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VANCOUVER, B.C.**

Cascadia International Ventures Inc.

**1997 DRILLING PROGRAM
ON THE FAWN 1-7 CLAIMS**

Volume I - Text

Located on the Nechako Plateau
Omineca Mining Division
NTS 93F/3E
53° 12' North Latitude
125° 08' West Longitude

-prepared for-

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**GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT**

October, 1997

25,190

SUMMARY

The Fawn property covers 140 contiguous claim units (3500 hectares) in the Nechako Plateau of central British Columbia, approximately 120 kilometres southwest of Vanderhoof and 180 kilometres west of Quesnel. The Kluskus-Malaput Forest Road passes through the claim group, along with a network of tributary roads and logging spurs. Topography is moderate and lower elevations have been partially clear-cut.

BP Minerals Limited carried out extensive soil sampling over the Fawn property in 1982 and 1983, identifying a northwesterly trend of coincident lead+zinc+silver+arsenic anomalies over 700 x 3,000 metres. Prospecting, mapping and geophysical surveying by Western Keltic Mines Inc. in 1991 showed these anomalies to correspond to four east-west VLF-EM conductors with a minimum aggregate length of 6,500 metres. A hand trench over one of these conductors (V2) revealed sericite-clay alteration with epithermal quartz stockwork which averaged 0.6 g/tonne Au across a true width of 8.2 metres in the Giver Zone. In 1994, Western Keltic drilled 617 metres in widely spaced reconnaissance holes, returning intersections of 8.1 metres of 2.0 g/tonne Au and 4.4 metres of 1.5 g/tonne Au from the Giver Zone. In March and April of 1997, Cascadia International Ventures Inc. drilled a further 620 metres on the Fawn property, consisting of five 150-metre stepouts along the V2 conductor from the Giver Zone and two holes on the Giver Splay, a weakly conductive cross-structure nearby.

The Fawn property is underlain by a sequence of Early to Middle Jurassic Hazelton Group rhyolitic and andesitic volcanics with lesser epiclastic sediments. These have been intruded by a Late Cretaceous diorite and by later felsic dykes. Drilling shows that VLF-EM conductor V2 is caused by a major, steeply north-dipping, fault zone enveloped by 15-30 metres of intense clay-sericite alteration with local chalcedonic quartz stockwork and chalcedony breccias. Sulphide content is variable, consisting mainly of pyrite, lesser arsenopyrite and minor chalcopyrite, sphalerite and galena. Cretaceous or Tertiary quartz-feldspar rhyolite dykes are locally emplaced along the V2 fault zone, as well as faulted and brecciated by it. There is evidence of repeated faulting, dyke emplacement and hydrothermal activity along the fault zones, with several generations of veining and brecciation.

To date, 900 metres of conductor V2 have been tested by drilling, with each of the eight holes intersecting wide clay-sericite alteration zones, variable amounts of quartz veining and gold values in the range of 100-3000 ppb. The best 1997 hole intersected 10.2 metres grading 1.08 g/tonne Au. The remaining 1000+ metres of conductor V2 have not been drilled, nor have the 4,600+ metres along three parallel conductors, although each is accompanied by soil geochemical anomalies of similar intensity to that associated with V2. In addition, it is possible that the fault/alteration zones may be more focused, with narrower widths of better mineralization, where they pass eastwards out of andesitic lapilli tuffs into rhyolite. An analogue for this would be the Tommy epithermal deposit, 17 kilometres to the south, where similar Hazelton Group rhyolites host 478,000 tonnes grading 8.7 g/tonne Au across an average width of four metres.

1997 DRILLING PROGRAM ON THE FAWN 1-7 CLAIMS

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1.0 INTRODUCTION

The Fawn property is located on the Nechako Plateau, approximately 120 kilometres southwest of Vanderhoof in central British Columbia. It is underlain by felsic and andesitic Hazelton Group volcano-sedimentary rocks cut by the Late Cretaceous Capoose Lake Batholith and by feeder dykes to the Eocene Ootsa Lake Group felsic to andesitic volcanics. BP Minerals Ltd. carried out geological mapping, soil sampling and backhoe trenching on the property from 1981 to 1984, defining coincident zinc-silver-lead soil anomalies over an area of 3000 metres by 700 metres. It was restaked as the Fawn property and Western Keltic Mines Inc. conducted mapping, prospecting, soil sampling, geophysical surveys and 617 metres of diamond drilling from 1991 through 1994. Four open-ended, subparallel VLF-EM conductors, with a total strike length of 6,400 metres, were defined within the soil geochemical anomaly. Drilling on one of these, the Giver Zone, showed it to correspond to epithermal chalcedony stockwork/breccia within a 18-32 metre wide zone sericite-clay alteration; the best intersection assayed 2.0 g/t Au across 8.1 metres.

