?) mica is often present.

Biotite schist layers become more prevalent upsection, leading into biotite-sillimanite schist and quartzite of Unit 6 occurring in the core of the Copeland synform. This highly tectonized and locally migmatitic unit weathers to a strongly Feoxidized surface. Chaotic ptygmatic folding is common, and displacement along foliation planes may be significant, but is difficult to measure. Several northerly trending biotite-lamprophyre dykes cut through the deposit, particularly in the central and eastern regions of the Copeland synform. Often occurring in swarms, the lamprophyres weather to a dark brown colour, with fine grained biotite and subordinate amphibole within an aphanitic groundmass. Thickness of individual dykes ranges from <0.5m to 3m.

2.2.2 Structure

The King Fissure Deposit lies within a southeasterly trending, southwesterly dipping syncline with approximate dimensions of 2.5km long by 0.8km wide. The fold has been named the Copeland synform by Fyles (1970).

The Copeland synform is open and concentric in the western end, but tightens considerably to the east. In the western end, an anticline superimposed on the keel of the Copeland synform has created a "W" shaped folding pattern, effectively raising the structural level of the keel and establishing easterly plunges to Structural measurements in the West zone indicate that folds. the Copeland synform plunges approximately 30° towards 150° (Fyles, 1970). The central antiform, plunging more steeply than the Copeland synform, diminishes in magnitude towards the east, at some point disappearing entirely as three fold axes coalesce Near this point on the surface a major northerly into one. trending fault zone, known as the Camp fault, cuts across the synform with a dextral offset of approximately 20m. This late structure may be related to stress created at the junction of the earlier folding. East of the Camp fault the Copeland synform is assumed to have a near horizontal keel. East of the King Fissure Deposit, structural mapping indicates that fold axes in Unit 4 rocks plunge approximately 15° to the west (Fyles, 1970).

On L24+00E on the East zone grid, massive sulphides on each limb of the Copeland synform are approximately 150m apart. Geological mapping and magnetic survey data indicate that the closure of the synform probably lies under talus and thick bush cover between L27+00E and L29+00E (Figures 8 & 9).

2.2.3 Mineralization

Exploration on the King Fissure Deposit is focused primarily on stratiform base-metal massive sulphides (Unit 5s) which occur near the top of the Unit 5 carbonate sequence. The sulphide horizon is well exposed along both limbs of the Copeland synform. Numerous trenches and shallow adits occur in the Cliff, East, and Northeast zones.

Cliff Zone:

In the Cliff zone, massive sulphides range from 1.5m to >3m thick. A vertical zonation within the massive sulphide layer is recognizable; at the base is a dark weathering 0.2-1.0m layer of mostly sphalerite and galena, with minor pyrrhotite. This is overlain by 0.5-2m of rusty weathering, massive, fine-grained pyrrhotite containing eyes of grey quartz and fine grained sphalerite and galena. A representative of this type of mineralization is photographed in Fyles (Plate XV, 1970).

Above the pyrrhotite-dominant middle layer is a 0.2-1.0m siliceous horizon hosting coarse grained pyrite with galena, sphalerite, and minor pyrrhotite. This siliceous upper layer is most easily distinguished by its abundant pyrite and light grey to white weathered surfaces. Brecciation and footwall sulphide stockworking were noted in the Cliff zone. Barite has not been recognized.

East Zone:

In the East zone, massive sulphide layers are approximately 0.5-1.0m thick, consisting mostly of sphalerite and galena with lesser pyrrhotite and pyrite within a siliceous matrix. Barite has not been noted. On the north limb (near the north end of East zone grid L25+00E) is a pyrrhotite-rich zone containing wallrock breccia fragments. This zone is similar in mineralogy and appearance to the middle layer of the Cliff zone massive sulphide unit. Multiple layering over an interval of 3m occurs on the north limb. The extrusive carbonatite layer is present in the East zone but has not been sampled.

Northeast Zone:

In the Northeast zone there are up to three massive sulphide layers separated by calcareous and siliceous layers with barite, spanning a total interval of 1.5m-3.0m. Three sulphide layers, intersected in diamond drill holes drilled by Bralorne Pioneer Mines Ltd., were previously interpreted to be structural repetitions of the same unit. Small sulphide replacements and mantos occur adjacent to late structures in the Northeast zone.

The well-exposed extrusive carbonatite layer is locally highly fragmental and reaches approximately 5m in thickness. Fragments exceeding 25cm size were noted. Two carbonatite samples from the Northeast zone were relatively enhanced in light rare-earth content compared to the Cliff zone.

West Zone:

Massive sulphide mineralogy in the West zone consists of galena, sphalerite, pyrite, and pyrrhotite. The massive sulphide layers display several important characteristics. Perhaps most highly notable is the occurrence of massive barite interbedded with the sulphides. The light grey coloured barite is medium to coarse grained and contains a fine mesh-work of galena. One section of six massive sulphide layers plus barite was measured to be 4.5m thick. The mineralized horizon contains highly brecciated lenses with <1cm-10⁺cm wallrock fragments in a massive sulphide and barite matrix. Flow textures are visible around breccia fragments which often display an internal foliation and are slightly elongated parallel to layering. The breccias are thought to have formed below areas of slope instability during sulphide deposition. Near these breccia zones is an underlying sulphide stockworking. This stockworking, along with brecciation, multiple layering, and the occurrence of barite are indicative of proximity to sulphide venting.

The extrusive carbonatite occurs in the West zone but has not been sampled.

Lake Zone:

Mineralization in the Lake zone is distinct in that the sulphide layer has a unique mineralogy and does not exceed 1m in thickness. Galena, sphalerite, pyrite and minor greenockite are dominant with pyrrhotite being notably absent. An emerald greencoloured silicate mineral, found intimately associated with the sulphide layer in the Lake zone, was recently identified as gahnite, a zinc-bearing spinel (Hoy, pers. comm., 1991). This mineral also occurs in the sulphide layer at the Cottonbelt deposit (Hoy, 1987) and in the metamorphosed early Proterozoic Zn-Pb-Cu Saxberget deposit in Sweden (Vivallo and Rickard, 1990).

The extrusive carbonatite is well exposed in the Lake zone, but was not sampled.

Summary:

The King Fissure Deposit has previously been described as being formed by replacement of an impure marble (ex. Riley, 1961). The writers of this report suggest a sedimentaryexhalative (sedex) origin. Supporting evidence includes multiple layering of barite and sulphides, and footwall stockworking adjacent to brecciated zones within the massive sulphides. Sulphide deposition and widespread amphibolite intrusions within a thick metasedimentary sequence are indicative of episodic basinal subsidence during an extensional tectonic regime.

KING FISSURE DEPOSIT SUMMARY OF ASSAY RESULTS 1991 SAMPLING PROGRAM						
Sample	Zone	Туре	Pb(%)	Zn(%)	Ba (%)	Ag(g/t)
CZ-1	Cliff	1.0m chip	29.80 2	5.70	0.01	324.0
CZ-2	Cliff	2.0m chip	2.26 12	2.20	0.45	29.0
EZ-1	East	dump grab	12.90	5.70	0.01	87.5
EZ-2	East	talus grab	3.90 1	8.90	0.01	57.5
EZG-1	East	0.5m chip	6.50	9.24	0.78	62.6
EZG-2	East	0.5m chip	0.83	5.79	0.42	11.0
EZG-4	East	1.0m chip	46.70	3.65	0.73	375.0
EZG-5	East	1.0m chip	5.45 12	2.30	3.81	61.8
NEZ-1	Northeast	1.5m chip	10.40	2.25	21.54	71.5
NEZ-2	Northeast	1.5m chip	2.05	2.74	1.45	18.9
NEZ-3	Northeast	1.0m chip	14.20	0.19	43.53	78.2
NEZ-4	Northeast	1.0m chip	3.30	9.50	3.25	29.3
WZ-1	West	2.0m grab	1.20 1	1.35	0.89	15.7
WZ-2	West	2.6m chip	8.10	1.70	40.92	68.8
CF-1	Camp fault	grab -	79.20	1.22	0.005	132.0

2.3 Frisby Group Geology

The Frisby Group is underlain by units 1-6 (Figure 3). Units 4, 5, and 6 outline a large synform having an overturned southern limb and a warped axial plane dipping to the south and southeast. The fold, which trends across the Jordan River, is similar to the Copeland synform and has been informally named the Frisby synform. Grids on Frisby Ridge were established over areas indicated on regional maps to be underlain by Unit 5, and in particular the marker marble (Unit 5m). The overturned south limb of the Frisby synform is repetitively folded along the top of Frisby Ridge, resulting in complex outcrop patterns. A massive sulphide layer (unit 5s) up to 1m thick was traced intermittently for 1.2km in the Big Slide zone. The extrusive carbonatite (unit 5c) is present throughout the explored Frisby Ridge area.

2.3.1 Lithology

Frisby Ridge North Grid (Figure 29):

The Frisby Ridge North Grid covers favourable Unit 5 stratigraphy on the Frisby 1 claim. The lowermost (structurally highest) stratigraphy, outcropping in the northeastern region of the grid, consists of interbedded calc-silicate schists and quartz-biotite schists. Stratigraphically above this lies the extrusive carbonatite, which subcrops near TL 37+00E, 78+00N. The carbonatite contains rare, small (<2cm) breccia fragments, and appears highly recrystallized; a sample from this location was highly anomalous in Ce (488.0 ppm), La (304.0 ppm), and Nd (150 ppm). Thickness of the carbonatite is impossible to determine at this location. Above the carbonatite are poorly exposed guartz-biotite and calc-silicate schists, in turn overlain by the marker marble (Unit 5m). The marble, averaging 5m thickness, weathers white to light grey and is coarsely recrystallized. Where not exposed, the marble can often be traced by following sinkholes. Marble in the western region of the grid contains scattered 0.5-3cm crystals of beige-coloured Scapolite within marble in the Monashee Complex is scapolite. indicative of deposition under hypersaline conditions common in restricted lagoons and tidal flats (Hoy, 1987). Above the marker marble lies feldspar-porphyroblastic quartz-biotite schist and calc-silicate schist. The uppermost stratigraphy exposed on the grid consists of fine to medium grained, well-bedded, brown to grey quartzites and quartz-mica schists. Graded bedding and cross-laminations are visible in guartzites outcropping in the northwest area of the grid. Minor amphibolite sills(?) are present in the southwest corner of the grid, stratigraphically underlying the marker marble. The sulphide horizon was not found in the North Grid area.

Frisby Ridge South Grid (Figure 15):

The Frisby Ridge South Grid covers favourable Unit 5 stratigraphy on the Frisby 2 and Frisby 3 claims. The stratigraphic sequence and rock descriptions are similar to those on the North Grid, with a few notable exceptions. Scapolite was not noted in the marker marble, and garnets are more common in the stratigraphically underlying schists. The extrusive carbonatite-marker marble sequence is well exposed in cliffs around TL 36+50E, 64+00N, where the carbonatite is estimated to be 2-3m thick.

Two carbonatite samples from the South Grid contained similar amounts of light rare-earth elements to samples from the Cliff zone on the King Fissure Deposit. The sulphide layer was not found in the South Grid area.

Big Slide Zone (Figure 11, 12, 14):

In the Big Slide zone the favourable unit 5 stratigraphy, including the extrusive carbonatite (Unit 5c), marker marble (Unit 5m), and massive sulphide layer (Unit 5s), is well exposed in cliffs (Figure 12). On the Big Slide grid in the southern end of the zone, highly deformed, grey calcareous schists outcrop both above and below massive sulphide exposures.

South Frisby Ridge:

Traverses in this area did not locate the favourable Unit 5 stratigraphy where indicated on regional geological maps. Samples of pyritic quartzites from old trenches were not anomalous (FRS-SL1, SL2).

2.3.2 Structure

The dominant structures on Frisby Ridge are repetitive Phase 1 folds with southerly dipping axial planes. The folds plunge at low angles towards $070-110^{\circ}$, and are best outlined by the marker marble (Figures 3, 15, and 29). Shallowly dipping limbs combine with relatively gentle topography along the ridge top to create sinuous outcrop patterns. Complex folding with thickened hinge zones is visible in cliffs along the west side of the ridge. A northeasterly trending, southerly dipping thrust fault cuts across Frisby Ridge near BL 52+00N, and is most likely related to the larger Frisby Ridge fault a short distance to the south (Figure 3). The fault area is marked by a large area of brecciation, silicification, and ankeritization.

2.3.3 Mineralization

Massive sulphides (Unit 5s) outcrop at the base of cliffs along the western edge of Frisby Ridge in the Big Slide zone (Figures 11-14). Mineralogy consists of massive sphalerite, galena, pyrrhotite, pyrite, and marcasite in a siliceous or calcareous gangue. Coarse green amphibole crystals are locally intergrown with the massive sulphides. No massive sulphides (Unit 5s) were noted in the Frisby Ridge North or South Grid areas. This, combined with the decreased thickness of the carbonate units and presence of scapolite in the marker marble, suggests shallow water deposition, distal from sulphide and carbonatite sources.

Silicification and quartz veining containing minor chalcopyrite and malachite occur within complexly folded Unit 5m marble in the South Grid area. A sample of this material contained 0.48% Cu and 460ppb Au (sple FRS-CU1).

3.0 Geochemistry

3.1 Copeland and Frisby Groups Rock Geochemistry

Significant results from massive sulphide and extrusive carbonatite samples have been summarized in previous sections. Additional rock samples taken from fault zones, veins, and distinctive stratigraphic units were analyzed by ICP and whole rock methods. These results, along with those from past exploration programs, will help form a geochemical database for depositional environment modelling and comparison with other mineral deposits.

3.2 Frisby Group Soil Geochemistry

A soil geochemical survey totalling 173 samples was conducted over the Frisby Ridge South Grid. Samples were collected from the red-brown "B" Horizon, which was usually well developed between 10-20cm depth, and placed in standard kraft paper soil envelopes. Lab preparation consisted of drying then sieving samples to -80 mesh. All samples were analyzed at Acme Labs of Vancouver, B.C., by ICP methods for 35 elements; a summary of the analysis procedure is included with results in Appendix 3.

Nine elements representing a base-metal plus rare-earth element suite were examined in detail: they are Pb, Zn, Ag, Ba, Sr, La, Y, Nb, and Sc. Extreme high values were discarded from the data base before statistical analysis.

	Pb ppm	Zn ppm	Ag ppm	Ba ppm	Sr ppm	La ppm	Y ppm	Nb ppm	Sc ppm
# samples	173	173	173	169	172	172	173	172	172
Mean value	24.4	83.2	0.14	603	214	43.5	22	19.5	9.94
Maximum value	153	715	0.6	977	505	105	87	104	30.2
Minimum value	2	31	0.1	235	132	14	9	3	5
Std. Deviation	15.7	58.1	0.08	138	56	15.5	9.24	9.36	3.03
Variance	249	3376	0.0	18959	3141	242	85.4	87.6	9.22

Values above the mean plus one- and two standard deviations were considered possibly and definitely anomalous, respectively. Each element was contoured separately (Figures 16-24, 26-28).

Lead anomalies, with thresholds of 40ppm and 56ppm, are well constrained. Distribution of anomalous Zn (141ppm, 200ppm) and Ag (0.3ppm, 0.4ppm) is more widespread, with many apparently random single sample anomalies throughout the entire grid area. Barium (741ppm, 878ppm) and strontium (270ppm, 326ppm) display similar distribution patterns, with several anomalies scattered in the southern half of the grid. Lanthanum (59ppm, 75ppm), ytrium (31ppm, 40ppm), niobium (29ppm, 38ppm), and scandium (13ppm, 16ppm) all exhibit similar distribution patterns. Of these elements, Nb and Y show the greatest dispersion, while Sc shows the least.

Two multi-element anomalies exist on the Frisby Ridge South Grid. The first, bounded by L65+50 and L63+50N, consists of Pb, Zn, Ag, Sr, La, Y, Nb, and Sc. The anomaly trends approximately 060° with dimensions of 250m long by 50m wide, and is open both to the east and west. While the extrusive carbonatite (Unit 5c) outcrops within the anomaly area, no appreciable sulphides were noted during geological mapping and prospecting. The combination of base-metal and rare-earth element constituents of the anomaly indicates that the sulphide horizon and the extrusive carbonatite are probably within 50m of each other.

The second multi-element soil anomaly on the Frisby Ridge South Grid, occurring between L66+25N and L68+50N, consists of La, Y, Nb, and Sc. With dimensions of approximately 375m long by 30m wide, the anomaly forms a rough "S" shape. The westernmost finger of the anomaly, seen in the distribution of Y and La, is interpreted to be a downslope dispersion effect. The anomaly is successful in tracing the carbonatite for 150m north of the last outcrop in the area.

A single sample anomaly lies at BL 35+00E, 53+50N. The sample is highly elevated in Pb (153 ppm), Zn (715 ppm), La (1048 ppm), Ba (1098 ppm), Y (67 ppm), Nb (1125 ppm), and Sc (127.7 ppm) content. Although no massive sulphide or carbonatite outcrops were noted, the favourable sulphide-host carbonate stratigraphy occurs within 50m of the sample location.

4.0 Geophysics

Magnetic and VLF-EM surveys were conducted over grids on both the Copeland and Frisby groups. Instruments used were a Scintrex MP2 magnetometer and a Ronka EM-16 VLF receiver. Readings were taken at 25m intervals along grid lines and recorded in a field notebook. When time permitted, or anomalous zones were encountered, readings were taken at 12.5m intervals. For each magnetic survey, readings were taken at a base station at intervals throughout each day. Significant diurnal variations requiring correction occurred only during the Cliff Zone survey on the King Fissure Deposit. Corrections for diurnal variation were calculated by graphing the diurnal variation as a function of time, then interpolating appropriate correction factors for each reading. The Seattle VLF transmitting station was used for all VLF-EM surveys.

4.1 Copeland Group

A magnetic survey was conducted over the Cliff Zone grid, while both magnetic and VLF-EM surveys were run on the East Zone grid. Data for the East Zone was interpreted separately by a consulting geophysicist, D. R. MacQuarrie. The complete interpretation is supplied in Appendix 6.

4.1.1 Magnetic Surveys

For the East Zone (Figure 9), a contoured plan of the relative magnetic values reveals a possible fold closure between. L27+00E and L29+00E. Such a structure, consistent with geological mapping, would be the eastern closure of the Copeland synform, although definition is limited with the 100m spaced grid lines. Three probable pyrrhotite concentrations have been identified at the north end of L25+00E, at 9+87.5 to 10+00N on L26+00E, and at BL 10+00N on L27+00E. These zones are near surface with interpreted widths of 10-15m, based on the 1/2 width of the magnetic profile. On L27+00E, L30+00E, and probably L25+00E, magnetic highs coincide with positive fraser filter VLF-EM anomalies, suggesting that the pyrrhotite is massive over reasonably continuous distances (MacQuarrie, Appendix 6).

Magnetic data from the Cliff Zone grid (Figure 7) required correction for diurnal variation. Corrected values are still widely varied from one station to the next, and as a result the entire survey should be regarded with some caution. The plan map was contoured using relative values of 100, 200, 500, & 1000 gammas above the mean. Single station and spurious anomalies were discarded. A 200m long positive anomaly at the southern edge of the grid between L10+00E and L12+00E corresponds to mapped exposures of pyrrhotite-rich massive sulphides. As the anomaly diminishes to the west the sulphide layer, with pyrrhotite becoming subordinate to sphalerite, is covered by snow and ice. At the eastern end of the anomaly the sulphide layer becomes covered by ice and glacial debris. The north-central part of the grid contains widely varied readings with extreme lows and highs. This area is underlain by Unit 6 rusty biotitesillimanite schist with lesser quartzites and amphibolites. These extremes appear to be suppressed in the northeast part of the grid where Unit 6 is covered by snow and ice.

4.1.2 VLF-EM Survey

On the East Zone Grid, a Fraser filtered plan map of the in phase % data reveals a 300m long, arcuate conductor at the north ends of L26+00E and L27+00E (Figure 10 and Appendix 6). The conductor is open to the north. While coinciding with massive sulphides observed on L27+00E, much of the conductor is over stratigraphy directly below the massive sulphide horizon. Possible disseminated sulphides below the massive sulphide layer are indicated (MacQuarrie, Appendix 6).

4.2 Frisby Group

4.2.1 Magnetic Survey

A magnetic survey was conducted over 0.485km of grid in the Big Slide zone (Figure 13). Readings were taken every 5m along grid lines, and no corrections for diurnal variation were required. Test readings taken directly over massive sulphide exposures were magnetic lows (<57931 gammas). Contours of standard deviations from the mean show a 150m long linear anomaly extending southwest along the projected strike of the massive sulphide layer. The anomaly consists of 2 magnetic lows (<57931 gammas) separated by a magnetic high (>58012 gammas), and is open to the southwest.

4.2.2 VLF-EM Survey

A 3.0km VLF-EM survey was conducted over the southern portion of the Frisby Ridge South Grid (Figure 25). Results do not reveal any significant anomalies or trends.

5.0 Recommendations

5.1 Copeland Group

In the Cliff Zone, a channel sampling program of the massive sulphides is recommended to complement a comprehensive diamond drill program. Drilling should systematically test downdip width and grade continuity of the zone.

In the East zone, a well-controlled grid with 50m spaced lines should be established. Work on the grid should include the geophysical surveys recommended by D. R. MacQuarrie (Appendix 6) in conjunction with detailed geological mapping and sampling. Compilation of these surveys will assist in planning of a diamond drill program. This drill program should be designed to thoroughly test the keel area of the Copeland synform for potential tectonic thickening.

The Northeast zone warrants detailed geological mapping, sampling, and geophysical surveys. This program should be conducted as an extension of grid work in the East zone.

The West zone, with the thickest known exposures of massive sulphides plus barite, warrants aggressive exploration. A program of detailed sampling, geological mapping, and geophysical surveys is necessary to define drill targets.

The Lake zone warrants detailed geological mapping, sampling, and geophysical surveys.

The extrusive carbonatite layer is present throughout the entire King Fissure Deposit, and as yet is poorly understood. Geological mapping, detailed sampling, and petrographic studies are required to better understand the distribution and economics of Nb plus light rare-earth elements present.

Reported sulphide showings in the North Copeland ridge area remain unexplored and should be located and evaluated.

5.2 Frisby Group

In the Big Slide zone, massive sulphide exposures merit further geological mapping, detailed sampling, and geophysical surveys. The zone remains unexplored southwest to the Jordan River. Surface exploration in this area should begin with establishment of a cut-line grid followed by geological mapping, rock and soil sampling, and geophysical surveys. Cost-effective access may be made possible by upgrading and extending the old pack trail from the road along the Jordan River into this area.

The first South Grid soil geochemical anomaly, bounded by 63+50N and 65+50N, should be re-examined and sampled. The area was anomalous in Pb, Zn, and Ag, and is underlain by the massive sulphide host stratigraphy. The area of the highly anomalous soil sample located at 53+50N on the Frisby Ridge baseline should also be examined and sampled in detail.

On the Frisby Ridge North and South grids, the extrusive carbonatite layer merits further evaluation as per that recommended for the King Fissure area.

6.0 ITEMIZED COST STATEMENTS

COPELAND GROUP

Wages: T. Clarke, geologist (Aug. 23-28, Sept. 6-9)		
10 days @ \$200/day	\$	2,000.00
J. Laird, prospector (Aug. 23-28, Sept. 6-9) 10 days @ \$200/day	\$	2,000.00
Food & Accommodation: 2-man camp (Aug. 25-27, Sept. 7-9) 8 days @ \$25/day	\$	200.00
Hotel (Aug. 23, Sept.6)	\$	138.79
Food/meals (Aug. 23-28, Sept. 6-9)	\$	449.16
Transportation		
Truck rental (Aug. 23, 28) 1200km x \$0.25/km gas, highway toll	\$ \$	
Helicopter charter (Aug. 24-28, Sept. 7-9) 5.5 hours @ \$700/hour	\$	3,850.00
Geophysical Equipment Rental: Scintrex MP-2 Proton Precession Magnetometer		
and Geonics EM-16 VLF-EM Receiver (Sept. 1-14) 0.5 month @ \$1,836.00/month	\$	918.00
Geochemical Analyses: Min-En Labs:		
7 samples assayed for Ag, Ba, Pb, Zn @ \$29.75/sample	\$	208.25
8 samples assayed for Pb, Zn @ \$13.25/sample 2 samples assayed for Ba @ \$10.00/sample	\$ \$	106.00 20.00
17 sample prep. @ \$3.75/sample 1 rock geochem. 31 element ICP @ \$6.00/sample	\$ \$	63.75 6.00
2 geochem. F @ \$4.50/sample 7 geochem. 26 element major ICP @ \$13.50/sample	\$ \$	9.00 94.50
1 rock geochem. Ag @ \$2.50/sample	ې \$	94.50 2.50

Acme Labs:		
3 sample prep. @ \$3.25/sample 3 samples analyzed for 35 element ICP	\$	9.75
@ \$6.50/sample	\$	19.50
3 Au analyses by acid leach @ \$5.00/sample	\$ \$	15.00
1 Cu assay @ \$7.50/sample	\$	7.50
Chemex Labs:		
9 assay prep. @ \$4.35/sample 9 whole rock + immobile element analyses	\$	39.15
@ \$28.00/sample	\$	252.00
4 rock samples analyzed for light rare-earth elements plus ultra-trace 19 @ \$55.50/sample	\$	222.00
sub-total	\$	1,074.90
G.S.T.		•
Total lab costs		1,150.14
Report preparation, drafting, supplies	\$	1,500.00

TOTAL COPELAND GROUP EXPLORATION EXPENDITURES \$12,625.00

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FRISBY GROUP

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Wages: T. Clarke, geologist (Sept. 9-29) 21 days @ \$200/day	\$	4,200.00
J. Laird, prospector (Sept. 9-29) 21 days @ \$200/day	\$	4,200.00
M. Andrews, geologist (Sept. 14-29) 15 days @ \$160/day	\$	2,400.00
Food & Accommodation: 2-man camp (Sept. 9-13) 5 days @ \$25/day 3-man camp (Sept. 15-20, 22-27) 12 days @ \$35/day	\$ \$	
Hotel (Sept. 14,21, 28)	\$	346.97
Food/meals (Sept. 10-29)	\$	911.94
Transportation: Truck rental (Sept. 6, 28, 29) 2400km x \$0.25/km gas, highway toll Helicopter charter (Sept. 10-28)	\$ \$	
5 hours @ \$700/hour	\$	3,500.00
Geophysical Equipment Rental: Scintrex MP-2 Proton Precession Magnetometer and Geonics EM-16 VLF-EM Receiver (Sept. 15-30) 0.5 month @ \$1,836.00/month	\$	918.00
Geochemical Analyses: Min-En Labs: 2 rock sample prep. @ \$3.75/sample 2 rock geochem. 31 element ICP @ \$6.00/sample 2 rock geochem. Au (fire) @ \$7.25/sample	\$ \$ \$	7.50 12.00 14.50
Acme Labs: 173 soil samples analyzed for 35 element (ICP)		

0	6.50/sample	\$	1,124.50
1	73 soil sample prep. @ \$1.00/sample	\$	173.00
	6 rock sample prep. @ \$3.25/sample	\$	84.50
20	5 samples analyzed for 35 element ICP		
0	\$6.50/sample	\$	169.00
2	6 Au (acid leach) @ \$5.00/sample	\$	130.00
2	geochem. Au, Pt, Pd @ \$8.50/sample	\$	17.00
	Cu assays @ \$7.50/sample	\$	15.00
7	Pb, Zn assays @ \$10.50/sample	\$ \$ \$	73.50
	Zn assay @ \$7.50/sample	\$	7.50
4	Labs: sample prep. @ \$4.35/sample whole rock analyses @ \$28.00/sample rare-earth 10 plus ultra-trace-19 analyses	\$ \$	112.00
0	\$55.50/sample	\$	222.00
	sub-total G.S.T. <u>Total lab costs</u>	\$	•
Report	preparation, drafting, supplies	\$	1,500.00

TOTAL FRISBY GROUP EXPLORATION COSTS

\$21,691.79

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APPENDIX 1

QUALIFICATIONS

.

STATEMENT OF QUALIFICATIONS

I, Tiro Clarke, hereby certify that:

- 1. I am a Geologist residing at 301-357 East 2nd Street, North Vancouver, British Columbia, V7L 1C6
- I am a graduate of the University of British Columbia with a B.Sc.(Hon.) in Geology and Oceanography, 1988.
- 3. I have practised geology since 1986, and mineral exploration geology continuously since graduation.
- 4. I personally conducted or supervised fieldwork described in this report.
- 5. I currently own 500 shares of First Standard Mining, Ltd., purchased in 1990. I do not plan to acquire or receive any other interest, direct or indirect, in First Standard Mining Ltd. or the Jordan River Property.
- 6. I consent to the use of this report, or excerpts therefrom, in any prospectus, statement of material facts, or compilation as required by First Standard Mining Ltd..

Dated at Vancouver, British Columbia, this 5th day of December, 1991.

The Clarke

Tiro Clarke, B.Sc. (Hon.)

STATEMENT OF QUALIFICATIONS

I, James W. Laird, hereby declare that:

- 1. I reside and maintain a business office at 3869 Mount Seymour Parkway, North Vancouver, B.C., V7G 1C4.
- 2. I am a self-employed prospector and mining exploration contractor and have been so full-time for 11 years.
- 3. I have completed the B.C. Department of Mines course "Advanced Mineral Exploration for Prospectors", 1980.
- I am a member in good standing of the Canadian Institute of Mining and Metallurgy and the B.C. and Yukon Chamber of Mines.
- 5. I have extensively explored the Kootenay-Revelstoke area for mineral deposits for several years, and am very familiar with the geology and mines thereof.
- I currently own 3500 shares of First Standard Mining, Ltd., purchased in 1990.

Dated at Vancouver, British Columbia, this 5th day of December, 1991.

James W. Laird

STATEMENT OF QUALIFICATIONS

I, Martin Andrews, hereby certify that:

- 1. I am a geologist residing at 203-2105 West 7th Street, Vancouver, British Columbia, V6K 1X9.
- I am a Graduate of the University of British Columbia with a B.Sc. in Geology (1989).
- 3. I personally conducted fieldwork described in this report.
- 4. I do not hold, nor plan to acquire or receive any interest, direct or indirect, in First Standard Mining Ltd. or the Jordan River Property.
- 5. I consent to the use of this report, or excerpts therefrom, in any prospectus, statement of material facts, or compilation as required by First Standard Mining Ltd..

Martin Andrews, B.Sc.

APPENDIX 2

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ROCK GEOCHEMISTRY RESULTS

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• EN LABORATORIES (DVISION OF ASSAYERS CORP.)

SPECIALISTS IN MINERAL ENVIRONMENTS CHEMISTS • ASSAYERS • ANALYSTS • GEOCHEMISTS **VANCOUVER OFFICE:**

705 WEST 15TH STREET NORTH VANCOUVER, B.C. CANADA V7M 1T2 TELEPHONE (604) 980-5814 OR (604) 988-4524 FAX (604) 980-9621

SMITHERS LAB.: 3176 TATLOW ROAD SMITHERS, B.C. CANADA VOJ 2NO TELEPHONE (604) 847-3004 FAX (604) 847-3005

Assay Certificate

1V-0967-RA1

Company: LAIRD EXPLORATION Project: JORDAN RIVER Atta: JAMES LAIRD Date: SEP-05-91

Copy 1. LAIRD EXPLORATION, NORTH VANCOUVER, B.C.

He hereby certify the following Assay of 7 ROCK samples submitted AUG-30-91 by J.LAIRD.

Sample - Number		AG g/tonne	AG oz/ton ,	BA %	PB 7X	ZN %	F PPM	
EZ-1	2	87.5	2.55	.01	12.90	5.70	· .	
EZ-2		57.5	1.68	.01	3.90	18,90		
CZ-1		324.0	9.45	.01	29.80	25.70		
NEZ-1	J.	71.5	2.09	21.54	10.40	2.25		
NEZ-2		18.9	.55	1.45	2.05	2.74		
NEZ-3		78.2	2.28	43.53	14.20	. 19	40	
NEZ-4		29.3	.85	3.25	3.30	9.50		
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Certified by

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COMP: LAIRD EX PROJ: JORDAN R ATTN: J.LAIRD										705 Wi	EST 1! ((5TH S1 604)98	r., NO 30-581	Arthiv 14 or	- IC) (ANCOUN (604)	/ER, B 788-45	.C. V 24	7M 1T										± p	DAT	E: 91, (AC)	67-RJ1 /09 /06 T:F 31)
SAMPLE NUMBER	AG PPM	AL		8 PPM	BA PPN	BE	B I PPM	CA	CD PPM	CO PPM	CU	FE PPH	K PPM	LI	MG PPM	MN PPM	HO PPM	NA PPM	NI PPM	P PPN	P8 PPN P	SB PPM F	SR PPM P	TH 1 PM PF	TI Phi pe	V ZI M PPI	N GJ N PPH	SN PPM	W C PPM PP	R AU-I H	FIRE
EZ-3 FR-1 FR-2	.6	3130 15910 3510	16	1	1570 878 1865	.4	2	14970 11690 69570	1	18	17 1	14810	080	1	680 9390 52390	310	5 3 1	10 90 10	9 27 4	240 1110 170	868 529 255	1 5 1 27	20 19 795	1 5 5 55 1 3	5 6. 1 20. 1 20.	4 81 7 24 6 13	1 1 1 2 5 1	1	10 24	8 1	1
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المحمولات المراجعة ا المراجعة الم

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ACY . ANALYTICAL LABORATORIES LTD.

852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6

GEOCHEMICAL ANALYSIS CERTIFICATE

PHONE(604)253-3158 FAX(604)253-1716

James W. Laird File # 91-4986

3868 Mt. Seymour Parkway, North Vancouver BC V7G 1C4

SAMPLE#	Мо	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	v	Ca	P	La	Cr	м	g Ba	Ti	A1	Na	κ	W	Zr	Sn	Y	Nb	Be	Sc
	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	*	%	ppm	ррт		% ppm	%	*	*	%	ppm	ppm	ppm	ppm	ppm	ppm	Ppr
RS-1	6	40	32	5560	. 3	37	15	719	5.78	5	5	ND	6	535	1.9	2	2	73	11.16	.039	27	82	3.3	4 3069	. 19	6.20	. 39	2.16	2	6	1	16	13	5.2	8.0
RS-2	3	28	38	5296	. 4	30	11	337	2.27	2	5	ND	4	768	1.3	3	2	58	20.51	.025	17	71	2.3	1 1297	. 12	5.08	.23	1.44	2	6	1	11	10	1.7	6.9
RS-3	2	30	26	5752	.6	21	8	252	1.79	2	5	ND	1	644	1.5	2	2	43	18.48	.022	13	56	2.3	1 651	. 10	4.05	. 30	.93	2	4	1	12	12	1.4	5.
RS-4	2	24	22	2085	. 4	24	12	524	2.80	2	5	ND	5	581	.5	2	2	41	13.15	.015	30	73	2.1	4 726	. 18	6.50	. 40	1.68	2	5	1	15	15	2.3	9.
RS-5	2	34	20	391	.4	28	12	303	3.12	4	5	ND	4	506	.4	4	2	40	13.85	.018	27	73	2.1	3 1155	. 18	5.93	. 37	1.64	2	4	1	11	15	2.4	8.
RS-6	23	58	714	22473	.7	19	26	189	28.00	19	5	ND	3	35	9.2	2	2	29	. 40	.013	2	20	.2	1 2171	.05	1.37	.04	. 30	2	2	1	3	1	15.3	1.
RS-7	25	61	1921	34604	.8	22	23	273	26.66	16	7	ND	4	45	12.7	2	2	42	. 34	.010	2	38	.2	5 1935	.07	1.55	.07	.43	2	4	1	4	4	13.3	1.
RS-8	17	111	3866	61631	.7	73	19	478	25.94	18	5	ND	1	31	24.0	6	2	35	.24	.005	2	44	.0	9 2345	.02	. 48	.03	. 16	2	1	1	2	1	18.3	
FRS-9	10	47	269	733	.2	24	11	337	1.78	9	5	ND	4	634	.6	2	2	110	15.83	.022	18	45	2.3	0 2889	. 12	3.87	.28	1.33	2	7	1	11	13	1.7	5.
FRS-10	3	35	43	194	.2	36	18	561	4.09	2	5	ND	10	473	.5	2	2	92	5.00	.035	55	131	2.5	9 3460	. 33	10.24	.72	2.30	2	9	1	20	17	5.9	14.9
RS-11	2	18	104	403	. 1	30	10	652	2.27	8	5	ND	6	710	1.0	2	2	83	12.59	.047	29	70	2.2	7 549	. 20	5.35	. 42	1.39	2	7	1	15	14	2.1	7.
RS-12	15	203	2081	24911	3.5	104	38	407	40.35	5	5	ND	5	73	16.2	2	2	36	. 47	.005	3	29	.1	5 986	.03	. 96	.04	. 30	2	1	1	4	1	19.8	•
RS-13	51	69	8991	31178	6.7	16	8	360	6.18	26	5	ND	5	335	41.6	18	27	136	2.74	. 029	30	74	1.2	5 1299	. 19	4.92	.28	2.06	2	5	1	14	14	6.4	7.
RS-14	17	80	5591	99999	. 1	39	17	1293	17.96	48	5	ND	1	37	115.4	2	2	19	.27	.012	2	15	i .1	5 1097	.02	. 54	.03	.25	2	2	1	2	1	9.4	
RS-15	12	113	5994	99999	2.6	73	25	706	29.47	31	5	ND	3	19	57.4	2	3	14	.04	.013	2	1	.0	3 432	.01	.28	.02	. 12	2	2	1	1	1	16.0	•
FRS-16	2	3285	32	728	6.6	10	9	1088	1.15	2	5	ND	1	182	1.2	2	2	20	10.29	.013	90	35	i .1	5 247	.05	1.99	.07	.77	2	1	1	32	9	.7	7.
FRS-17	18	168	4381	60669	1.5	69	9	345	30.93	16	5	ND	2	32	19.8	15	2	37	. 18	.004	2	99	.0	6 2188	.03	.53	.01	.01	2	1	1	2	1	19.0	
FRS-CU 1	3	4833	74	948	. 9	10	3	110	1.16	2	5	ND	1	2	.8	4	6	1	.04	.014	4	20) .0	1 92	.01	. 22	.02	.08	2	1	1	5	1	. 3	
FRS-QV 1	3	162	7	101	. 5	27	8	89	. 96	2	8	ND	1	10	. 3	2	2	5	.21	.007	2	19	.0	9 40	.01	. 40	.03	. 10	2	1	1	1	2	.2	
RS-FZ 1	4	66	41	298	.2	100	30	422	7.70	4	5	ND	23	312	.2	8	2	144	.08	.050	65	216	5 .5	7 1789	. 36	8.69	1.59	3.83	2	3	1	12	11	7.6	14.
FRS-SL 1	15	20	11	119	.5	69	67	195	5.15	6	5	ND	1	6	.2	3	2	5	.01	.002	з	31		1 59	.01	. 18	.02	.07	2	2	1	1	1	3.5	
RS-SL 2	4	24	11	74	1.0	29	48	165	1.63	7	5	ND	11	13	.2	2	2	9	.02	.014	5	41	.0	1 40	.02	. 30	.01	.07	4	1	1	5	19	. 2	
E FRS-CU 1	3	5102	72	930	2.1	15	3	107	1.19	2	5	ND	1	2	1.0	2	3	1	.04	.006	5	20) .0	1 96	.01	.22	.01	.08	2	1	1	5	1	. 2	
RN-1	2	108	52	203	3.1	43	20	460	4.75	2	5	ND	7	474	. 5	2	2	81	7.18	.051	48	162	2 1.2	5 305	. 29	10.80	.25	.89	2	3	1	16	11	5.3	14.
RN-2	5	102	34	126	1.4	94	38	351	5.44	3	5	ND	9	410	.7	7	2	159	6.90	.058	56	181	1.6	4 803	.43	12.40	.61	2.49	3	4	1	18	14	5.8	17.
RN-3	5	64	84	148	3.5	61	24	310	4.68	2	5	ND	7	549	.4	2	2	82	6.11	.035	42	153	3 1.7	5 897	. 35	10.41	.84	2.47	2	3	1	15	11	3.1	14
F-CU 1	3	12642	7	1	.8	13	6	309	1.43	2	5	ND	1	30	1.5	2	8	2	2.21	.014	5	22	2.0	3 28	.01	. 35	.02	.09	2	1	1	16	1	. 2	: 1
Z-FZ 1	1	120	40	146	1.5	100	35	848	7.79	21	5	ND	6	558	1.4	2	2	121	7.05	.088	29	106	5 1.4	0 467	. 12	7.56	2.45	1.95	2	1	1	25	1	6.5	11
(F-4	4	28	49	60	11.1	24	7	1112	4.61	8	5	ND	1	353	.8	2	2	154	18.60	.135	19		3.5			1.70	.06	.71	2	1	3	14	5	1.6	2
B-1	28	137	47	328	.8	77	19	470	3.54	4	5	ND	5	535	2.6	11	2	401	3.67	. 183	65	155	5 1.1	7 2764	. 34	8.18	2.73	1.32	2	2	1	28	178	1.5	10
TANDARD HEC	23	61	44	148	7.5	103	52	1306	4.81	41	25	7	36	60	20.0	17	21	84	.64	. 124	38	112	2 1.2	1 224	. 08	2.11	.08	. 17	11	4	17	8	3	. 2	: 6

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 10ML HCLO3-HNO3-HF AT 200 DEG. C TO FUMING AND IS DILUTED TO 10 ML WITH DILUTED AQUA REGIA. THIS LEACH IS PARTIAL FOR MAGNETITE, CHROMITE, BARITE, OXIDES OF AL, ZR & MN AND MASSIVE SULFIDE SAMPLES. AU DETECTION LIMIT BY ICP IS 3 PPM. AS, CR, SB SUBJECT TO THE LOST OF VOLATILIZATION DURING HCLO4 FUMING. Samples beginning 'RE' are duplicate samples. - SAMPLE TYPE: ROCK

DATE RECEIVED: OCT 9 1991 DATE REPORT MAILED: Ot 11/91 SIGNED BY.... D. TOYE, C.LEONG, J.WANG; CERTIFIED B.C. ASSAYERS

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716 **GEOCHEMICAL ANALYSIS CERTIFICATE** James W. Laird FILE # 91-4986 3868 Mt. Seymour Parkway, North Vancouver BC V7G 1C4 Pb Zn Pd** Cu SAMPLE# Au* Au** Pt** જ ૪ ppb 8 ppb ppb ppb FRS-1 7 -----2 _ -FRS-2 FRS-3 2 -----4 _ FRS-4 _ _ _ 4 FRS-5 2.87 .11 FRS-6 11 3.86 FRS-7 20 --.30 11 _ -.51 6.92 FRS-8 --FRS-9 4 _ FRS-10 2 FRS-11 4 _ -.26 2 1 2.92 FRS-12 2 1 1.29 3.50 FRS-13 16 _ --.76 32.25 FRS-14 34 --.74 16.23 FRS-15 77 23 .32 FRS-16 --6.89 1 4 1 _ **FRS-17** 4 -460 .48 FRS-CU 1 --FRS-QV 1 10 ---FRS-FZ 1 7 _ -. FRS-SL 1 7 _ -FRS-SL 2 21 RE FRS-CU 1 390 _ --FRN-1 5 _ _

- SAMPLE TYPE: ROCK

STANDARD AU-R

FRN-2

FRN-3 CF-CU 1

KF-4

FB-1

CZ-FZ 1

Samples beginning 'RE' are duplicate samples.

483

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494

1.19

.85

1.36

2.39

DATE REPORT MAILED: Oct 17/91. DATE RECEIVED: OCT 9 1991 SIGNED BY. D. TOYE, C.LEONG, J.WANG; CERTIFIED B.C. ASSAYERS

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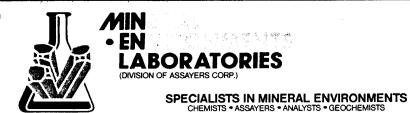
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VANCOUVER OFFICE: 705 WEST 15TH STREET NORTH VANCOUVER, B.C. CANADA V7M 1T2 TELEPHONE (604) 980-5814 OR (604) 988-4524 FAX (604) 980-9621

SMITHERS LAB .:

3176 TATLOW ROAD SMITHERS, B.C. CANADA VOJ 2NO TELEPHONE (604) 847-3004 FAX (604) 847-3005

1V-1261-RG1 Certificate Analysis Geochemical

Company:	LAIRD	EXPLO	ORATION	
Project:	JORDAN	RIVER	PROJECT	1991
Attn:	JAMES L	AIRD		

Date: OCT-18-91 Copy 1. LAIRD EXPLORATION, NORTH VANCOUVER, B.C.

He hereby certify the following Geochemical Analysis of 9 ROCK samples submitted OCT-09-91 by JAMES LAIRD.