A 620 metre diamond drilling program was carried out in March and April of 1997, to intersect the Giver Zone conductor along strike from the 1994 drilling and to test one of its splays which had yielded auriferous subcrop mineralization. Equity Engineering Ltd. conducted this drill program for Cascadia International Ventures Inc. and has been retained to report on the fieldwork. Limited mapping and soil sampling were carried out in conjunction with the drillsite reclamation in September 1997 and will be reported separately.

2.0 LIST OF CLAIMS

The Fawn property comprises seven contiguous claims totalling 140 claim units (3,500 hectares), located in the Omineca Mining Division (Figure 2). Records of the British Columbia Energy and Minerals Division indicate that the Fawn 1-7 claims are owned by Western Keltic Mines Inc.. Separate documents indicate that Cascadia International Ventures Inc. has an option to earn an interest in them. Claim data for the Fawn property is summarized in Table 2.0.1.

Table 2.0.1
CLAIM DATA

Claim Name	Mineral Tenure No.	No. of Units	Record Date	Expiry Year
Fawn 1	243221	20	March 15, 1991	2007*
Fawn 2	301430	20	June 26, 1991	2007*
Fawn 3	301431	20	June 26, 1991	2007*
Fawn 4	301432	20	June 26, 1991	2004
Fawn 5	305450	20	October 13, 1991	2007*
Fawn 6	322193	20	October 28, 1993	2007*
Fawn 7	323869	20	February 26, 1994	2004
		140		

* Subject to approval of assessment work covered by this report.

The position of the legal corner posts for the Fawn 1-5 claims has been verified by the authors.

3.0 LOCATION, ACCESS AND GEOGRAPHY

The Fawn property is situated on the Nechako Plateau of central British Columbia, approximately 120 kilometres southwest of Vanderhoof and 180 kilometres west of Quesnel (Figure 1). The claims are



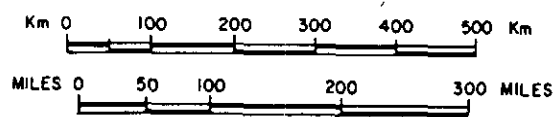
CASCADIA INTERNATIONAL VENTURES INC.