Sample Numb er	AU-FIRE PPB	AG PPM	F PPM	
EZG-1 EZG-2 EZG-3 EZG-4 EZG-5	21 13 4 128 59			
CZ-2 CF-1 WZ-1 WZ-2	61 23 18 76	132.0	30	

Certified by

MIN-EN LABORATORIES



SPECIALISTS IN MINERAL ENVIRONMENTS CHEMISTS • ASSAYERS • ANALYSTS • GEOCHEMISTS VANCOUVER OFFICE:

705 WEST 15TH STREET NORTH VANCOUVER, B.C. CANADA V7M 1T2 TELEPHONE (604) 980-5814 OR (604) 988-4524 FAX (604) 980-9621

SMITHERS LAB.: 3176 TATLOW ROAD SMITHERS, B.C. CANADA VOJ 2NO TELEPHONE (604) 847-3004 FAX (604) 847-3005

Assay Certificate

Company: LAIRD EXPLORATION

Project: JORDAN RIVER PROJECT 1991 Attn: JAMES LAIRD Date: OCT-22-91

Copy 1. LAIRD EXPLORATION, NORTH VANCOUVER, B.C.

He hereby certify the following Assay of 8 ROCK samples submitted OCT-09-91 by J.LAIRD.

Sample Number	BA %	PB %	ZN %	
EZG-1 EZG-2 EZG-4 EZG-5 CZ-2		6.50 .83 46.70 5.45 2.26	9.24 5.79 3.65 12.30 12.20	
CF-1 WZ-1 WZ-2	.89 40.92	79.20 1.20 8.10	1.22 11.35 1.70	

Certified by

MIN-EN LABORATORIES

1V-1261-RA1

I: JAMES L											980-58														* (A	
AMPLE JMBER	AL203	BA X	BE %	CAO X	CO (CR203	X	FE203	K20 %	X	MNO2 X	MO X	NA20 %	NB X	NI X	P205 %	PB %	RB X	\$102 %	SN X	SR X	T102 X	V X	W X	ZN X	Z
ZG-1 ZG-2 ZG-3 ZG-4 ZG-5	8.57 6.77 8.43 1.23	.780	.005	5.81 3.21 15.45 .10 1.61	.005 .005 .005 .005 .005	.03 .03 .12 .03 .03	.010 1 .005 2 .010 .035 .035 3	13.61 28.22 8.52 4.96	1.99 1.33 1.06 .19 .80	3.75 1.35 7.70 .27 .69	.26 .17 .08 .17 .28	.005 .005 .005 .005 .005	.01 .01 .31 .01 .01	.01 .01 .01 .01 .01		.05 .01 .45 .01 .03	1.045 .435 .220 45.280 .200	.01 .01 .01	39.96 37.94 49.35 30.56 27.48	.005 .005 .010 .005	.05 .01 .02 .01 .06	.38 .26 3.39 .05 .18	.005	.005	10.450 5.040 .420 3.575 12.250	.01
2-2 F-1	5.11 .16	.450 .005	.001 .001	5.16 .01	.005	.03	.005 1 .150	15.73 1.96	1.05 .08	1.49 .02	.20 .12	.005 .005	.03 .01	.01 .01	.005 .005	.01 .03	.130 77.490	.01 .01	48.18 1.76	.005 .005	.02 .01	.20 .01	.010 .005	.005 .005	13.135	.00
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Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 To: LAIRD, JAMES W.

3868 MT. SEYMOUR PARKWAY NORTH VANCOUVER, BC V7G 1C4 .

Page Number :1 Total Pages :1 Certificate Date: 22-OCT-91 Invoice No. :19123055 P.O. Number :

6

Project : JORDAN RIVER Comments:

									•		CE	ERTIFI	CATE	OF	ANALY	SIS	4	9123	055		
SAMPLE	PRE	-	A1203	Ba0 %	CaO %	Fe203	K20 ¥	Mg0 ¥	MnO %	Na20 %	P205 %	SiO2 %	Ti02 %	101 \$	TOTAL %	Cr ppm	ND ppm	Rb ppn	Sr ppm	Y ppm	2r ppm
CZ-CB1 CZ-CB2 CZ-GA1 FRN-CB1 FRN-SMB1	208 208 208 208 208 208	294 294 294	13.87 13.57 18.68 10.29 0.10	0.12 0.03 0.26	18.06 22.03 5.56 24.35 37.66	5.09 4.38 20.35 4.99 0.12	2.18 1.86 1.49 1.60 0.02	3.30 3.07 1.75 4.42 2.55	0.03 0.04 0.64 0.26 0.02 <	0.97 2.32 0.62 2.96 (0.01	0.38	41.37 33.56 44.30 29.13 28.42	0.48 0.75 0.36	12.49 17.25 0.32 19.04 29.91	98.75 94.86 98.04	81 68 327 48 22	12 17 16 122 < 5	120 100 82 67 < 5	544 751 100 3390 237	26 23 60 27 < 5	88 87 467 166 20
FRS-CB1 FRS-CB2 KF-1 KF-2 KF-3	208 208 208 208 208 208	294 294 294	4.99 11.50 13.90 15.58 15.75	0.07 0.30 0.01 0.34 0.06	9.86	1.86 3.63 11.30 11.02 0.79	0.74 1.09 0.55 2.39 6.61	3.93 3.54 9.82 6.69 0.11	0.04 0.05 0.11 0.20 0.02	0.79 1.28 1.20 2.08 3.18	0.09 0.08 0.81 0.43 0.38	18.93 26.78 46.40 43.35 74.32	0.19 0.33 1.69 1.89 0.02	20.52 1.90 2.00	98.05 100.05 97.55 95.68 101.90	25 30 594 218 81	16 18 67 72 < 5	32 56 13 65 162	749 793 56 857 210	12 16 27 29 19	54 100 179 178 38
KP-5 NEZ-CB1 NEZ-CB2	208 208 208	294	10.85 14.01 9.56		7.83 17.39 24.45	7.08 3.35 4.02	8.89 1.43 1.20	6.07 4.12 4.48	0.14 0.12 0.32	0.56 4.18 3.82	1.60 0.13 0.37	44.84 38.37 29.16		9.07 14.36 21.08	98.06	201 47 22	43 80 146	183 64 66	3860 1575 2570	50 17 22	611 90 94
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CERTIFICATION:



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Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221

To: LAIRD, JAMES W.

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3868 MT. SEYMOUR PARKWAY NORTH VANCOUVER, BC V7G 1C4

Project : JORDAN RIVER Comments:

Page Number :1-A Total Pages :1 Certificate Date: 18-NOV-91 Invoice No. :19123057 P.O. Number : Account HDI

											CE	RTIFI	CATE	OF A	NAL	YSIS	ŀ	9123	057		
SAMPLE	PRI		ybu ybu	As ppa	Ba. pp a	Bi PP n	Cd ppm	Cu ppa	Fe %	Ga. ppm	Eg pp n	La ppm	Mn ppm	Mo	Pb ppm	Sb ppa	Sr ppa	T1 ppa	V PP m	M M	Zn ppa
CZ-CB1 CZ-CB2 FRN-CB1 FRN-SMB1 FRS-CB1	299 299 299	288 288 288 288 288 288	0.05	< 0.5 < 0.5 < 0.5	2290 1360 2910 50 830	0.4 0.4 0.2 0.2 0.2	0.6 0.2 0.4 < 0.1 0.5	33.5 15.5 14.5 < 0.5 14.0	3.85 3.08 3.55 0.28 1.33		0.2 0.9 < 0.1 < 0.1 0.3	80 80 300 90 100	330 370 2220 280 385	2.5 < 0.5 2.0 < 0.5 1.0	24 10 11 3 11	< 0.2 < 0.2 0.2 < 0.2 < 0.2 0.2	645 871 3780 349 867	5.0 1.5 1.0 2.0 2.0	184 105 50 4 104	10 10 < 10 < 10 < 10	187 114 97 3 57
FRS-CB2 NEZ-CB1 NEZ-CB2	299	288 288 288		0.5 0.5 0.5	3220 1850 2330	< 0.2 0.2 0.4	0.2 0.5 0.9	3.0 11.5 9.0	2.41 2.47 2.81	18	< 0.1 < 0.1 < 0.1	90 100 170	425 980 2440	0.5 1.0 6.5	47	< 0.2 < 0.2 < 0.2	902 1795 2680	2.0 2.0 4.0	84 55 29	10 < 10 10	95 103 120
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CERTIFICATION:

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Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221 To: LAIRD, JAMES W.

3868 MT. SEYMOUR PARKWAY NORTH VANCOUVER, BC V7G 1C4

Project : JORDAN RIVER Comments: Page Number :1-B Total Pages :1 Certificate Date: 18-NOV-91 Invoice No. :19123057 P.O. Number : Account :HDI

CERTIFICATE OF ANALYSIS

CERTIFICATION;

A9123057

SAMPLE	PREP CODE		Ce NA PP		NAA ppm	La NAA ppa	Lu NAA ppa	nd naa ppa	Sm NAA S	id naa ppm	Th NAA (ppm	J NAA M ppm	id naa ppm	
FRN-SMB1	299 2 299 2 299 2 299 2 299 2 299 2	88 88 88	70. 66. 698. 6. 68.	0 0 0 <	1.00 1.50 3.50 0.50 0.50	2.0	0.10 0.20 0.10 < 0.10 < 0.10	25 20 200 < 5 15		0.20 0.50 0.50 < 0.10 < 0.10	14.0 12.0 9.0 < 1.0 5.0	4.0 2.0 1.0 < 1.0 1.0	1.30 1.40 1.40 0.10 0.60	
FRS-CB2 NEZ-CB1 NEZ-CB2	299 2 299 2 299 2	88	60. 162. 488.	0		40.0 112.0 304.0	0.10 0.10 0.10	20 55 150	4.60 8.00 19.50	0.40 1.10 0.80	6.0 19.0 5.0	2.0 1.0 2.0	1.00 0.80 1.00	
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VANCOUVER OFFICE: 705 WEST 15TH STREET NORTH VANCOUVER. B.C. CANADA V7M 1T2 TELEPHONE (604) 980-5814 OR (604) 988-4524 FAX (604) 980-9621

SMITHERS LAB .:

3176 TATLOW ROAD SMITHERS, B.C. CANADA VOJ 2NO TELEPHONE (604) 847-3004 FAX (604) 847-3005

Assay Certificate

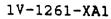
SPECIALISTS IN MINERAL ENVIRONMENTS CHEMISTS • ASSAYERS • ANALYSTS • GEOCHEMISTS

> Date: DEC-02-91 Copy 1. LAIRD EXPLORATION, NORTH VANCOUVER, BC.

He hereby certify the following Assay of 8 ROCK samples submitted OCT-09-91 by JAMES LAIRD.

Sample Number	AG g/tonne	AG oz/ton
EZG-1 EZG-2	62.6 11.0	1.83 .32
EZG-3	5.2	.15
EZG-4	375,0	10.94
EZG-5	61.8	1.80
cz-2	 2 9. 0	.85
WZ-1	15.7	. 46
WZ-2	68. 8	2.01

Certified by





Company:

Project: Attn:

LAIRD EXPLORATION JORDAN RIVER PROJECT 1991

JAMES LAIRD



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221

CERTIFICATE

A9123055

LAIRD, JAMES W.

Project: JORDAN RIVER P.O. # :

Samples submitted to our lab in Vancouver, BC. This report was printed on 25-OCT-91.

	SAM	PLE PREPARATION
CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION
208 294 200	13 13 13	Assay ring to approx 150 mesh Crush and split (0-10 pounds) Whole rock fusion
* NOTE	1:	

Code 1000 is used for repeat gold analyses It shows typical sample variability due to coarse gold effects. Each value is correct for its particular subsample. To: LAIRD, JAMES W.

3868 MT. SEYMOUR PARKWAY NORTH VANCOUVER, BC V7G 1C4

Comments:

A9123055

		ANALYTICAL	PROCEDURES		
CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION	METHOD	DETECTION LIMIT	upper Limit
CODE 594 542 588 586 821 593 596 599 597 592 595 475 540 894 973 1067 898 974 978	SAMPLES 13 13 13 13 13 13 13 13 13 13	DESCRIPTION Al203 %: Whole rock BaO %: Whole rock CaO %: Whole rock Fe2O3(total) %: Whole rock MgO %: Whole rock MaO %: Whole rock Na2O %: Whole rock SiO2 %: Whole rock SiO2 %: Whole rock L.O.I. %: Loss on ignition Total % Cr ppm Nb ppm Rb ppm Sr ppm Zr ppm	METHOD ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES ICP-AES FURNACE CALCULATION ICP ICP	LIMIT 0.01 0.01 0.01 0.01 0.01 0.01 0.01 0.01 0.01 0.01 0.01 50 5 50 5 5 5 5 5	LIMIT 99.99 99.99 99.99 99.99 99.99 99.99 99.99 99.99 99.99 99.99 105.00 N/A 10000 N/A 10000 10000



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver British Columbia, Canada V7J 2C1 PHONE: 604-984-0221

CERTIFICATE

A9123057

LAIRD, JAMES W.

Project: JORDAN RIVER P.O. # :

Samples submitted to our lab in Vancouver, BC. This report was printed on 24-OCT-91.

	SAM	PLE PREPARATION
CHEMEX	NUMBER SAMPLES	DESCRIPTION
299 288	8 8	Sample split from other certif NAA encapsulation/irradiation
* NOTE	٦.	

The 32 element ICP package is suitable for trace metals in soil and rock samples. Elements for which the nitric-aqua regia digestion is possibly incomplete are: Al, Ba, Be, Ca, Cr, Ga, K, La, Mg, Na, Sr, Ti, Tl, W.

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To: LAIRD, JAMES W.

3868 MT. SEYMOUR PARKWAY NORTH VANCOUVER, BC V7G 1C4

Comments:

CODE	NUMBER SAMPLES	DESCRIPTION	METHOD	DETECTION LIMIT	UPPE LIMF
1941	8	Ag ppm: Ultra trace package	EXT-ICP	0.05	200
1092	8	As ppm: Ultra trace package	EXT-ICP	0.5	5000
1093	8	Ba ppm: Ultra trace package	ICP	5	10000
1094	8	Bi ppm: Ultra trace package	EXT-ICP	0.2	5000
1095	8	Cd ppm: Ultra trace package	EXT-ICP	0.1	1000
1097	8	Cu ppm: Ultra trace package	EXT-ICP	0.5	5000
1099	8	Fe %: Ultra trace package	ICP	0.01	25.00
1098	8	Ga ppm: Ultra trace package	EXT-ICP	1	5000
1935	8	Hg ppm: Ultra trace package	EXT-ICP	0.1	5000
1930	8	La ppm: Ultra trace package	ICP	5	10000
1934	8	Mn ppm: Ultra trace package	ICP	5	10000
1939	8	Mo ppm: Ultra trace package	EXT-ICP	0.5	5000
1933	8	Pb ppm: Ultra trace package	EXT-ICP EXT-ICP	1	5000 1000
1089	8.	Sb ppm: Ultra trace package	ICP	0.2	1000
1942	8	Sr ppm: Ultra trace package	EXT-ICP	0.5	5000
1943	8	T1 ppm: Ultra trace package	ICP	0.5	1000
1945	8	V ppm: Ultra trace package W ppm: Ultra trace package	ICP	5	10000
1944 1946	8	Zn ppm: Ultra trace package	EXT-ICP	5 1	5000
135	8	Ce ppm: Trace rock, soil	NAA	0.5	1000
135	8	Eu ppm: Trace rock, soil	NAA	0.05	100.0
110	8	La ppm: Trace rock, soil	NAA	0.5	10000
136	8	Lu ppm: Trace rock, soil	NAA	0.05	500.0
128	8	Nd ppm: Trace rock, soil	NAA	1	100
134	8	Sm ppm: Trace rock, soil	NAA	0.05	500.0
141	8	Tb ppm: Trace rock, soil	NAA	0.05	100.00
150	8	Th ppm: Trace rock, soil	NAA	0.1	10000
131	8	U ppm: Gamma counting	NAA	0.5	10000
138	8	Yb ppm: Trace rock, soil	NAA	0.05	1000.0

A9123057

APPENDIX 3

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ROCK SAMPLE DESCRIPTIONS

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	KEY
ру	pyrite
ро	pyrrhotite
sp	sphalerite
gi	galena
ba	barite
qtz.	quartz
biot.	biotite
mssx	massive sulphides
tr.	trace

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	SAMPLE DESCRIPTIONS
	COPELAND GROUP - KING FISSURE DEPOSIT
Cliff Zone	
CZ-1	Massive suiphide; sp-gl-py-po rich basal layer underlain by
1.0m chip	mssx stockwork breccia; overlain by several metres of fine-
	grained po, py, sp, gl with breccia fragments.
CZ-2	Massive sulphide layer; po, sp, py, and gl; fine grained
2.0m chip	with breccia fragments.
CZ-CB1	Extrusive carbonatite tuff-breccia
1.0m chip	
CZ-CB2	Extrusive carbonatite tuff-breccia
1.0m chip	
CZ-FZ-1	Coarse py crystals in qtz-ankerite-calcite altered late
1.0m chip	fault zone.
CZ-GA1	Skarny contact zone between amphibolite sill and thin marble
grab	layer; large pinkish-red pyralspite garnets, qtz, biot., and
-	others.
East Zone	
EZ-1	Massive sulphide layer; gl, sp, po, py; easternmost exposure
dump grab	of southern limb mssx; old workings, possibly caved adit.
EZ-2	Massive sulphide layer; sp, gl, po, py; north limb.
talus grab	
EZ-3	2m qtz. vein in biot. schist.
grab	•
EZG-1	Massive sulphide layer; po, py, sp, gl; north limb.
0.5m chip	
EZG-2	Massive sulphide layer; po, py, sp, gl; north limb; 25m east
0.5m chip	of sple EZG-1.
EZG-3	Disseminated po in calc-silicate schist and marble.
1.0m chip	
EZG-4	Massive sulphide layer; gl, sp, po, py in old trenches near
1.0m chip	west end of south ridge on East Zone ridge.
EZG-5	Massive sulphide layer; coarse "buckshot" textured py, po,
1.0m chip	sp, gl; north limb.
Northeast Zone	
NEZ-1	Massive sulphide layer (hanging wall); ba, gl, sp, py, po;
1.5m chip	face of an old shallow adit.
NEZ-2	Massive sulphide layer (footwall); po, py, sp, gl, ba;
1.5m chip	contiguous with spie NEZ-1.
NEZ-3	Massive sulphide layer; ba, gl, sp, po, py; central ba-rich
1.0m chip	core layer within mssx.
NEZ-4	Massive sulphide layer; po, sp, py, gl, ba in an old open
1.0m chip	cut; 10m west of NEZ-1,2,3 on same horizon.
NEZ-CB1	Extrusive carbonatite tuff-breccia with minor biot. schist
3.0m chip	and marble; down-section from NEZ-1,2,3,4.
NEZ-CB2	Extrusive carbonatite tuff-breccia with minor schist and

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	COPELAND GROUP - KING FISSURE DEPOSIT
	(continued)
West Zone	
WZ-1	Massive sulphide layer; coarse dark sp, py, gl, ba;
2.0m grab	sulphide layers vary from 0.3–2.0m width with up to
	6 layers observed over 4.5m max. width. Interlayers
	include barite, marble, and siliceous schist.
WZ-2	Massive sulphide layer; coarse grey ba, gl, sp, py, po, and
2.6m chip	minor cp with wallrock breccia fragments.
Miscellaneous	
CF-1	Subcrop; massive sheared gI with minor sp and cp; coated
0.3m grab	with manganese stain, cerussite, and tr. malachite; pods(?)
	of mssx within lamprophyre dyke swarm in Camp fault.
CF-CU1	Malachite stained chalcopyrite stringers in quartzite.
talus grab	
KF-1	Coarse grained, well crystallized amphibolite sill(?);
0.75m chip	downsection and west of sple NEZ-CB2.
KF-2	Fine-grained streaky layered amphibolite sill(?); approx.
1.5m chip	100m upsection from KF–1.
KF-3	Tourmaline-bearing granitic pegmatite.
grab	
KF-4	Pyrrhotite-bearing qtz-calcite breccia in late fault zone.
grab	
KF-5	Biotite lamprophyre dyke.
grab	

	SAMPLE DESCRIPTIONS
Big Slide Zone	FRISBY GROUP
FRS-1	Gossanous calc-silicate wallrock adjacent to massive
2.0m chip	sulphide layer; prob. stratigraphic hangingwall, now structural footwall.
FRS-2	Contiguous to, and same description as FRS-1.
2.0m chip	
FRS_3	Contiguous to FRS-2; same description as FRS-1.
2.0m chip	
FRS-4	Contiguous to FRS-3; same description as FRS-1.
2.0m chip	
FRS-5	Contiguous to FRS-4; same description as FRS-1.
2.0m chip	
FRS-6,7	Massive sulphide layer; po, py, sp, gl, and marcasite.
0.5mx0.5m panel	
FRS-8	Massive sulphide layer; po, py, sp, gi, and marcasite.
0.5x1.0m panel	
FRS-9, 10	Same description as FRS-1-5; hanging wall and footwall
1.0m chip	adjacent to FRS-8.
FRS-11	Grey carbonate layer adjacent to FRS-8,9,10; stratigraphic
0.2mx2.0m chip	footwall.

	SAMPLE DESCRIPTIONS
	FRISBY GROUP (cont.)
FRS-12	Massive sulphide layer; po, py, sp, gl, intergrown with
0.3m chip	coarse green amphibolite.
FRS-13	Massive sulphide layer; po, py, sp and gl within siliceous
0.3m chip	layer.
FRS-14	Massive sulphide layer; large talus blocks broken off
talus grab	showings near FRS-15; coarse sp, po, py, gl, and
-	marcasite.
FRS-15	Massive sulphide layer; coarse sp, po, py, gl, and
1.0m chip	marcasite; isoclinally folded with thickened hinges.
FRS-16	Malachite stained cp stringers in quartzite.
grab	
FRS-17	Massive sulphide layer; po, py, sp, gl; intergrown with
talus grab	coarse green amphibolite.
FRS-FZ1	Brecciated low-angle fault zone; strongly silicified and
grab	ankeritized with minor graphite and pyrite; brecciation
	visible for several 100's metres along ridge.
FB-1	Disseminated pyrite in biotite schist.
grab	
FR-1	Malachite coated cp stringers in quartzite.
talus grab	
FR-2	Quartz-ankerite; probably late fault breccia.
grab	
South Frieby Ridg	
FRS-SL1	Fine-grained py in gossanous quartzite; site of old trench
grab	near northwest corner of a small lake.
FRS-SL2	Same as spie FRS-SL1; several 100's m west along strike,
	near larger lake; more old trenches.
North Grid	
FRN-1	Py and po in qtz-calcite veins in schist approx. 1m above
grab	carbonatite layer.
FRN-2	Py and po in siliceous fine-grained schist; 1m below
0.5m chip	carbonatite layer.
FRN-3	Same as FRN-2, approx. 2.5m west.
0.5m chip	
FRN-CB1	Extrusive carbonatite tuff-breccia; same location as
0.75m chip	FRN-1,2,3
FRN-SMB1	Scapolite-bearing white marker marble.
grab	
South Grid	
FRS-QV1	Rusty qtz. vein cross-cutting carbonatite and biot. schist.
1.0m chip	
FRS-CU1	Quartz vein stockwork in folded marble with widespread
3.0m chip	malachite-coated chalcopyrite stringers.
FRS-CB1	Extrusive carbonatite tuff-breccia with interlayered biot.
1.5m chip	schist, marble, and amphibolite.
FRS-CB2	Extrusive carbonatite tuff-breccia with dissem. phlogopite
1.5m chip	mica.

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APPENDIX 4

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SOIL GEOCHEMISTRY RESULTS

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852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6

PHONE (604) 253-3158 FAX (604) 253-1716

GEOCHEMICAL ANALYSIS CERTIFICATE

James W. Laird File # 91-4871 Page 1 3868 Mt. Seymour Parkway, North Vancouver BC V7G 1C4

bbu b					Ni ppm p		Mn ppm					Th S opmipp	r Co m ppi	1 Sb 1 ppm			Ca %	- 1997 - 1997 - 199	La ppm				Ti X	Al %	Na %	к Х	W ppm	ppm	Sn ppm	r ppm	ND ppm	Be ppm	Sc ppn
4	5	10	45	.1	3	4	359	2.40	2	10	ND	4 20	9.	2	2	39	.82	.117	21	13	.36	416	.32	6.80	1.99	1.47	2	110	1	20	13	1.0	6.9
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4	14	16	55	.1	7	6	382	1.91	2	5	ND	3 20	9	27														44	1	17	11	1.0	5.8
2	10	23	73	.2	17	8	491	2.96	7	8	ND	8 24	1 .	5 4	2	55	1.13	.216	33	34	.89	573	.34	5.99	1.79	2.44	9	48	2	22	21	1.5	8.1
1 :	22	11	53	.1	9				2	5	ND																		1	15	7	.9	8.3
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James W. Laird FILE # 91-4871

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Page 2

AMPLE#	Mo C							Mr		As				Sr ppm		Sb		۷	Ca %		La ppm		Mg %	Ba	Ti X	Al %	Na %	K S	W N	Zr	Sn	Y	Nb	Be	Sc
	ppm pp		hiii i	Jun	hhiii	ppa	ppn	ppri		ppii	ppii	ppii	ppii	phi	ppin	ppm	ppin	ppii		^	ppii	ppii		ppm		~	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	·····	(ppm	hhii	ppii	ppii	ppm p	- mqc	ppr
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67+50N 36+50E	-		27		.1	16	6		2.26		5	ND	9	199	.5	2							.50							129	4	26			7.7
67+50N 37+00E	3 1				.1				2.75	7		ND	7	266	.7	2							.71			6.50						17			7.6
67+00N 34+50E	2 2	0	23	56	.1	19	8	326	4.06	5	5	ND	11	159	1.0	2							.75								7		23		9.2
67+00N 35+00E	3 2	3	21	77	.1	27	12	706	2.90	2	5	ND	9	150	1.6	2							.79							45	6	10	14		7.3
57+00N 35+50E	2 1	5	32	123	.1	28	15	699	3.65	8	10	ND	23	189	1.1	2	2	52	1.38	.214	84	53	1.00	727	.34	7.49	1.57	2.91	4	39	9	47	29 1	1.5	13.(
57+00N 36+00E	4 1	5	10	73	.1	15			2.75		5	ND	10	229	.7					.195		31	.67	514							9	19	21		7.8
57+00N 36+50E	4 2	.3	7		i . 1 1				2.23			ND		199	.7		2	42	.76	.252	34	25	.54	399	.25	5.57	1.37	1.56	52		14	16	17	.6	7.2
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66+50N 34+50E	3 5	3	25	166	.1	94	38	1128	4.78	2	8	ND	9	258	2.4	2	2	109	2.27	.227	73	91	1.85	667	.39	6.34	.94	1.57	72	25	8	20	40	.4	11.7
66+50N 35+00E	3	8	37	57	.1	15	6	313	4.20		6		11	157	.8	2							.53							31	9	25	30 f	1.0	7.(
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6+50N 37+00E	5 1	-	28			17			2.95	-	5		9	191	.2	2							.68							63	1	21	20	.9	7.
6+00N 34+50E		_	23		.1				4.11	-	-		10	198		2							1.10							54	1		17		
56+00N 35+00E	32								3.87														1.43			7.72				41	1	21	20		102.5.5
66+00N 35+50E	2 1			57					3.38					187	.7								.63							57					
66+00N 36+00E	3 1	5	25	68	•1	15	8	382	3.01	5	5	ND	13	196	.8	2	2	59	1.01	.128	46	40	.83	675	.32	6.51	1.38	2.38	32	68	1	30	24	1.9	9.
66+00N 36+50E	5 1	7	34	100	.1	29	17	1437	4.18	7			12	201	.8	2							1.15							53		17	22	.9	11.
66+00N 37+00E	2 1		26			12			3.98		- 5		20	180	- 5	2							.60							65	2		26 °		1000
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65+50N 35+00E									4.08						1.1					.219			.89			7.39				81					· · · ·
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5+50N 36+00E	4 1	12	31		.1				5 3.72		5		12	258	.9					. 158				700						113					
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65+00N 36+50E									5.8			ND			1.2								2.16							28					
65+00N 37+00E	2 3			162	S. 2. 2				5 4.30						.2								1.35			8.10				38					11.4.4.4.4
64+50N 34+50E	4 2	29	32	115	.1	22	13	1236	5 4.43	6	5	ND	16	243	.9	2	2	83	1.03	.239	64	53	1.10	734	-40	7.61	1.47	2.7	52	51	3	22	23	1.4	12.
4+50N 35+00E	3 1	18	24	98	-1	21	15	1583	5 4.00					209									.90							46		23	23	1.9	11.
64+50N 35+50E	1 5								6.7						1.3								2.33							11			46	1.3	30.
TANDARD HFC	22 6	51	45	149	7.6	100	50	1287	4.69	43	18	7	35	60	20.0	15	20	78	.58	.128	40	110	1.10	230	.07	2.07	.06	.1	5 13	- 4	17	7	5	.2	6.

Samples beginning 'RE' are duplicate samples.

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James W. Laird FILE # 91-4871



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ACHE A	MALYTICAL		-				_																											*	CHE ANALYTICAL
	SAMPLE	Мо	Cu	Pb	Zn	Ag	NI	Co	Mn	Fe	As	U	Au	Th Sr	Cđ	Sb	BI	v	Ca	Ρ	La	Cr	Mg	Ba	Ti	Al	Na	k	C W	Zr	Sn	Y	Nb	Be	Sc
		ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm (mqq	ppm	ppm ppm	ppm	ppm p	pm ;	ppm	%	*	ppm	ppm	%	ppm	*	%	%	7	ppm ک	ppm r	ppm r	ppm	ppm p	pm	ppm
																									·										
	L64+50N 36+00E	1	80	26	135	. 1	270	68	1600	6.72	4	5	ND	8 490	1.1	2	2	194 7	. 50	. 589	78	281 3	2.34	811	1.03	7.62	. 93	1.19) 2	9	1	31	104 1	.4	20.8
	L64+50N 36+50E	2	45	39	94	. 4	56	34	871	6.66	11	5	ND	11 132	1.2	2	3	58 3	1.31	. 287	41	65	1.18	448	.29	6.56	. 42	2.32	: 2	3	1	26	13 1	.1	13.2
	L64+50N 37+00E	6	30	29	103	.2	32	16	595	3.74	8	7	ND	15 214	.2	2	3	71 1	. 60	.220	65	58	. 92	522	. 38	7.90	1.52	1.81	2	106	з	33	19 1	.4	14.2
	B.L.35+00E 64+00N	3	18	31	163	.2	32	19	1374	4.34	4	5	ND	13 181	.2	2	4	93 1	.14	.203	54	68	1.28	733	. 39	6.82	1.16	2.03	J 2	49	2	23	19 1	.1	13.4
	B.L.35+00E 63+75N	2	18	43	171	.2	31	20	1158	3.98	5	5	ND	10 217	1.4	2	5	102 1	. 62	.231	59	81	1.35	512	. 49	6.46	1.12	1.37	/ 2	47	1	19	33	.9	13.5
	8.L.35+00E 63+50N	1	24	22	155	.2	37	19	1147	3.67	2	5	ND	10 174	1.2	2	2	77 1	. 32	. 227	43	68	1.45	500	. 31	7.79	. 90	1.40) 2	47	1	15	12 1	.0	11.6
	8.L.35+00E 63+25N	1	25	16	138	.3	37	17	820	3.26	7	5	ND	11 174	.8	2	2	72 1	.66	.194	43	69	1.49	464	.28	6.60	.64	1.18	32	47	1	16	12	. 9	11.0
	B.L.35+00E 63+00N	4	31	25	152	. 3	57	23	813	3.70	4	5	ND	9 143	1.3	2	2	95 1	.03	. 153	36	63	1.05	678	.27	5.72	.74	1.43	32	29	1	14	13	.8	10.8
	B.L.35+00E 62+50N	5	12	24	109	.1	28	16	894	4.11	2	6	ND	14 163	1.0	2	4	87 1	. 00	.178	52	62	1.08	789	. 35	6.07	1.09	1.84	¥ 2	39	1	25	20 1	. 1	12.7
	B.L.35+00E 62+00N	3	17	30	107	.2	24	12	697	3.96	4	5	ND	11 206	.7	2	5	98 1	. 30	.161	50	68	. 95	691	.43	6.17	1.14	1.55	2 ذ	56	1	21	27 1	.1	12.4
	B.L.35+00E 61+75N	2	6	25	65	.1	9	5	423	2.53	2	9	ND	8 230	.2	2	3	68 1	1.13	.095	41	36	. 56	712	. 39	5.99	1.75	2.4	j 2	71	2	25	22 1	.6	9.6
	B.L.35+00E 61+50N	1	35	26	168	.1	53	19	409	4.35	6	5	ND	13 172	. 9	7	2	102 1	1.23	. 172	51	96	1.84	714	. 37	9.40	.76	1.68	32	39	1	19	15 1	.2	14.0
	RE B.L.35+00E 60+00N	3	21	33	94	.1	26	11	883	4.83	7	10	ND	26 158	.2	2	2	80	.76	.147	111	68	1.00	686	.41	7.24	1.12	2.42	2 2	57	1	24	20 1	. 1	13.8
	B.L.35+00E 61+00N	3	11	20	82	.1	17	9	611	3.74	6	5	ND	10 189	.2	2	2	70	.93	. 198	44	49	. 88	549	. 38	6.79	1.46	2.03	32	85	2	20	18 1	. 1	11.5
	B.L.35+00E 60+75N	1	12	25	69	.1	10	8	328	3.21	7	9	ND	9 505	.7	2	3	76 1	1.69	.282	56	60	.86	871	. 45	7.06	1.67	1.9	2 ز	111	2	25	23 1	. 3	11.8
	8.L.35+00E 60+50N	4	15	22	54	.1	11	4	346	3.03	2	5	ND	7 218	.9	2	3	52	.88	.103	29	25	. 38	511	. 35	6.48	1.98	1.77	12	130	3	16	16	. 9	7.8
	B.L.35+00E 60+25N	4	17	21	69	.1	14	5	348	3.33	4	21	ND	11 273	.2	2	4	64 1	1.18	.184	42	32	.62	549	. 40	7.86	2.18	1.92	2 2	211	1	24	19 1	.4	10.9
	B.L.35+00E 60+00N	4	24	27	96	.1	30	11	925	5.07	6	5	ND	21 178	1.4	2	2	86	.76	.158	84	69	1.04	805	.43	7.49	1.50	3.24	1 2	50	2	25	23 1	.0	14.3
	B.L.35+00E 59+75N	4	17	11	65	. 3	11	5	324	2.99	2	7	ND	8 249	.2	2	2	52 1	1.12	. 164	-				. 34					163	1	16	14	. 9	8.9
	B.L.35+00E 59+50N	4	19	21	79	. 3	15	8	649	3.57	7	5	ND	11 227	.7	2	2	64 1	1.13	. 198	40	39	.73	520	. 37	6.94	1.82	1.89) 2	112	2	18	17	. 9	10.1
	B.L.35+00E 59+25N	-	-				6	-		2.93	-	-		11 189	-	-				.123					.41					97			26 1		
	B.L.35+00E 59+00N	-	10		78			8		3.46				12 177	.2	2	_			. 119					. 35						3		22 1		
	B.L.35+00E 58+50N	-	23				24			4.01				14 156	.5					. 151					. 40						3		21 1		
	B.L.35+00E 58+00N		6		69					3.25				13 208	.2	2		-	.95	.145					. 39					71			22 1		9.8
	B.L.35+00E 57+50N	1	10	29	67	.1	13	4	293	2.70	2	5	ND	10 194	.8	2	3	50	.87	.113	43	30	. 44	719	. 35	5.82	1.81	2.5	4 2	53	4	26	24 1	.2	8.5
						_	_	-			•	-		e 10e		-	•				••		••		~~						~		•	•	7.3
	8.L.35+00E 57+00N			-	47	-				2.82			ND	6 186					.75			-			.28					176			-		
	B.L.35+00E 56+50N	-			69		16	-	-	4.06			ND	9 223	.2		3		.85	. 152					. 39										9.8 9.2
	8.L.35+00E 56+00N									3.42				7 200					. 98	.171					. 33										12.7
	B.L.35+00E 55+50N		22		98		21			4.35				12 171				_	.86	. 197					.45										11.3
	B.L.35+00E 55+00N	2	21	19	91	. 3	21	10	/42	4.18	2	2	NU	11 181	. 2	2	2	/3	. 80	.110	43	50	.//	270	. 40	1.76	1.00	1.0	J 2	00	2	10	10	• •	11.3
	B.L.35+00E 54+50N		24	20	75	,	12	£	640	3 40	2	5	NO	12 201	.6	,	2	76	۵Q	. 127	61	40	65	680	. 46	6.92	1.07	2 2	<u>ہ</u>	60	7	16	26	. 6	10.0
	B.L.35+00E 54+00N					-	11			3.53				8 228						. 196					. 37					196		14			9.3
	B.L.35+002 54+00N B.L.35+00E 53+50N											-		20 457				-		1.281												67 1			
	B.L.35+00E 53+00N		26) 98					3.67				8 204													1.83			113		13			12.1
	B.L.35+00E 53+00N		18		, 90 ; 68	-	19					-		11 159						.183					.35					113					10.8
	0121337VV6 3273UN	-	10	10	,	• •	13	0	504	9.01	-	J			• •	-	-				04	45									-	- •			
	B.L.35+00E 52+00N	2	27	11	64		16	4	338	2.98	7	5	ND	6 272	1.2	2	2	50	.94	.249	19	17	. 39	472	. 30	8.55	2.85	1.6	7 2	206	1	13	11	. 4	8.5
	B.L.35+00E 51+50N			-			24							13 214						. 182					. 42										11.9
	STANDARD HFC													36 61																					6.0
																		<u> </u>																	

Samples beginning 'RE' are duplicate samples.



James W. Laird FILE # 91-4871



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ACHE ANALYTICAL																																	ACHI	ANAL	TICAL
SAMPLE#	Mo (ppm pp														Cd ppm				Ca %		La ppm		Mg %	Ba ppm	Ti %						Sn ppm j				Sc ppm
B.L.35+00E 51+00N	2 1	13	22	78	.1	13	9	1018	3.01	4	5	ND	9	172	1.4	2	2	49	.57	. 160	37	33	.61	533	.27	6.36	1.75	1.80	4	117	1	12	14	8	8.2
B.L.35+00E 50+50N	3 1								3.23			ND			1.7															54		9			8.5
B.L.35+00E 50+00N					1				3.99			ND			2.0	2								715											11.6
T.L.36+00E 64+00N	1 1				.2				3.76			ND			1.8	2	2							775						53					10.2
T.L.36+00E 63+75N	2 1				.1				2.75	-	-	ND		304	.9	2	2							1006							1				
T.L.36+00E 63+50N	2 1	10	14	65	.1	12	4	349	2.85	2	5	ND	9	240	1.4	2	2	54	.89	.070	47	29	.52	698	.34	6.61	1.96	2.47	2	75	1	19	15	1.4	7.9
T.L.36+00E 63+25N	2 1	14	29	55	.1	5	3	320	1.57	2	5	ND	8	354	9	2	2	52	.94	.018	46	24	.47	1259	.40	6.69	2.61	3.72	2	80	. 1	16	20	1.8	7.4
T.L.36+00E 63+00N	3 1	13	19	52	.1	11	4	314	2.60	4	5	ND	8	214	.9	2	2	45	.83	.132	35	24	.45	662	.32	6.68	2.08	2.14	2	96	2	18	16	1.4	7.7
T.L.36+00E 62+75N	2 1	11	19	64	. 1.	9	6	525	3.28	2	5	ND	12	193	1.5	2	2	58	.75	.162	49	36	.62	703	.34	6.09	1.81	2.96	3	67	1	18	19	1.3	7.8
T.L.36+00E 62+50N	3 1	10	13		.2					5		ND			1.3	2								609						51				1.2	6.4
T.L.36+00E 62+25N	3 1	18	16	82	.2	12	7	672	3.07	2		ND		217	1.0									801						54	1	19	17	1.4	8.2
T.L.36+00E 62+00N	2 1	15	21	74	.1	17	7	511	2.99		5				1.4	2								628						74	1	21	22	1.3	8.3
T.L.36+00E 61+75N	3	4	37	93	.2	19	8	800	3.33					204	.9	2	2							782						60	1	18	17	1.3	9.3
T.L.36+00E 61+50N	2 2	23	33 ⁻	101					3.45			ND		208	.9	2	2							702						64	1	22	16	1.5	9.5
T.L.36+00E 61+25N	2	6	23	68	.1	16	6	471	2.94	4	5	ND	9	237	.4	2	2	57	.95	. 135	40	39	.60	804	.35	6.79	2.03	3.07	3	100	2	19	18	1.6	8.5
T.L.36+00E 61+00N	4	18	16		.1				3.46			ND		291		2								810						124	1	18	20	1.3	9.7
T.L.36+00E 60+75N	3	13	26	72	.1	15	8	466	3.43	i 4	5	ND		184	.5	2								840						65	1	29	23	1.9	10.0
T.L.36+00E 60+50N	1 1	10	16	85	.1	26			4.37			ND			1.5	2								540											12.2
T.L.36+00E 60+25N	4 2	21	23	90	.1	15	- 7	385	3.77	7 4	5	ND				2								755						77	. 1	18	19	1.3	10.6
T.L.36+00E 60+00N	5	14	23	66	.3	13	7	384	3.44	3	5	ND	10	216	.6	2	2	56	.88	.173	41	36	.64	677	.33	6.65	1.73	2.54	. 4	75	; 1	19	15	1.2	9.1
T.L.36+00E 59+75N	2	7	28		.1				3.41			ND			.6									802						37	2	20	21	1.9	10.0
T.L.36+00E 59+50N	4	8	27	68	.1	15	8	452	3.83	5 _ 4	6	ND	13	199	.4	2	2	65	.96	.124	63	54	.69	819	.36	6.38	1.39	2.96	5 2	45	2	23	23	1.9	9.7
T.L.36+00E 59+25N	2	16	33	87	.1	12	10	1231	3.38	4	5	ND	12	227	.2	2	2	59	1.14	. 159	62	42	.79	812	.33	6.55	1.59	2.78	2	43	1	24	21	1.8	9.3
T.L.36+00E 59+00N	4	10	25	102	.1	17	10	1176	3.56	5 3	5	ND	12	235	.7	2	2	66	1.02	.222	59	47	.86	827	.38	6.94	1.67	3.10	2	60	ା 1	21	21	1.7	10.1
T.L.37+00E 64+00N	6	20	21	154	.1	38	19	1348	3.57	2	8	ND	9	258	.5	2	2	80	1.61	.326	72	64	1.16	566	.36	7.84	1.50	1.59	2	64	1	29	16	1.8	10.9
T.L.37+00E 63+75N	5 3									i . 6		ND		272	.5	2								860							: 1				10.7
T.L.37+00E 63+50N	6								3.16			ND		204	.9									666							1				8.0
T.L.37+00E 63+25N	5								3.51			ND	11	178		2								666					_	. 80	4	24	20	1.8	8.7
T.L.37+00E 63+00N	3								2.86			ND		171	.2	2								622						72		18	20	1.5	6.5
RE T.L.37+00E 64+00N	6	20	30	152	-1.	31	20	1341	3.53	5 4	6	ND	8	234	.2	4	2	77	1.63	.312	66	64	1.10	501	.33	7.37	1.21	1.42	2 2	66	1	30	17	2.0	10_5
T.L.37+00E 62+75N	5	19	20	79	.1	17	17	631	2.89			ND												588						50					8.4
T.L.37+00E 62+50N	4	25	21	66	ો1	19	8	366	2.98	3 🔆 2	5	ND		186	8	2								558						69					7.9
T.L.37+00E 62+25N	4	14	24	123	1	12			4.00			ND			1.5	2								645											10.8
T.L.37+00E 62+00N	2	14	32	68	.1				2.12			ND		210										977						25					10.0
T.L.37+00E 61+75N	4	20	30	80	.1	15	11	532	3.14	5	5	ND	14	179	.3	2	2	54	.85	. 163	60	45	.76	681	.30	6.87	1.61	2.92	2 3	69	2	26	17	2.0	8.7
T.L.37+00E 61+50N	3	14	21	54	.1	12	6	320	3.51			ND		176		2								590						103	2				7.8
T.L.37+00E 61+25N									2.40			ND		199	3.3	2	2	48	.97	.154	36	41	.73	599	.27	5.33	1.68	3 2.51							7.9
STANDARD HFC	22	63	47	144	7.3	98	51	1302	4.78	3 43	17	6	- 38	59	20.0	15	22	80	.58	.128	38	110	1.10	219	.07	2.02	.06	5.17	7 11	- 4	17	7	- 4	.2	6.0

Samples beginning 'RE' are duplicate samples.