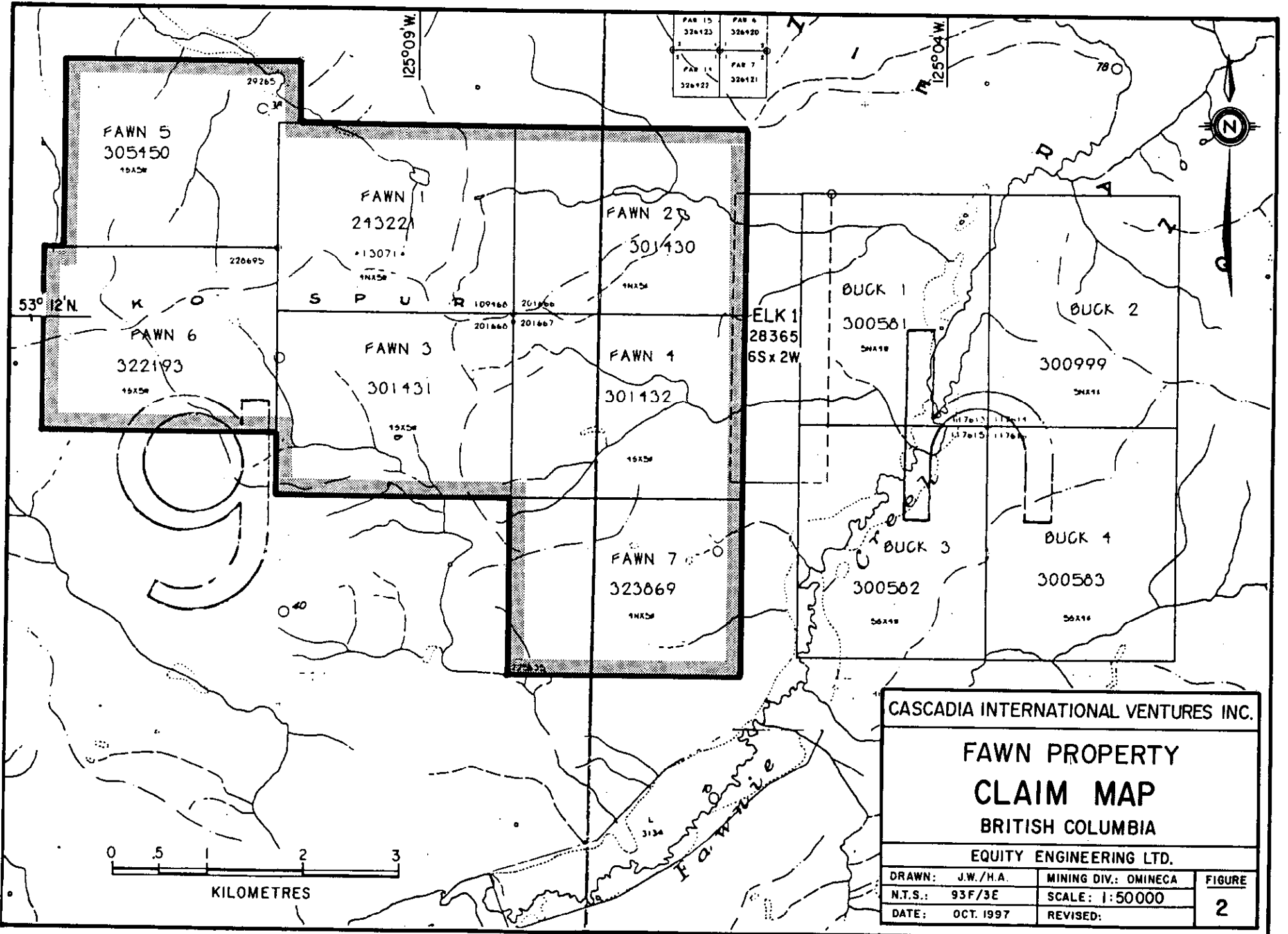
FAWN PROPERTY LOCATION MAP

BRITISH COLUMBIA.

EQUITY ENGINEERING LTD.

DRAWN: J.W./H.A.	MINING DIV. OMINECA	FIGURE
N.T.S.: 93F/3E	SCALE: As Shown	1
DATE: OCT. 1997	REVISED:	





PAR 15 326123	PAR 6 326120
PAR 14 326122	PAR 7 326121

FAWN 5
305450
15x25

FAWN 1
24322
13071
15x25

FAWN 2B
301430
15x25

FAWN 6
322193
15x25

FAWN 3
301431
15x25

FAWN 4
301432
15x25

ELK 1
28365
6Sx2W

BUCK 1
300581
56x18

BUCK 2
300999
56x18

FAWN 7
323869
15x25

BUCK 3
300582
56x18

BUCK 4
300583
56x18



CASCADIA INTERNATIONAL VENTURES INC.		
FAWN PROPERTY CLAIM MAP BRITISH COLUMBIA		
EQUITY ENGINEERING LTD.		
DRAWN: J.W./H.A.	MINING DIV.: OMINECA	FIGURE
N.T.S.: 93F/3E	SCALE: 1:50000	2
DATE: OCT. 1997	REVISED:	

located within the Omineca Mining Division, centred at 53° 12' north latitude and 125° 08' west longitude.

The property is accessed by a major logging road, the Kluskus-Malaput Forest Road, which reaches the north side of the property 146 kilometres south of the Plateau Forest Products mill at Engen on Highway 16. The Kluskus-Malaput road angles through the southeastern corner of the property, while a major branch, the Van Tine Forest Road, provides good access through its northern part. The M-4000 Forest Road, another major branch, leaves the Kluskus-Malaput south of the property and angles northwesterly through the southwestern corner of the Fawn 6 claim. Spur roads provide four-wheel drive access throughout each of several recent clear-cuts on the property. The Capoose access road, on the north side of Van Tine Creek, is also accessible by four-wheel drive vehicle but has not been maintained for several years.