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James W. Laird FILE # 91-4871



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ACHE ANALYTICAL	_																																					ACHE AN	NALYT	ICAL
SAMPLE#							•			Mr ppr				-				Cd ppm							a Cr appm								W ppm							
T.L.37+00E 61+00N		•	16	31	7.	3	.1	10	6	507	2.85	5	2	5	ND	6	225	.2	2	2	49	1.09	.16	> 30) 33	.5	2 643	3.3	75.	97 1	.75	2.39	2	111	3	32	19	2.0) 11	2
T.L.37+00E 60+75N		2	25	18	69	9.	.1	24	8	671	3.87	7 .	4	5	ND	8	217															2.63	2	118	2	38	35	2.2	2 15	2
T.L.37+00E 60+50N		2	22	24	75	9	.1	18	8	581	3.02	2 . 4	4:	5	ND	12	244	.5	2	2	54	1.22	.210) 61	49	.8	7 692	2 .3	5 6.	45 1	.69	2.39			2					
T.L.37+00E 60+25N	- I	1	26	30	7	1 8	.1	18	7	446	3.20) ්	6	5	ND	16	142	.2	5	2	49	.91	.11	69	> 52	.7	4 67	2 .2	95.	96 1	.12	3.05			2					
T.L.37+00E 60+00N	1	1	12	20	5	1 -	.1	9	3	454	2.5	1	2	5	ND	6	255	.6	2	2	49	.85	.08) 33	33	.4	0 79	0.3	5 5.	81 1	.88	2.85			1					
T.L.37+00E 59+75N		2	18	32	58	8	.1	8	3	432	1.9	5	5	5	ND	6	190	.4	2	2	45	.91	.07	5 38	3 32	.3	7 72	0.3	5.	16 1	.34	2.72	2	44	3	28	20	2.1	1 8	3.3
T.L.37+00E 59+50N		1	12	25	7	9	.1	11	7	1105	2.9	ו כ	6	5	ND	10	203	.7	2	2	59	.95	.13	t 5é	5 40	.5	9 58	1.3	55.	48 1	.41	1.92	2	53	1	23	16	1.4	10	J.6
T.L.37+00E 59+25N		2	10	26	5	5	.1	14	5	552	3.0	5	2	5	ND	9	163	.5	2	2	54	.79	.10	2 42	2 43	.5	2 65	4.3	2 5.	60 1	.32	2.43	2	39	2	26	17	1.8	3 10).2
T.L.37+00E 59+00N		4	20	26	7	1	.2	10	8	1528	3.1	5 🔅 :	3	5	ND	7	191	.5	2	2	54	.89	.18	5 34	42	.6	6 50	7.3	4 5.	72 1	.41	1.77	2	72	2	19	12	1.3	5 10	1.4
T.L.38+00E 64+00N		3	16	26	6	3	.3	9	6	460	2.7	4	2:	5	ND	6	206	.4	2	2	46	1.02	.04	5 30) 31	.4	6 58	3.3	45.	74 1	.62	2.01	2	81	2	27	15	1.8	88	3.9
T.L.38+00E 63+75N		3	21	17	6	3	.1	16	7	510	2.6	5	6	5	ND	6	166	.2	2	2	39	.86	. 16	5 27	7 33	.5	8 35	5.2	55.	49 1	.13	1.16	2	94	1	20	9	1.3	38	3.4
T.L.38+00E 63+50N		3	13	27	6	2	.1	17	8	553	2.5	7	6	5	ND	5	220	.2	2	2	42	1.00	.20	4 49	37	.7	3 49	3 .2	65.	09 1	.20	1.51	2	75	1	22	12	1.5	5 9	2.0
T.L.38+00E 63+00N		2	22	15	4	6	.1	12	5	380	2.5	5	6	5	ND	6	201	.2	2	2	49	.85	.12	1 29	7 35	.4	0 50	4.3	56.	48 1	1.65	1.72	2	126	. 2	17	9	1.0	0 9	2.8
T.L.38+00E 62+75N		3	19	29	5	4	.1	9	6	518	1.9	4	4	5	ND	6	266	.7	3	3	54	1.41	.06	0 37	7 35	.6	4 67	7.3	8 6.	00 2	2.01	2.12	2	93	3	21	13	1.5	5 5	2.4
T.L.38+00E 62+50N		3	30	14	5	1	.2	14	7	434	3.3	2	7	5	ND	12	160	.2	2	2	37	.69	. 16	2 37	7 37	.4	5 38	3.2	6 5.	71 1	1.17	1.23	2	106	2	26	10	1.0	0 5	7.5
T.L.38+00E 62+25N		2	31	12	4	5	.1	12	5	322	1.9	0	5	5	ND	10	150	.2	2	2	26	.73	. 15	3 23	3 26	.4	0 25	0.1	8 5.	65 1	1.10	.75	2	89	1	16	6		B 7	7.7
T.L.38+00E 62+00N											3.1								2	2	38	.89	.16	3 31	1 26	.5	3 38	4.3	2:5.	86 1	1.55	1.03	2	141	ė 1	24	10	.8	8 8	3.7
T.L.38+00E 61+75N		2	16	12	4	2	.1	6	- 4	425	1.7	3	2	5	ND	3	265	.2	2	2	27	1.16	5.10	1 10	5 11	.4	3 43	5.2	7 7.	11 2	2.25	1.24	2	170	§ 1	16	3	.8	8 8	3.3
RE T.L.38+00E 62+50N		2	35	16	4	5	.4	12	7	462	3.6	0	4	6	ND	14	168	.5	2	2	40	.73	.16	5 35	5 36	.4	9 41	1 .2	8 6.	16 1	1.26	1.11	2	1.10	6 1	25	10	.7	7 🕄 9	2.8
T.L.38+00E 61+50N	1	4	16	13	3	1	.2	9	4	288	3.2	8	4	5	ND	5	143	.2	2	2	32	.66	5.12	9 18	8 28	.3	6 25	0.2	3 6.	11	1.16	.75	2	101	1	12	6	5	5 7	7.8
T.L.38+00E 61+25N		4	11	23	5	9	.1	9	6	503	3.3	4	5	5	ND	6	245	.2	2	2	58	1.16	5 .07	0 32	2 35	.6	7 58	3.3	96.	26 ⁻	1.90	1.74	2	86	2	19	13		9 5	2.5
T.L.38+00E 61+00N	1	2	19	19	6	3 -	.2	17	8	482	3.0	5	2	5	ND	8	179	.2	2	2	- 44	. 95	5.13	1 39	37	'.7	1 43	9.2	7 5.	74 '	1.23	1.36	2	73	i 1	21	11	.5	9 8	3.9
T.L.38+00E 60+75N		2	23	18	7	1	.1	21	16	665	3.5	0	4	5	ND	12	201															1.69		66	1	29	14	1.3	3 11	6.1
T.L.38+00E 60+50N											2.0			5	ND	3	227	.2	2	2	42	1.21	.34	2: 18	8 29	.8	1 28	3 .2	8 5.	55	1.40	.92	2	60	i 2	10	9		6 8	8.5
T.L.38+00E 60+25N											2.9			5	ND	11	201	.4	2	2	51	.98	3 .26	5 37	7 42	.9	6 43	2.2	8 6.	53	1.35	1.16	2	75	2	20	10		7 10).4
T.L.38+00E 60+00N		3	16	23	6	7	.1	21	9	508	4.3	0 :	2	5	ND	12	180	.2	2	2	77	.92	2.13	9: 49	9 61	1.1	0 68	3.4	36.	93	1.17	2.09	2	66	2	24	16	5 _8	8 12	2.9
T.L.38+00E 59+75N											5.1						215															1.72		69	3	19	15		3 10	3.9
T.L.38+00E 59+50N											4.2																					2.56			1					
T.L.38+00E 59+25N		_		15	-						4.6		-				157															1.54			<u>i</u> 1					
T.L.38+00E 59+00N		-			-	-			_		5 4.3		-	-			187															2.03		60	F 2	12	14	4	4 1	1.4
STANDARD HFC	2	1	59	45	14	06	5.7	97	47	1240) 4.4	43	7	16	7	38	55	21.0	16	19	82	.58	3.12	0 31	8 110	1.2	5 23	0.0	6 1.	93	.07	.16	11	5	15	6	, 4		2	6.2

Samples beginning 'RE' are duplicate samples.

APPENDIX 5

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EAST ZONE VLF-EM DATA

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11+50N 11+25N 11+00N	L23+00E -6 -3	1.23+60E -10	L24+00E	L25+00E							
11+ 25N	-6	and a second			L26+00E	L27+00E	L28+00E	L29+00E	L30+00E L	31+00E	
	-3				-23		<u> </u>	ter for Ben Hard and an Andrea for State of State of State	18	8	11+601
	-3				-22				18	8	
11.00N		-6		-7	-12				13	10	11+25
11JON				-3	-9				18	9	
I I TAAII	0	-2		-1	-3				18	9	11+00
				0	3				18	15	
10+75N	1	0		3	2				20	13	10+75
			1	5	5				19	12	
10+50N	2	0	0	4	6	-27			20	10	10+60
			2	5	2	28			18	5	
10+25N	7	7	2	10	2	-19			19	5	10+25
			-2	8	0	-22			17	0	
10+00N	-3	-4	0	7	-4	-20	-4	4	15	0	10+00
			-3	8	. –7		0	2	16	-3	
9+75N	-10	-15	-5	. 12	-5	4	5	. –3	9	2	9+ 75
			-11	2	-4		7	-1	0	11	
9+50N	-12		-10	-9	-6	3	9	0	7	8	9+50
			-9	-14	-4		0	-5	-2	1	
9+25N	-9		-11	-13	1	6	-3	-3	0	5	9+25
			-12	-9	4		-7	-6	-5	0	
9+00N	-6		-12	-5	6	-6	8	0	1	1	8+001
			-13		10				0	4	
8+75N	-5		-10	-4	10	1	-2	0	-4	-8	8+75
			-9	_	10	_			-15	-20	
8+50N			8	-2	9	-5	-3	-3	-14	-28	8+50
			-6	_	12			-	-12	-20	
8+25N			-8	0	13	1	3	-3	-15	-15	8+25
			-8	_			•		-10	-13	
8+00N			-5	5	9	4	2	-3	-2	-11	8+00
ويتعلقوا الهورات			-4	-	8		^	~	-6	-5 10	7+75
7+75N			-2	8	8	10	3	-2	-1	-10 -11	7+70
			1	-	9		0	-1	2 6	-11	7+50
7+60N			0	7	13 8		0	-1	ь З	-14	7400
		•	3		8		^	1	3 5	-30 -24	7+25
7+25N			-3	10			3	1	5	-24	7 120
7+00N			3 -2	13			4	2	40		7+00

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			King F	issure D	Quadratur		one VLF	Survey			
	L23+00E	L23+50E	L24+00E	L25+00E	128+00E	L27+00E	L28+00E	L29+00E	L30+00E	L31+00E	
11+50N	24	23			19				26	23	11+60
					16				28	23	
11+25N	20	24		17	15				31	26	11+25
				16	16				30	27	
11+00N	15	14		18	14				30	30	11+00
				16	18				32	30	
10+75N	15	15		16	17				33	32	10+75
			26	19	18				36	30	
10+50N	19	18	26	15	19	19			37	34	10+60
			17	12	18	21			35	33	
10+25N	20	35	15	10	19	24			36	35	10+2
			18	6	15	25			38	37	
10+00N	17	16	14	4	12	28	30	36	36	40	10+00
			13	9	12		32	33	38	40	
9+75N	11	38	8	14	12	32	34	35	40	39	8+75
			8	-4	14		35	33	40	39	
9+50N	10		7	-10	16	38	37	31	37	38	9+50
			7	-9	16		40	29	36	37	
9+25N	16		7	-5	17	25	28	30	35	39	9+25
			6	1	16		22	26	36	40	
9+00N	18		6	7	20	14	18	24	35	40	9+00
			6		24				40	39	
8+75N	20		9	12	21	15	16	27	39	40	8+75
			9		23				39	38	
8+50N			11	12	28	18	20	28	38	39	8+50
			12		26				38	39	
8+25N			13	17	29	18	22	20	38	40	8+25
			14						40	39	l selet da La sel
8+00N			14	20	28	18	18	18	32	39	8+00
			16		26				31	38	
7+75N			18	22	28	22	23	23	35	40	7+75
			18		31				38	40	
7+50N			18	23	30		22	26	40	40	7+50
			16		24				40	40	
7+25N			19	22			23	32	38	38	7+25
			20								
7+00N			21	26			28	35	40		7+00

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APPENDIX 6

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EAST ZONE GEOPHYSICAL INTERPRETATION

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November 1, 1991

TO: First Standard Mining

ATTENTION: Mr. Tiro Clarke 301 - 357 East 2nd St., North Vancouver, B.C., V7L 1C6

Dear Sirs: RE: KING FISSURE DEPOSIT GEOPHYSICAL SURVEYS

This letter will confirm that I was requested by Mr. Clarke to review geophysical data from the King Fissure deposit, East Zone. The following data was supplied:

- 1) VLF-EM in phase and quadrature % data, Ronka EM16;
- 2) Total Field Magnetic data, Scintrex MP2 magnetometer;
- 3) Geology, preliminary plan map at scale 1:2,500

INTERPRETATION

VLF - EM Survey

The massive sulfide bands noted on the geology map are striking approximately perpendicular to the line drawn between the survey location and the VLF-EM transmitter in Seattle, Washington. This orientation results in minimum electromagnetic coupling which decreases the potential amplitude of any crossovers thereby negating much of the usefulness of the survey technique. More useful data would have been obtained using a transmitter location nearer on strike with the sulfide bands - for example Cutler, Maine.

In order to further interpret the VLF data, a contoured plan map of the Fraser filtered in phase % data was completed, Figure 1. With this technique, conductors are indicated by positive values, and topographic and geological noise are generally suppressed.

The data contours reasonably well indicating an arcuate conductor located at 11+00N on the north end of L26E, bending around to the baseline on L27E and then westerly toward 9+00N on L25E. This feature does not generally correlate with the observed massive sulfide bands nor with any noted distinct geological unit, and is not closed off by survey data to the north and north east. The generally high quadrature values relative to the in phase values are typical of poor conductors in a highly resistive host rock. It may be related to an area of increased conductivity or perhaps disseminated sulfide mineralization.

The most promising fraser filter anomaly occurs on L27E at between stations 9+62.5N to 10+12.5N, and correlates well with the mapped massive sulfide bands at L27E 10+00N and locally anomalous magnetic responses.

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MAGNETIC Survey

The contoured plan of relative magnetic values is shown as Figure 2. The values were calculated by subtracting 57,955 gammas (the average total field magnetic value for all the observed stations) from each station. Profiles of relative magnetic value and fraser filtered VLF-EM were plotted for L25 to 27E and L30E, figures 3a to 3d.

Given the generally wide line spacing (100 metre) for the data and a narrow target (sulfide bands are generally less than 10 metres wide), the data contouring was biased in the direction of the mapped strike of the massive sulfide bands. Further surveying with a 50 metre line spacing will be required to confirm the validity of the interpretation.

In general the data exhibits a fairly high geological noise background with readings varying 20 to 50 gammas between stations - fairly typical for metamorphic rocks of this type and indicating variable magnetite content.

With the exception of the anomalous high values (greater than 200 gammas) concentrated in a tight "fold structure" in the central to north western part of the grid and likely related to sulfide (pyrrhotite) mineralization - the magnetic data does not appear to differentiate between the geologic units mapped as calc silicate gneiss or quartz biotite schist.

The observed "fold structure" appears consistent with the geological observations of Mr. Clarke, that is sulfide bands (variable pyrrhotite/sphalerite/galena) occupying two limbs of a tight, recumbent fold with an axial plane oriented approximately at 130 to 147 degrees. The sulfide band occupying the southern limb of the fold appears to dip to the west south west.

Probable pyrrhotite concentrations in the bands occur on the north end of L25E (>1000 gammas), at BL 10+00N on L27E (884 gammas) and at 9+87.5 to 10+00N on L26E (675, 1064 gammas). Each of these zones is near surface with widths of 10 to 15 metres based on the 1/2 width of the magnetic profile. On L27E and L30E, and probably on L25E, these magnetic highs correlate with a positive fraser filter anomaly, suggesting the pyrrhotite is massive over reasonably continuous distances.

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CONCLUSIONS AND RECOMMENDATIONS

The survey has successfully demonstrated the usefulness of both magnetic and VLF electromagnetic techniques to outline the massive sulfide bands.

Further magnetic surveying on a 50 metre grid line spacing is recommended to fully outline and define the trace of the pyrrhotite mineralization. The grid should be expanded particularly to the north and north west. As this data is being acquired, further VLF - EM data, using Cutler, Maine should be obtained. My preference would be to use an instrument which measures field strength rather than dip angle, such as Scintrex IGS VLF (this instrument can measure and store both magnetic and VLF data).

The strongest fraser filter anomaly occurs on L27E near the BL, in the "nose" area of the interpreted fold. This strong response is probably more related to the strike direction of the sulfides rotating around to an increased coupling orientation, rather than to increasing conductivity in the sulfides.

On completion of the magnetic survey, a detailed low frequency horizontal loop electromagnetic survey, such as GENIE SE-88 should be carried out over noted magnetic highs to determine the precise location, width, conductivity and dip of the sources. The GENIE is particularly useful in areas of rough topography due to its general insensitivity to topography and station spacing accuracy.

Respectfully submitted,

Douglas R. MacQuarrie Geophysicist

enclosures

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CERTIFICATE

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I, Douglas R. MacQuarrie, certify that:

- 1. I am a Consulting Geophysicist with offices at 704 -850 West Hastings Street, Vancouver, B.C., V6C 1E1.
- I am a graduate of the University of British Columbia with an Honours degree in Geology and Geophysics (B.Sc., 1975).
- 3. I have been practising my profession since 1975 and have been active in the mining industry since 1971.
- 4. I am an active member of the British Columbia Geophysical Society.
- 5. This report is based on fieldwork presented to me by Mr. Tiro Clarke. I have not visited the property.

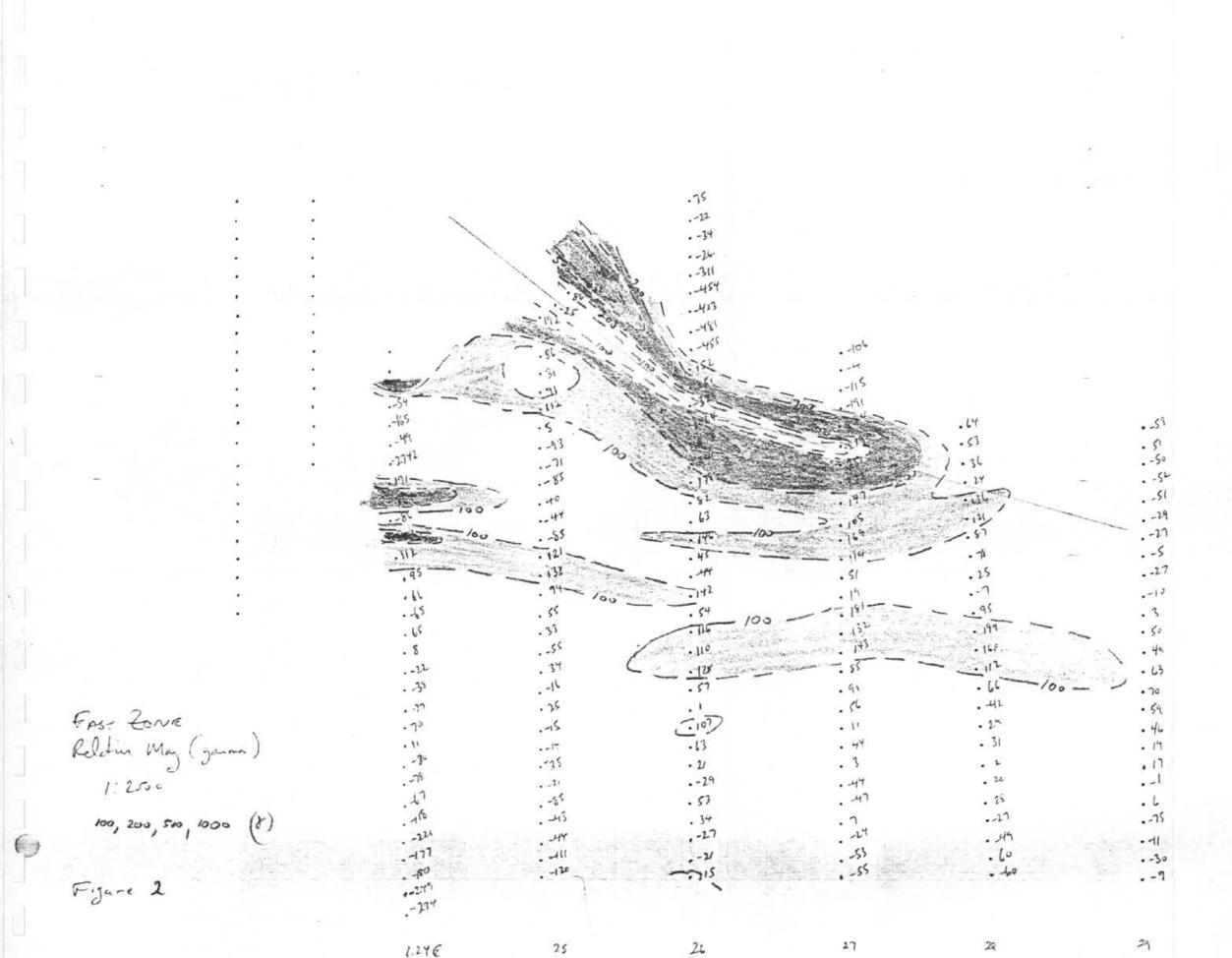
November 1, 1991 Vancouver, B.C.

Douglas R. MacQuarrie B.Sc.

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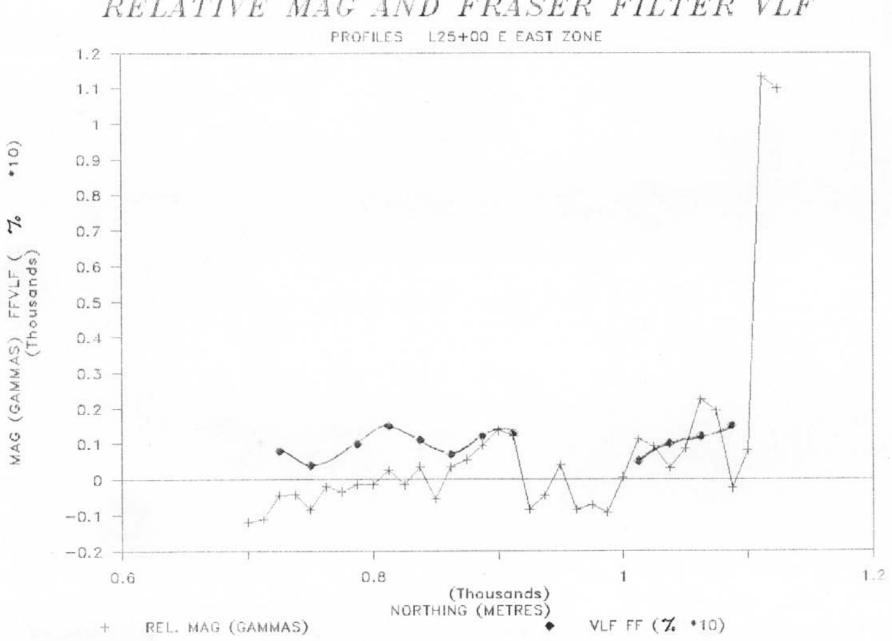
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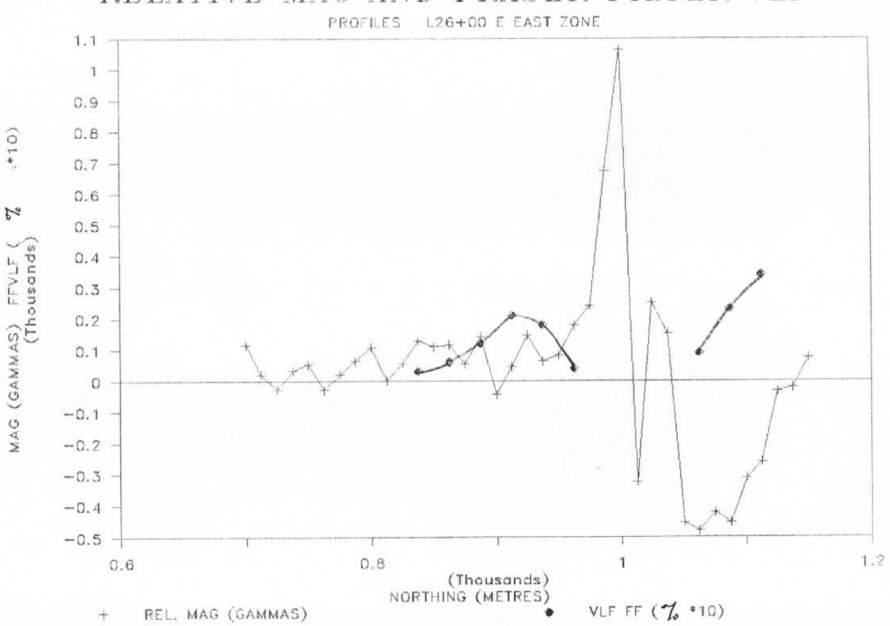
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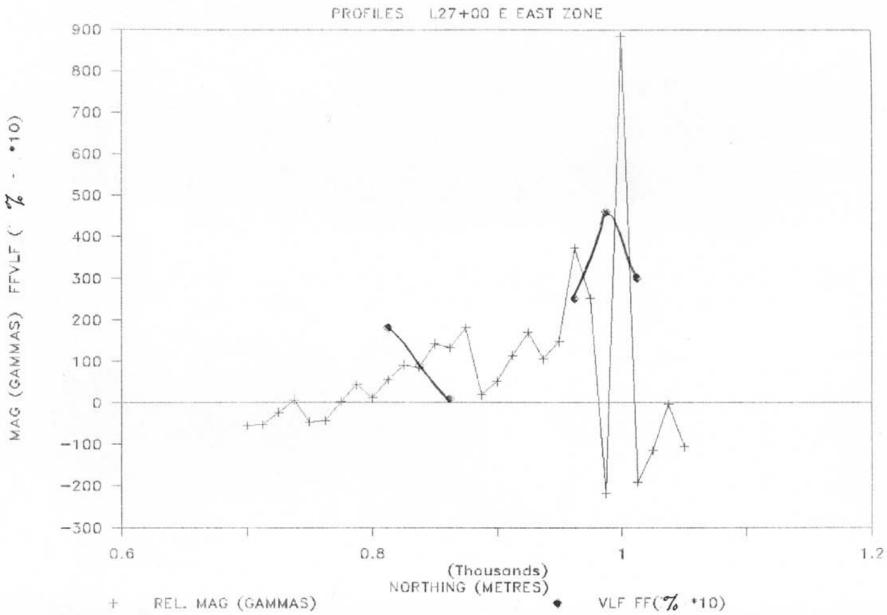
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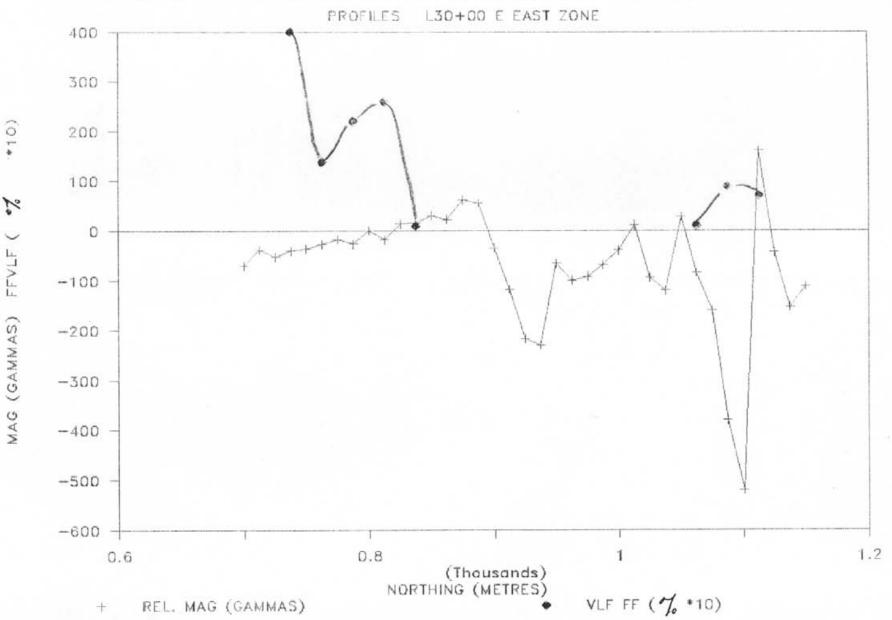


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APPENDIX 7

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FRISBY RIDGE SOUTH GRID VLF-EM DATA

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	FRISBY RIDGE SOUTH GRID VLF-EM DATA						
			(in phase		-		
							
	BL35+00E	TL36+00E	TL37+00E	TL38+00E	TL39+00E	TL40+00E	
64+00N	40	0	-25	-14	-14	-25	64+00N
	42	-3	-26	-14	-16	-26	
63+50N	39	-4	-27	-14	-18	-27	63+50N
فيتنابع ومعاورته	35	-5	-28	-14	-17	-27	
63+00N	29	-7	-28	-14	-20	-27	63+00N
	25	-8	-28	-15	-21	-25	
62+50N	24	_10	-27	–15	-21	-25	62+50N
	24	-12	-22	–15	-22	-25	
62+00N	26	-12	-16	-15	-24	-24	62+00N
	26	-12	-17	-18	-23	-22	
61+50N	19	-11	-18	18	-24	-21	61+50N
	12	-11	-17	-19	-24	-21	
61+00N	11	-10	-22	-21	-23	-21	61+00N
	7	-15	-21	-20	-21	-21	
60+50N	7	-15	-18	-19	-20	-20	60+50N
	7	-15	-19	-19	-21	-19	
60+00N	6	-17	-20	-21	-20	-18	60+00N
	5	-19	-22	-23	-18	-13	
59+50N	5	-21	-23	-24	-16	-7	59+50N
	6	-23	-24	-22	-11	2	
59+00N	7	-23	-24	-19	-6	-4	59+00N

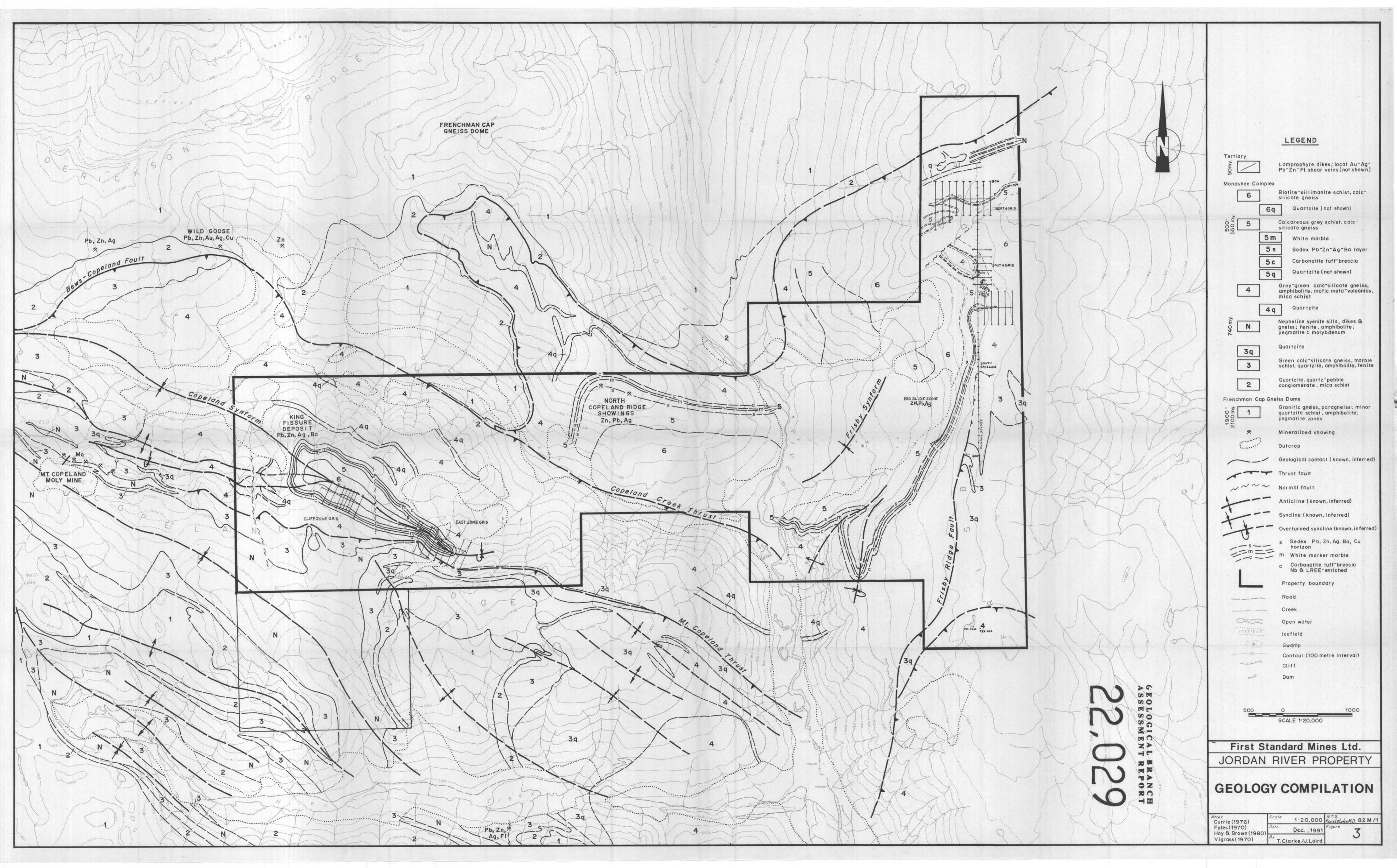
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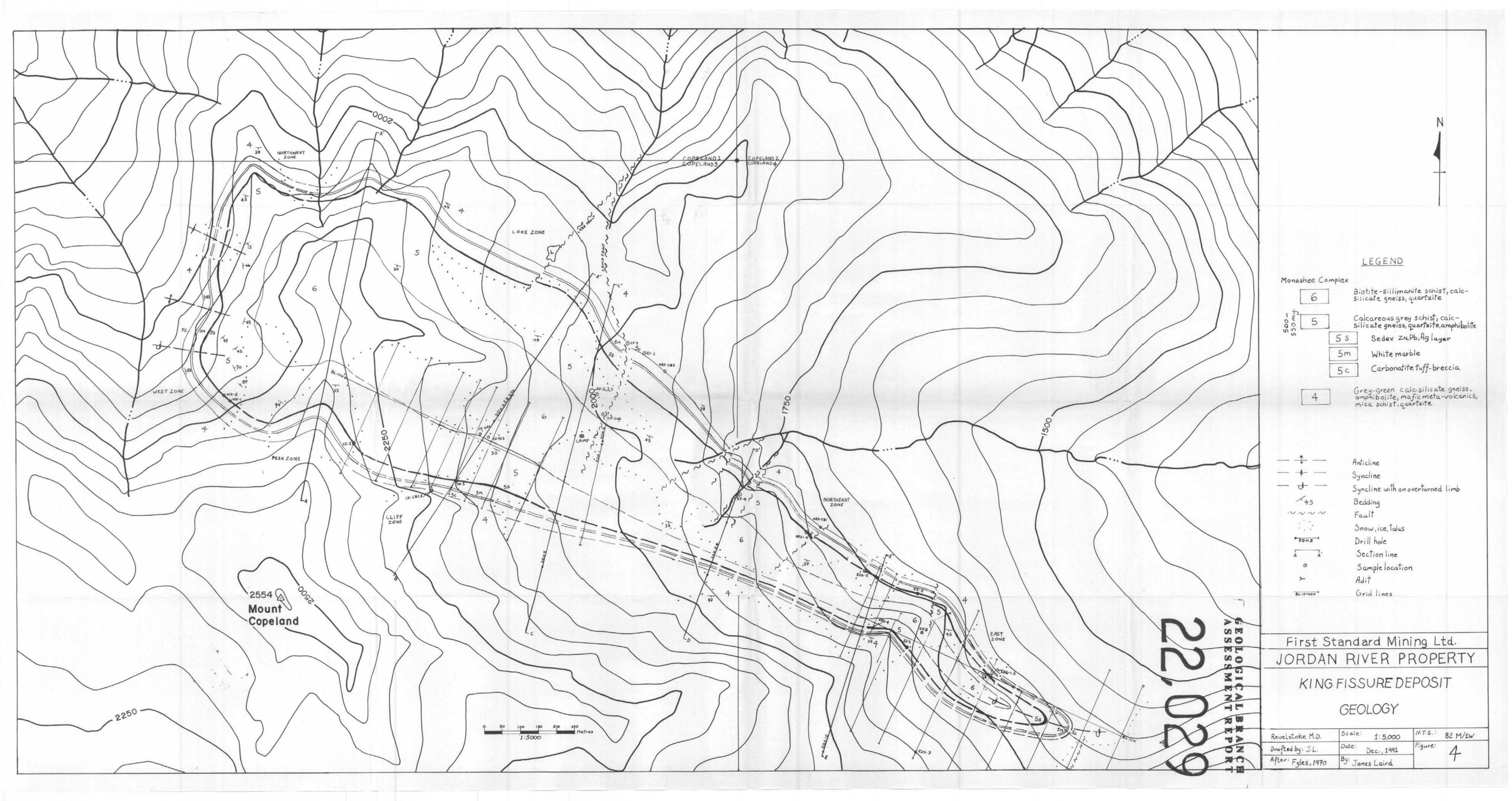
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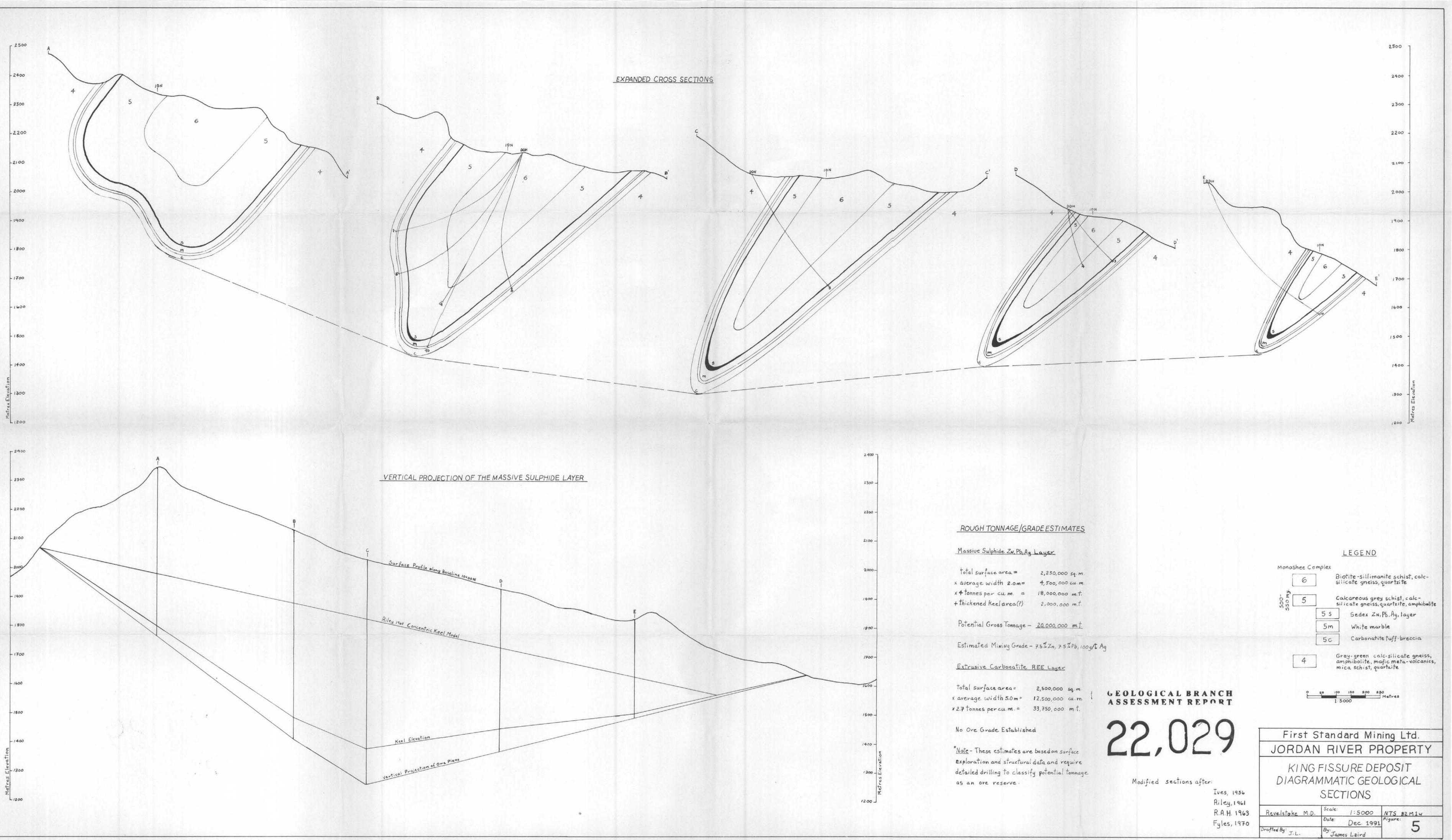
FRISBY RIDGE SOUTH GRID VLF-EM DATA (quadrature)							
	BL35+00E	TL36+00E	TL37+00E	TL38+00E	TL39+00E	TL40+00E	
64+00N	14	-6	-15	8	-4	-6	64+00N
	14	-7	~15	-6	-4	-6	
63+50N	10	-6	-18	-4	-2	-8	63+50N
	9	-6	18	-1	3	-8	
63+00N	6	8	~16	1	1	-7	63+00N
	4	-8	-14	3	0	-5	
62+50N	8	-7	-9	4	0	-5	62+50N
	10	-8	_1	6	-1	-6	
62+00N	13	-7	4	5	-2	-6	62+00N
	13	-5	0	1	-3	-5	
61+50N	8	-3	-2	0	-5	_3	61+50N
	3	-2	2	1	-6	-4	
61+00N	0	0	-10	-4	-4	-4	61+00N
	_4	-10	-8	-2	-2	-4	
60+50N	-3	-8	4	1	-2	-5	60+50N
	-3	-8	-5	-2	-6	-7	
60+00N	-4	-10	-8	-6	-9	-6	60+00N
	-5	-14	-11	-11	-9	-3	
59+50N	-5	-16	-14	-14	-7	3	59+50N
	-4	18	-16	-14	-4	6	
59+00N	-3	-19	-14	-10	1	1	59+00N

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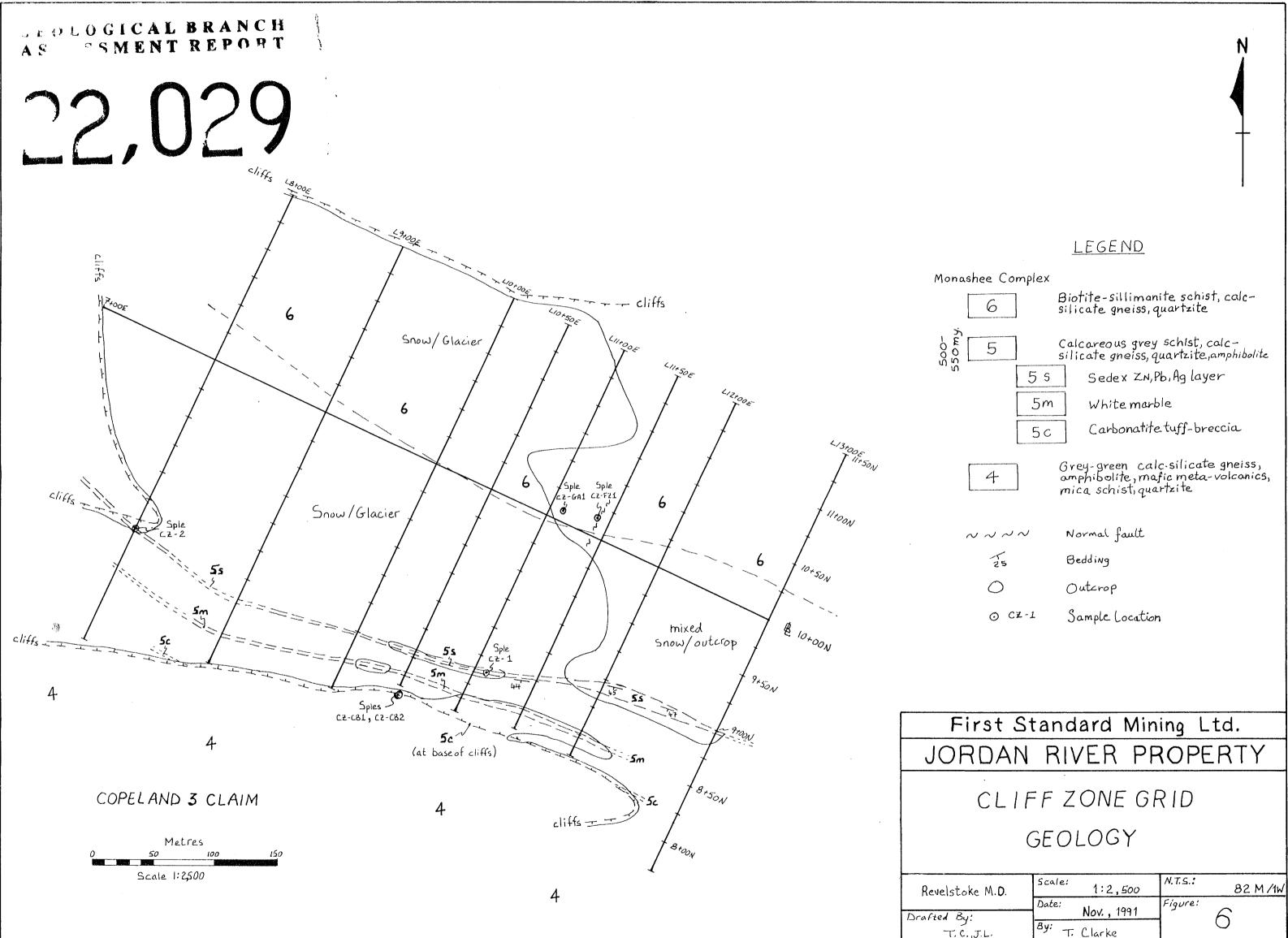