The claims cover the eastern portion of Entiako Spur, a range of rolling hills lying south of Van Tine Creek within the Nechako Plateau. Upland surfaces are generally well drained with few lakes or marshes. Lower creek valleys are broad and swampy. Topography is moderate, with elevations ranging from 1,100 metres in the Fawnie Creek valley to 1,739 metres at the highest point of Entiako Spur. Outcrop exposure is fairly good along the ridge top, but is increasingly masked by glacial till at lower elevations. Overall, the property would average less than 5% outcrop. Road cuts along the Van Tine Road expose up to 30 metres of glacial till. Glacial striae trend 060° on the Fawn 2 claim, and Tipper (1963) provides strong evidence regionally for a southwestern ice source.

The property is largely covered by spruce and lodgepole pine with a light undergrowth of huckleberry and alder. Recent clear-cuts at lower elevations on most of the claims have made the sparse outcrops easier to find and examine. The Fawn property is subject to a continental climatic regime, with warm summers and cold winters. Snowfall is moderate with an accumulation of one to two metres during the winter.

4.0 REGIONAL AND PROPERTY EXPLORATION HISTORY

4.1 Previous Work

The area around the Fawn property received little exploration until the late 1960's, when Rio Tinto Canadian Exploration Ltd. carried out stream and lake sediment sampling surveys throughout the Nechako Plateau, searching primarily for copper-molybdenum porphyry deposits (Hoffman, 1976). Follow-up work on one of their anomalies by Rio Canex (1969-71) and Granges Exploration Ltd./Cominco Ltd. (1976-present) led to the discovery in 1979 of the Capoose silver-lead-zinc deposit approximately seven kilometres north of the Fawn property. Reserves at Capoose have been estimated at 28 million tonnes grading 36 g/tonne silver and 0.9 g/tonne gold (Green and Diakow, 1993).

Following the recognition of a major silver resource at Capoose, BP Minerals Limited staked several other nearby high-priority silver-lead-zinc lake sediment anomalies from Rio Canex's data. Their Gran and Laid claims were staked in 1981 to cover the drainages surrounding Square Lake, a small lake at the head of Van Tine Creek near the northern boundary of the present Fawn 1 claim. Square Lake was extremely anomalous in lead, exceeding the values for the lakes which marked the Capoose deposit (Hoffman, 1976).

In 1982, BP Minerals carried out geological mapping over the area now covered by the Fawn property and laid out a compass and hipchain geochemical grid which used three different numbering systems. An east-west baseline was blazed and numbered from 0+00W to 28+00W, just north of the present Fawn 2 southern boundary. Cross-lines were run to the south from this baseline, with station numbering up to 24+00S. A second baseline was blazed to the north from station 28+00W on the first baseline, which was re-labelled 0+00N 0+00W. Cross-lines were run to the east and west from this second baseline (and labelled accordingly), which extended north to 18+00N. A western tie line was

blazed north-south 2,600 metres to the west of the second baseline, near the western boundary of the current Fawn 1 and 3 claims. This was used to tie in lines 0+00N to 14+00N, which were run west from the second baseline. Lines were also run and numbered east (Lines 14+00N to 20+00N) and west from the western tie line (and labelled east or west relative to the western tie line). A total of 1,152 soil and stream sediment samples were taken in 1982 and a further 1,517 in 1983 from ground currently covered by the Fawn property (Hoffman and Smith, 1982; Smith and Hoffman, 1983 and 1984). Samples were taken initially at 100 metre intervals on lines spaced 100 metres apart, with later infilling to 50 metre intervals in anomalous areas. The soil geochemistry delineated a northwesterly trend of coincident lead-zinc-silver anomalies measuring approximately 3,000 metres by 700 metres, centred on the Fawn 1 claim.

In 1983, limited trenching and a series of 40 backhoe test pits were excavated at 25 metre intervals near the eastern end of the lead-zinc-silver soil anomaly, exposing three or four "rhyodacite lapilli tuff" units with up to 94.5 ppm silver and 880 ppb gold (Smith and Hoffman, 1984). The following year, another grid was established for mapping purposes over the Fawn 1 soil anomaly. A 3,000 metre baseline oriented at 310° was cut and numbered from 0+00N to 30+00N. Cross-lines were run at 035° from the baseline at 200 metre intervals. Further backhoe trenching was carried out in the area of the 1983 trenching and near the western end of the soil anomaly, without encouraging results (Smith, 1985). BP Minerals allowed their claims to lapse in 1988.