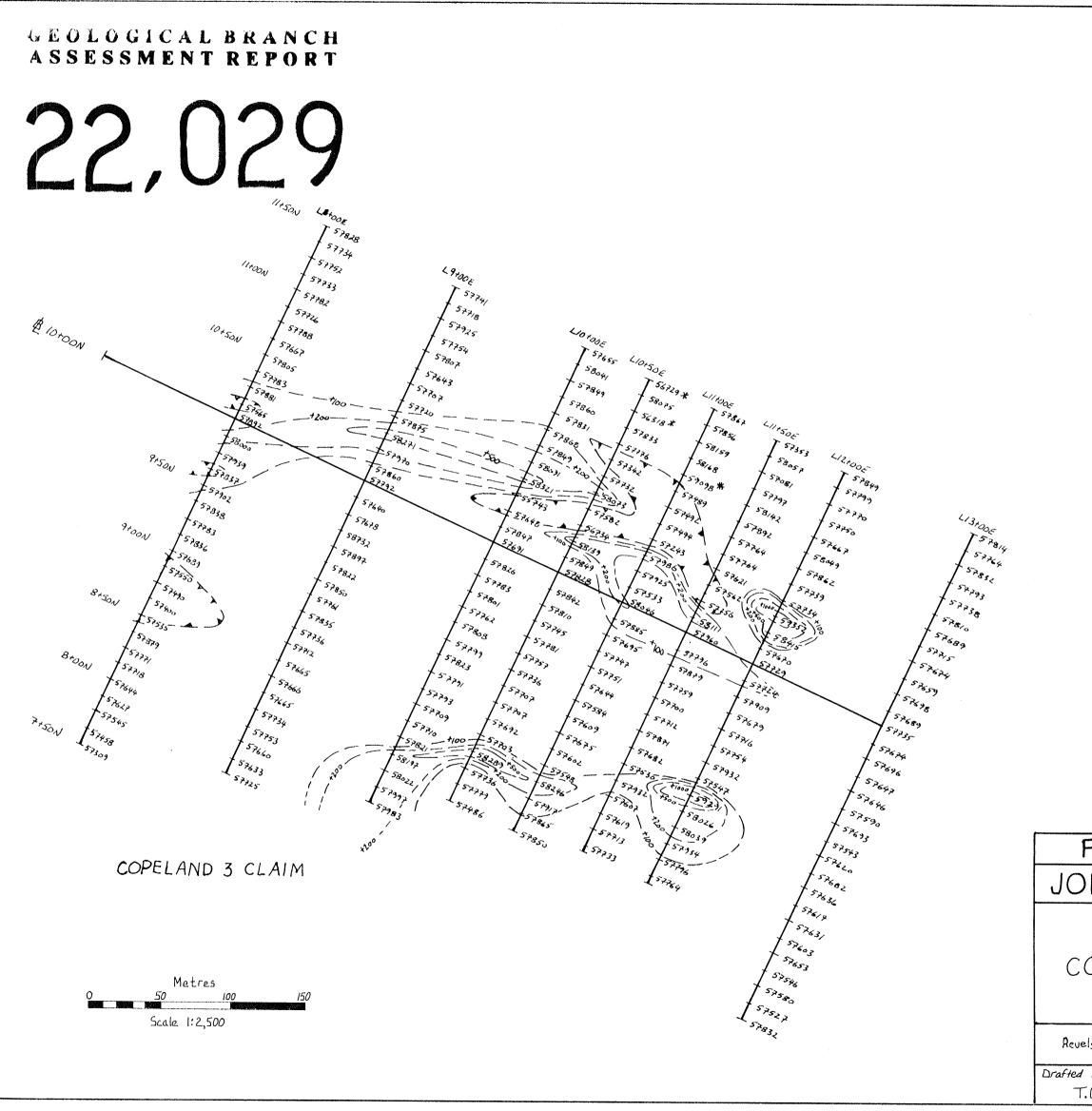
total surface area =	2,250,000 sq.m.
x average width 2.0m=	4,500,000 cu.m.
x 4 tonnes per cu.m. =	18,000,000 m.t.
+ thickened Keelarea (?)	2,000.000 m.t.

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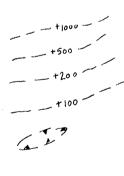
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Revelstoke M.D.	Scale: 1: Date: 0
afted By: J.L.	By: Tames La



elstoke M.D.	Scale: 1:2,500	N.T.S.: 82 M /1W
d By:	Date: Nov. , 1991	Figure:
т.с., <i>J</i> .L	By: T. Clarke	0



LEGEND



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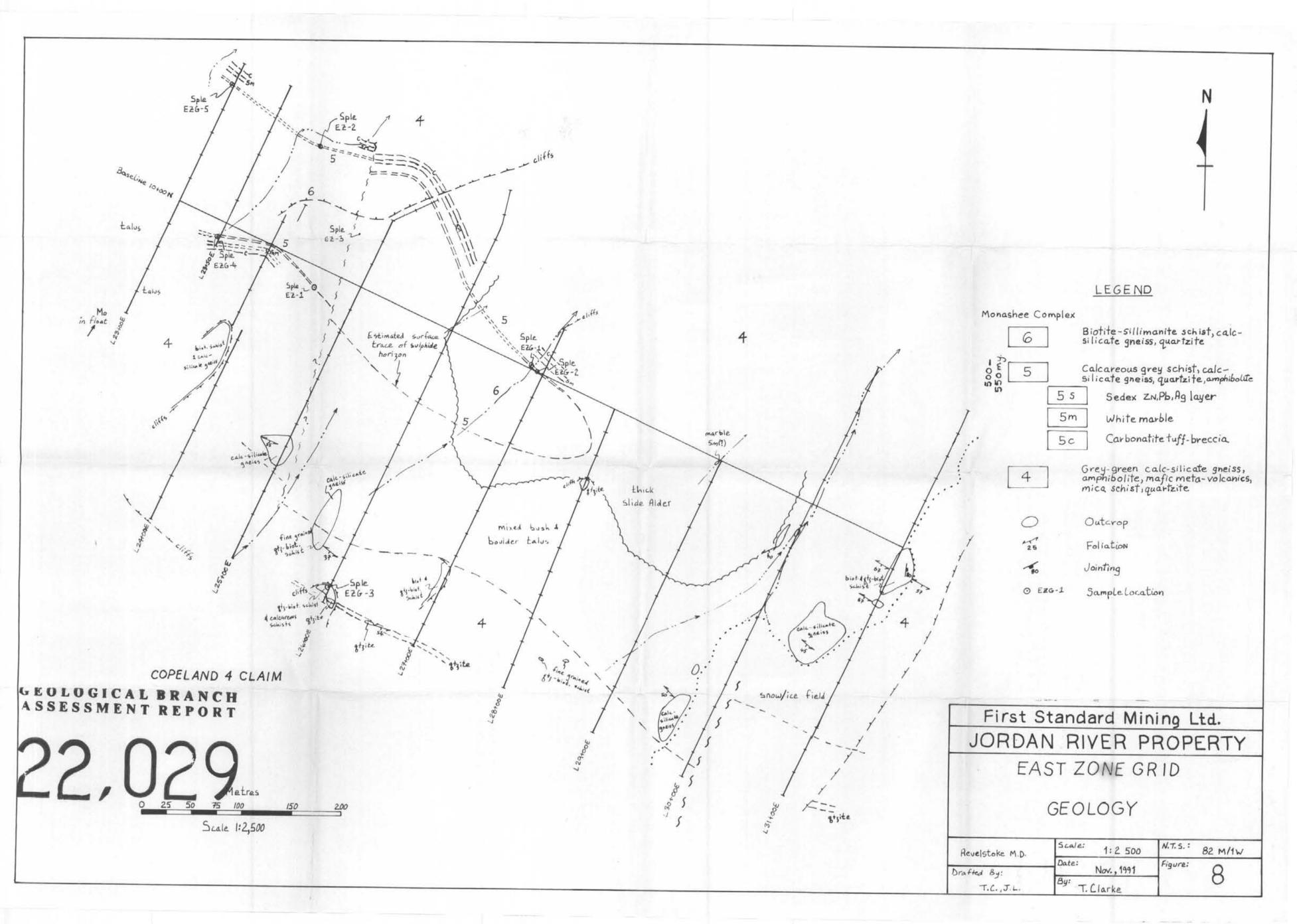
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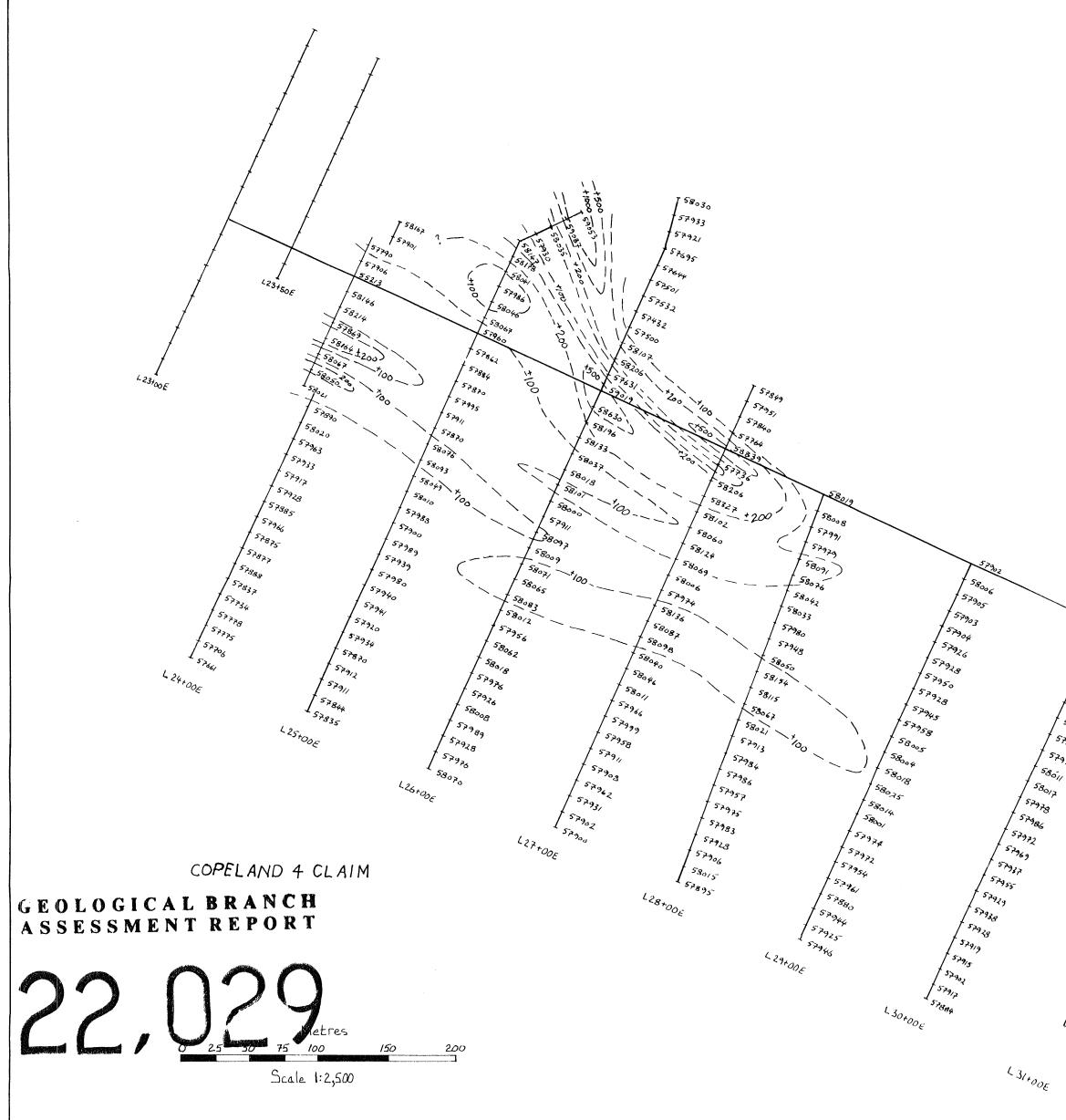
Suspect readings

First Standard Mining Ltd. JORDAN RIVER PROPERTY

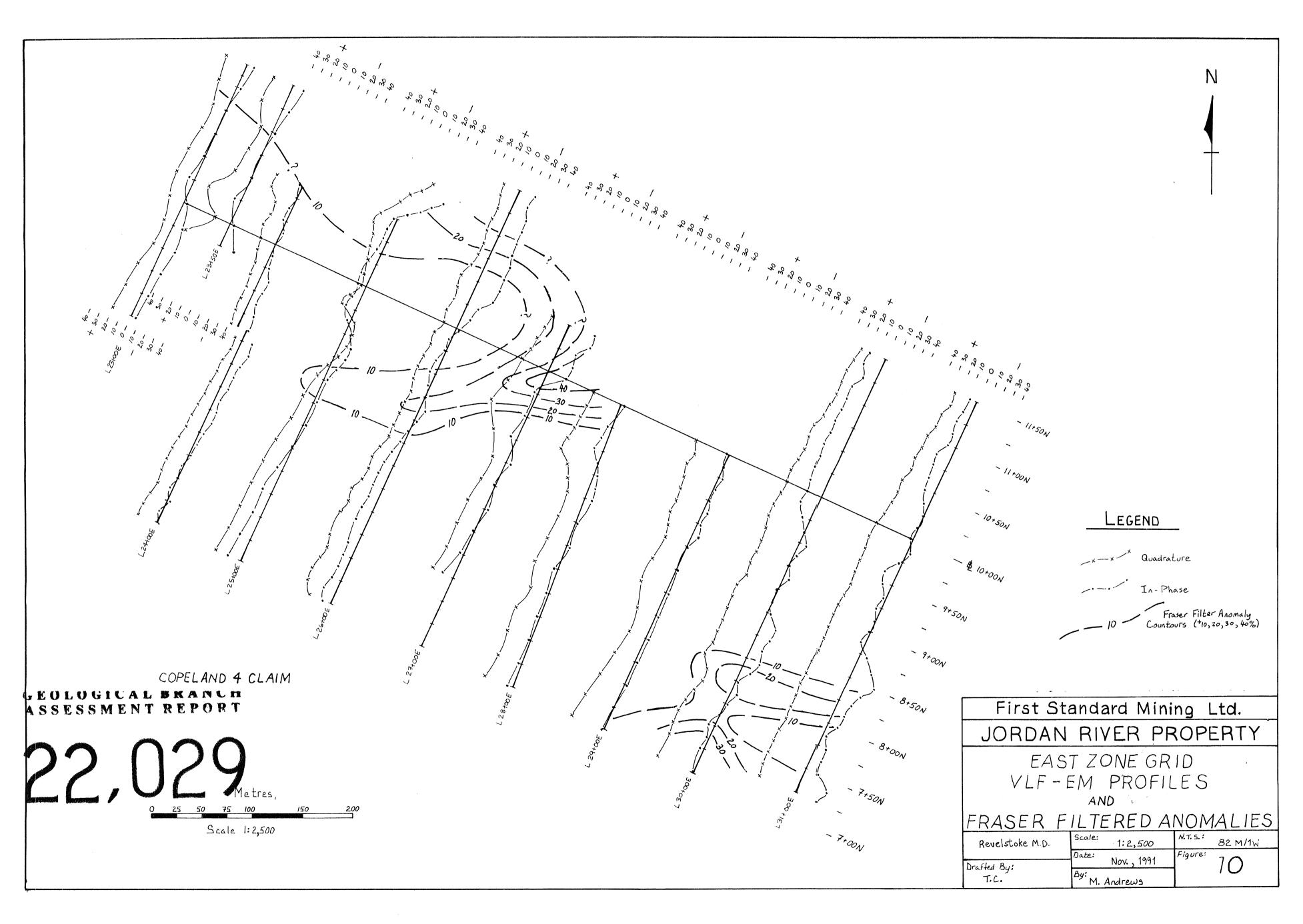
CLIFF ZONE GRID CORRECTED MAGNETICS (TOTAL FIELD IN GAMMAS)

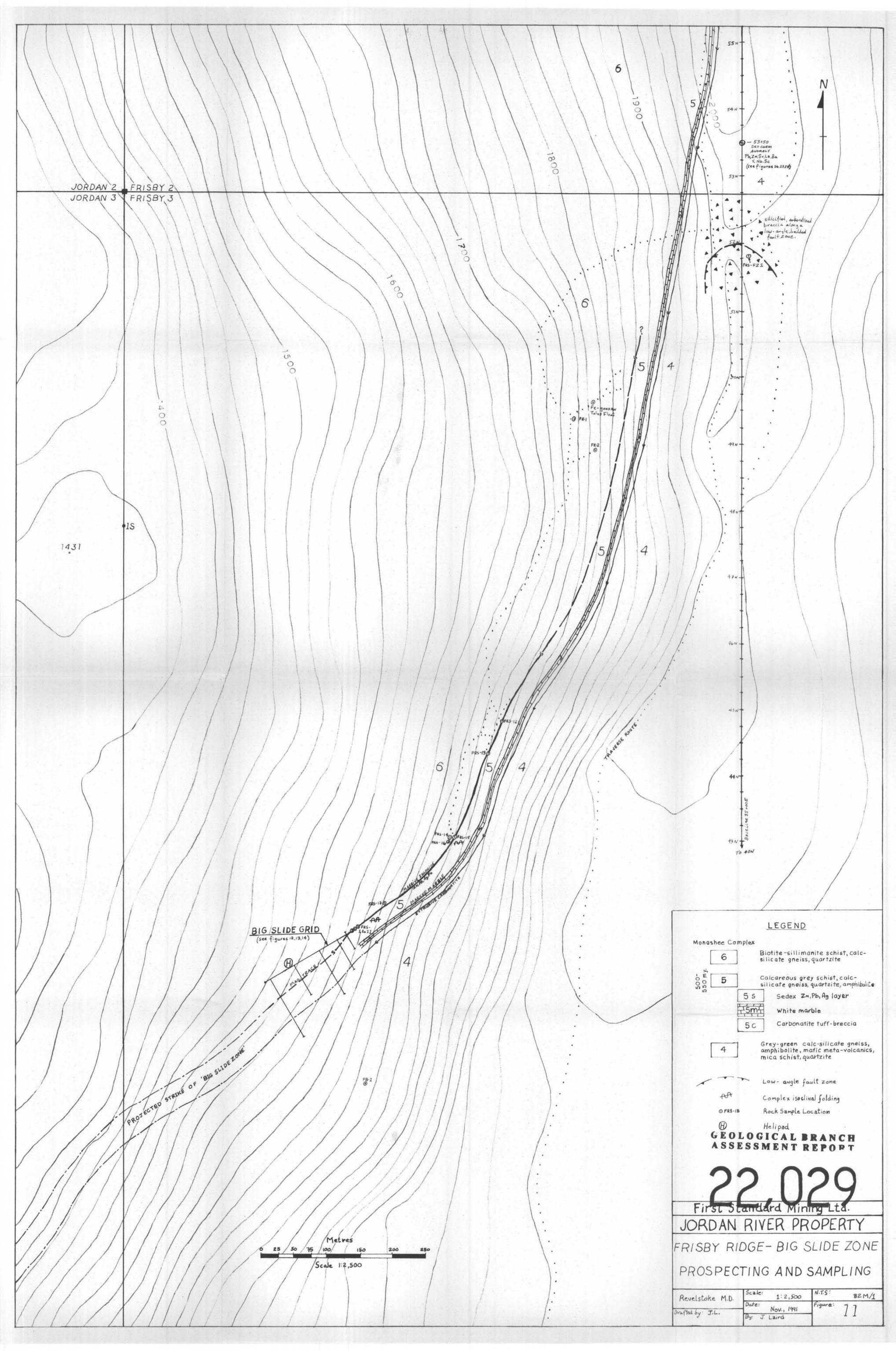
lstoke M.D.	Scale: 1:2,500	N.T.S.: 82 M/1W
<i>By:</i> 50.	Date; Nov.,1991 ^{By:} T. Clarke	Figure: 7

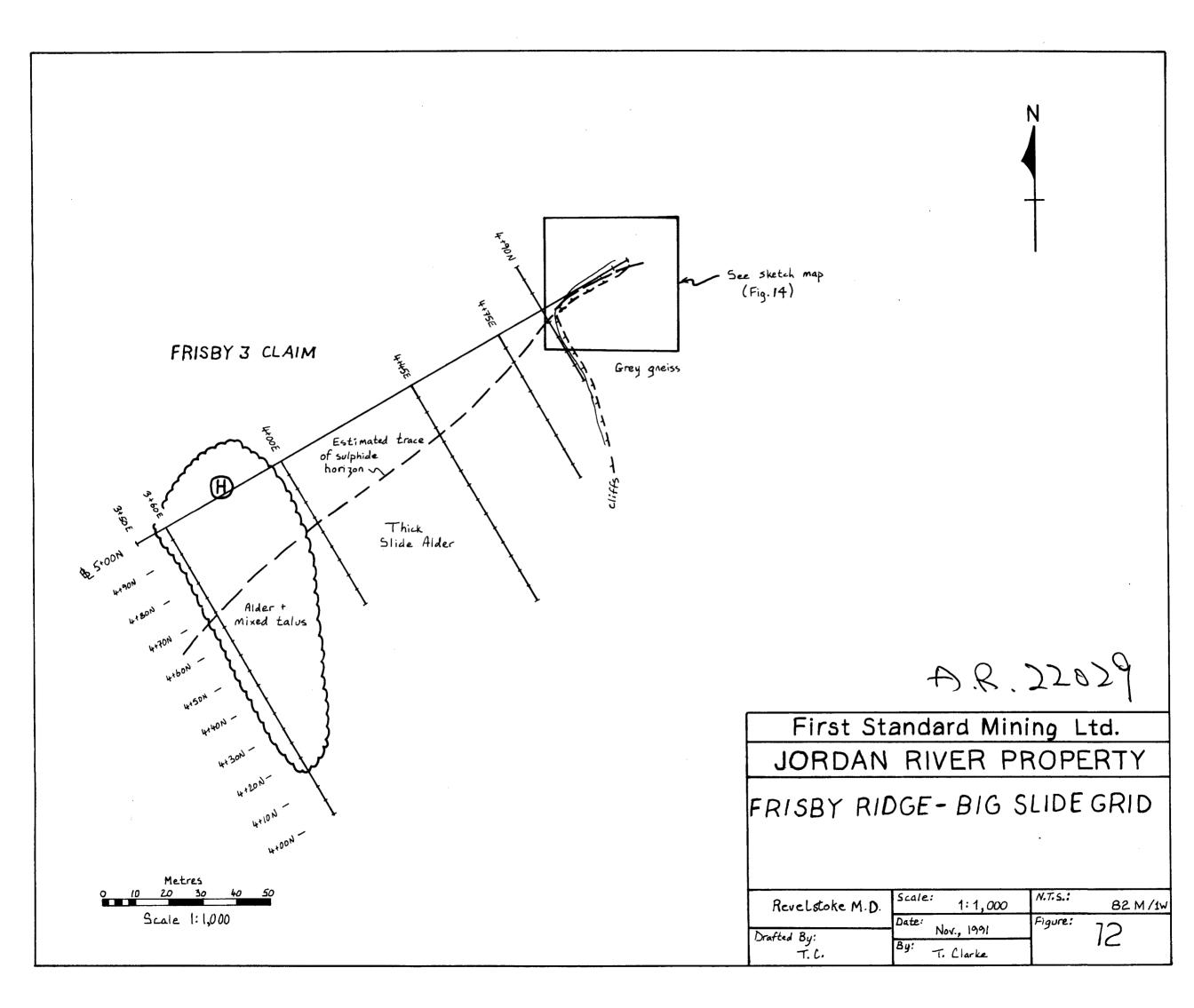


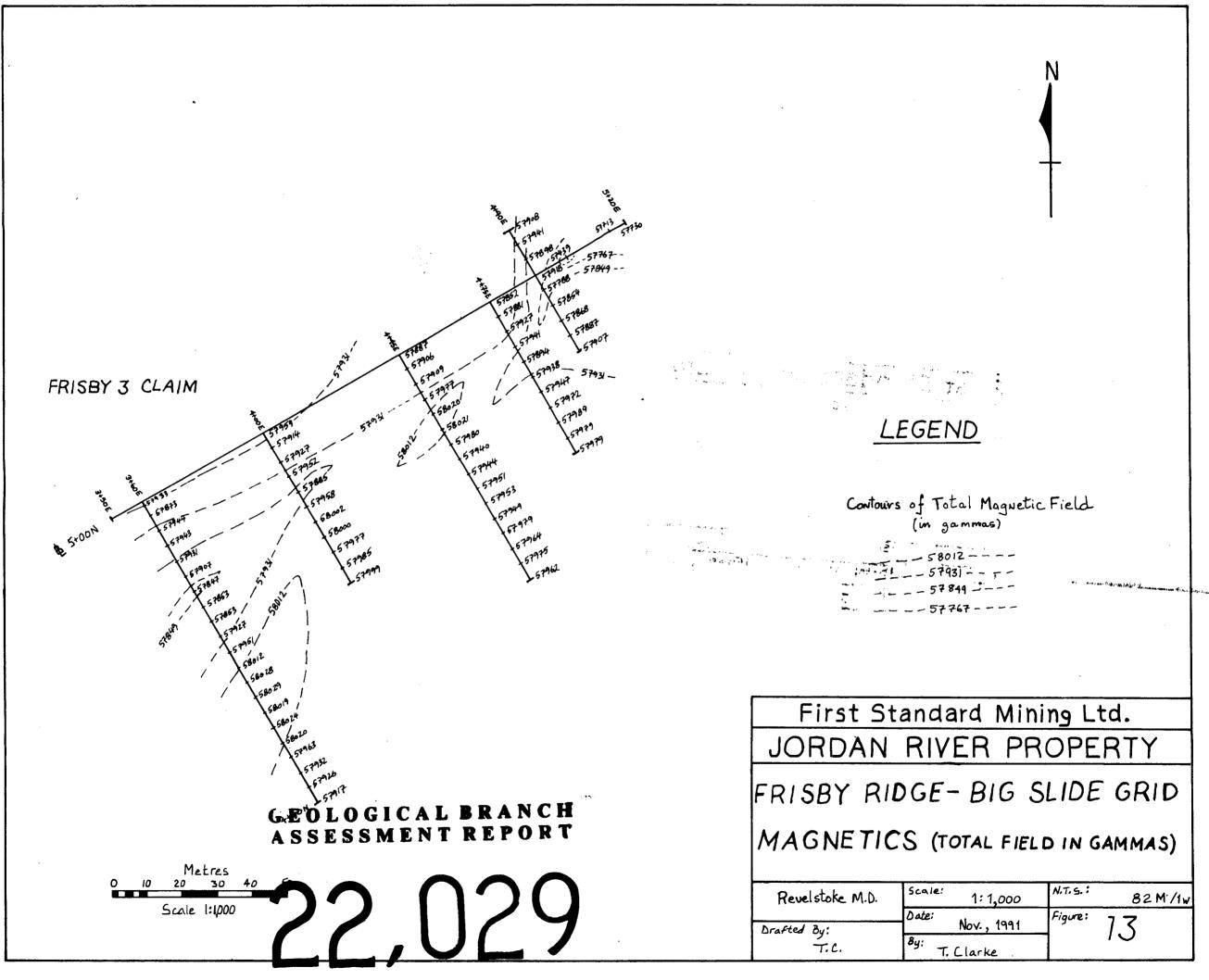


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5-29-2 5-29-2	EAST ZONE GRID
SAR TAS	MAGNETICS (TOTAL FIELD IN GAMMAS)
~ > + OON	After: Scale: 1:2.500 N.T.S.: 82. M/IW
	Drafted By: Date: Nov., 1991 Figure: 9
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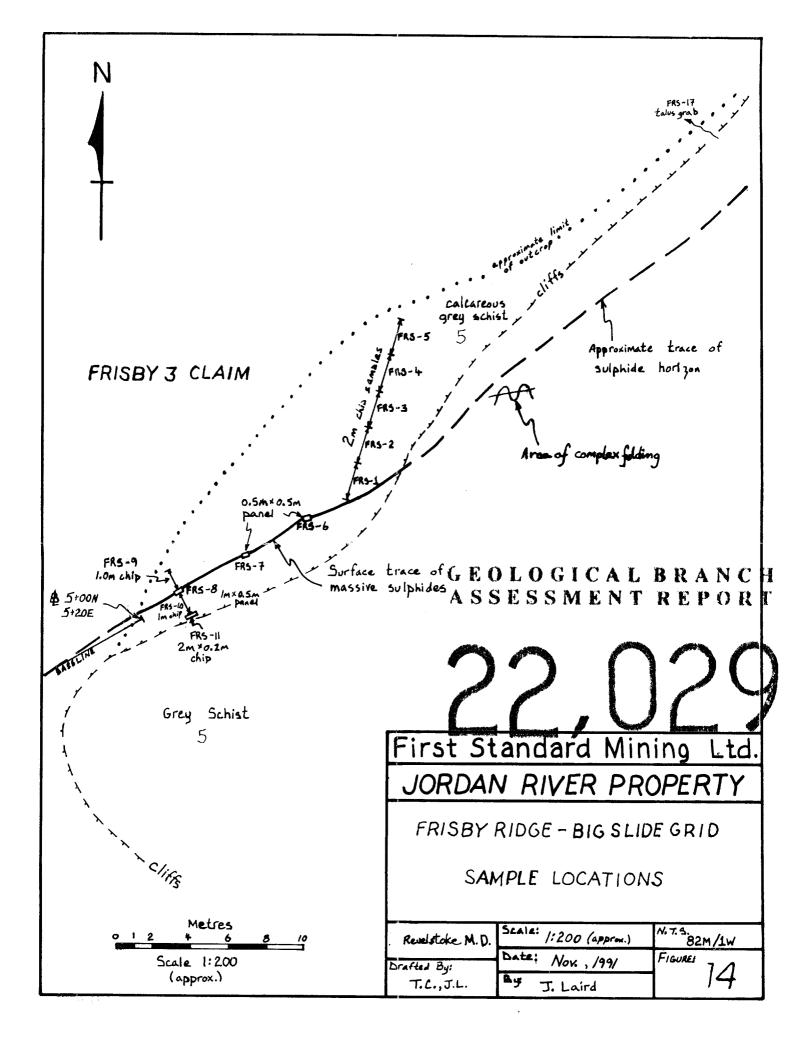


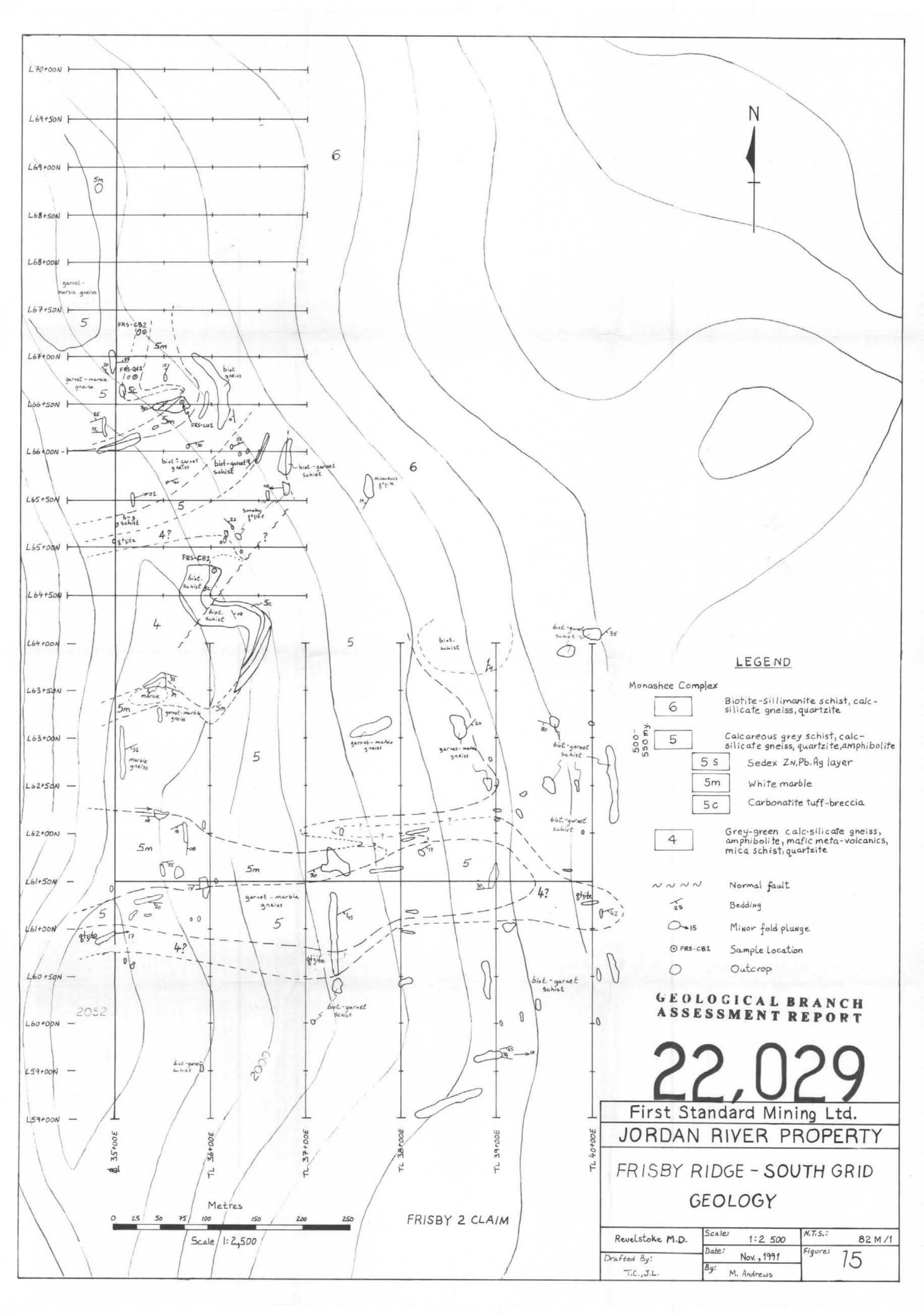


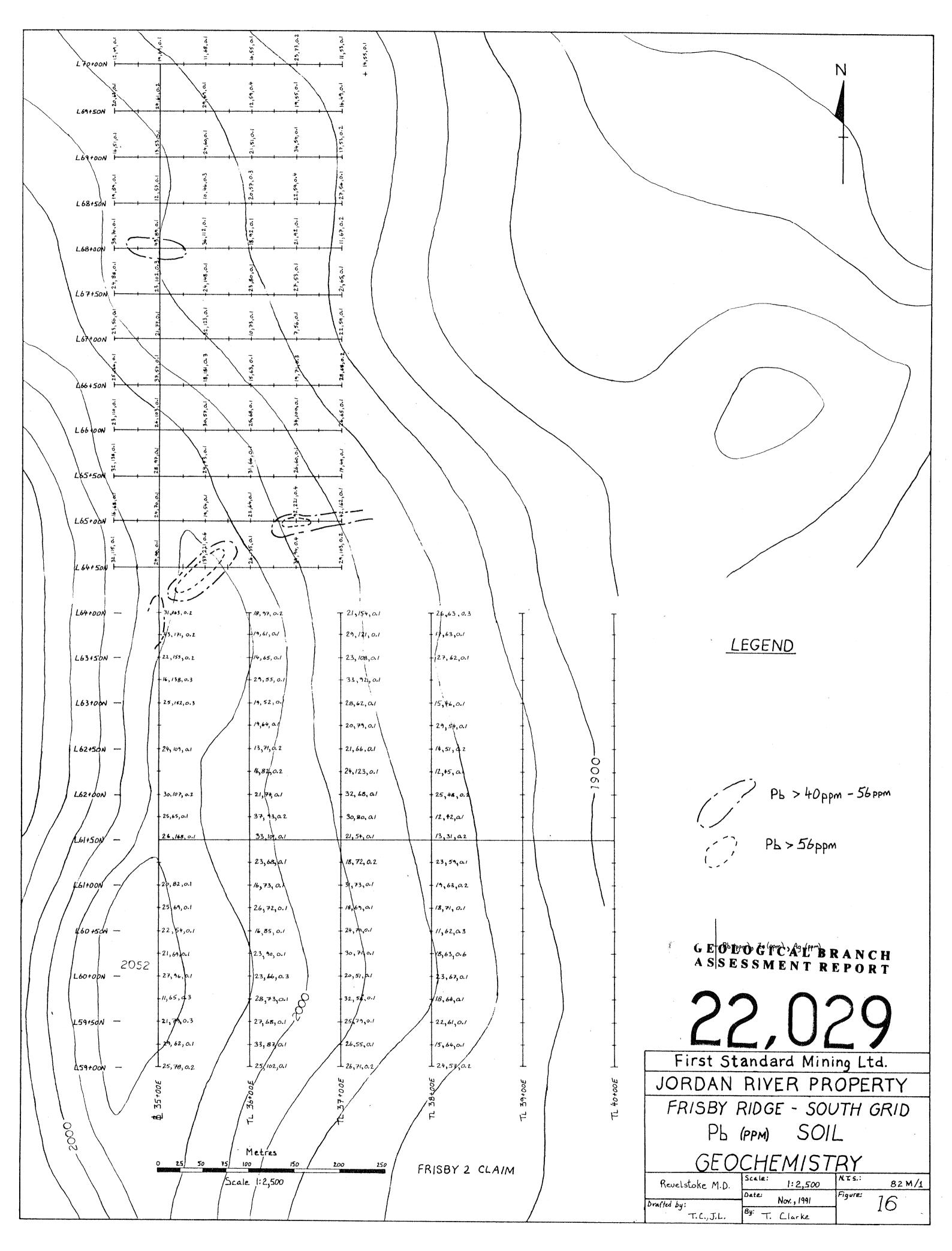


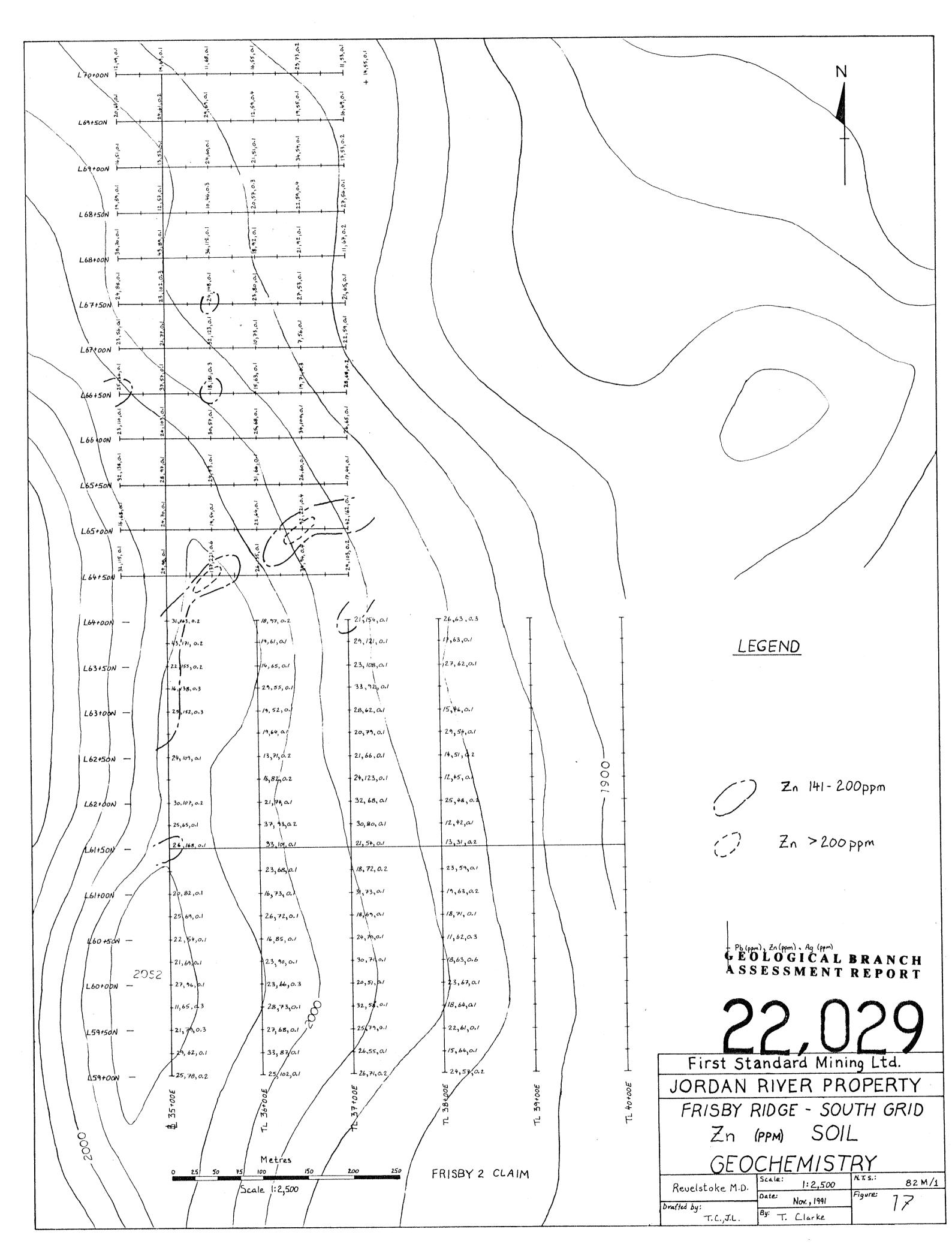


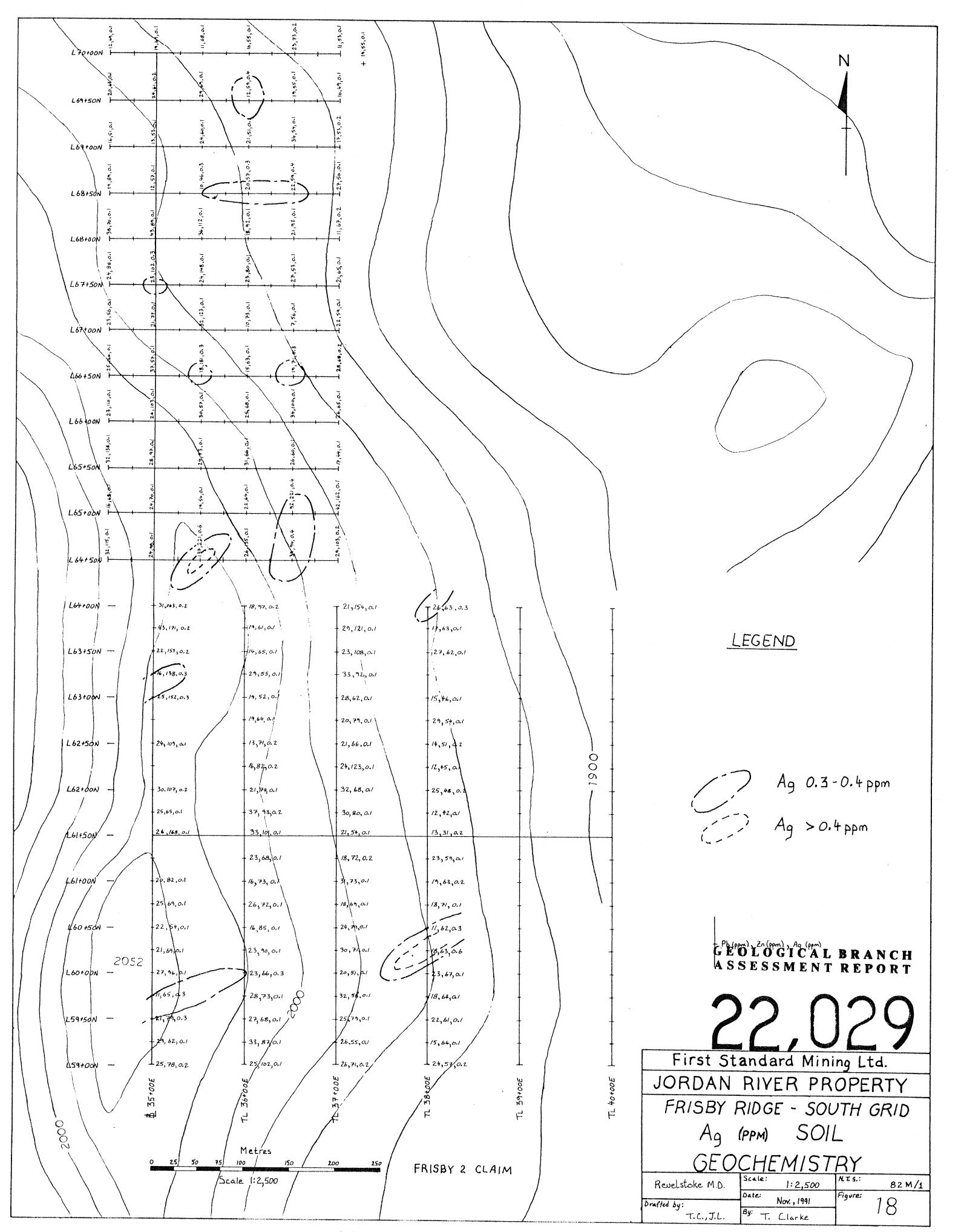
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