The Fawn 1-4 claims were staked in 1991 over BP Minerals' soil geochemical anomaly. In September and October of that year, Western Keltic Mines Inc. carried out geological mapping, soil and rock geochemistry and geophysical surveying, taking 239 rock, 144 soil and 41 deep overburden samples. The 1984 cut baseline was re-established and extended at 130° for 2,425 metres to the southeast. Cross-lines were run towards 040° at 100 metre intervals from 4+00N to 30+00N and at 200 metre intervals from 4+00N to 24+00S, with stations marked every 25 metres. Cross-lines, 500 metres in length, were run at a bearing of 220° from 5+00N to 27+00N at 100 metre intervals. Five widely-spaced lines were extended further to the southwest, in an area to the south of pre-existing coverage and soil samples were taken along them at 50 metre intervals. The BP Minerals soil anomalies were relocated relative to the new grid and verified by 41 soil samples taken from their most anomalous sample locations. Magnetometer and VLF-EM surveys were carried out over 31 line-kilometres of the grid between 2+00S and 30+00N. Deep overburden sampling and MaxMin EM were tested over the Giver Zone, a mineralized VLF-EM conductor (Awmack, 1991).

Four subparallel, easterly-trending VLF-EM conductors were defined along strike lengths of 700 to 2200 metres by the 1991 program, with each remaining open along strike in at least one direction. Each of the four VLF conductors is accompanied by silver+zinc+lead+arsenic soil geochemistry. Eocene(?) epithermal chalcedony-sulphide breccia was found in subcrop and float along one of the VLF conductors, with assays up to 12.9 g/tonne Au and 637 g/tonne silver in separate samples from the "Giver Zone" and one of its splays, the "Giver Splay" (Awmack, 1991). The Fawn 5 and 6 claims were subsequently staked to cover the projected westward extension of these VLF structures.

Western Keltic performed a 20.7 line-kilometre induced polarization survey on lines spaced 200 metres apart from 3+00N to 29+00N in October and November, 1993. This showed low resistivity and weak chargeability along the Giver VLF-EM structure and outlined a strong chargeability anomaly at the eastern end of the survey. Moderate chargeability and low resistivity anomalies were indicated near the northwestern end of the grid, in an area of strong soil geochemistry and two VLF-EM structures (Ballantyne, 1993).

During the course of regional mapping in 1993, the BC Geological Survey discovered the Malaput Showing, a zone of silicification and sericitization located four kilometres southeast of the Giver Zone (Diakow and Webster, 1994). The Fawn 7 claim was subsequently staked over the Malaput Showing.

The BC Geological Survey undertook regional lake sediment (Cook and Jackaman, 1994) and

basal till (Levson et al, 1994) sampling programs throughout portions of the 93F map sheet in 1993, taking three lake sediment samples and 18 till samples from the Fawn property. The lake sediment sample from Square Lake returned the highest lead, zinc and cobalt values for all 237 samples taken from the region, along with anomalous antimony, arsenic and gold. Six of the till samples exceeded the survey's 95th percentile for gold, lead, arsenic or antimony. Four of these anomalous till samples, including the sample with the survey's second highest gold value, were taken from the northeastern portion of the Fawn 7 claim, an area which has received no exploration.

In 1994, Western Keltic drilled 617 metres in six diamond drill holes on geophysical and geochemical targets on the Fawn 1 and Fawn 5 claims. Three of these were drilled across the V2 conductor (Giver Zone) showing it to be a steeply-dipping 18-32 metre wide zone of sericite-clay-pyrite alteration hosting epithermal chalcedony stockworks and breccias. The best intersections were 8.1 metres of 2.0 g/tonne Au in hole FWN94-02 and 4.4 metres of 1.5 g/tonne Au and 63.8 g/tonne Ag in hole FWN94-03. In addition, 55 soil samples were taken from a small grid over the Malaput Showing, returning up to 255 ppb Au, 336 ppm As, 226 ppm Pb and 1360 ppm Zn. Mapping showed it to be an easterly-trending, 25-30 metre wide zone of silicification which can be traced along strike for at least 300 metres (Baknes and Awmack, 1994a).

4.2 1997 Diamond Drilling Program

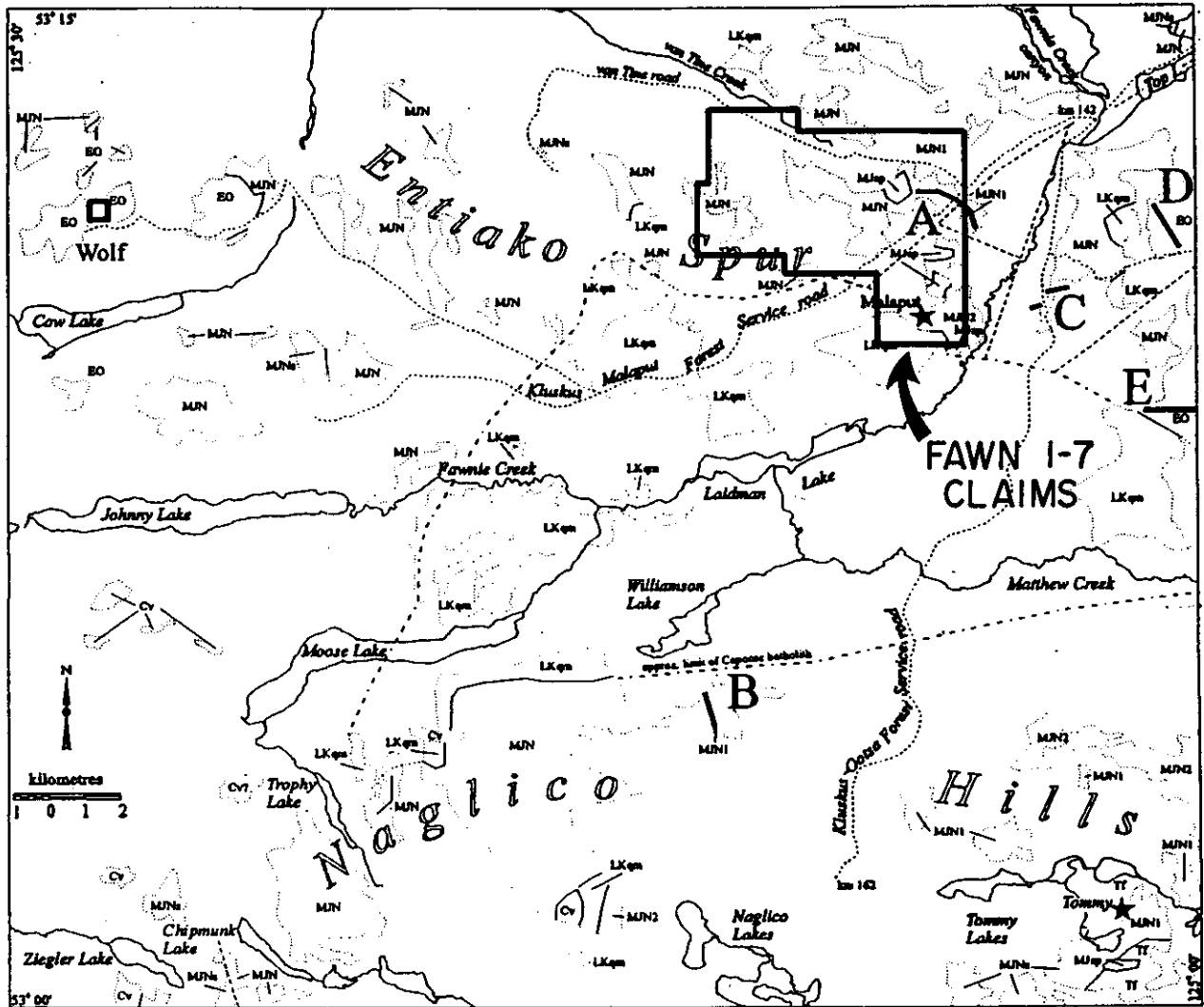
During March and April of 1997, Cascadia International Ventures Inc. carried out a second diamond drill program on the Fawn property, targeted at the Giver Zone and Giver Splay on the Fawn 1 claim. Seven holes, totalling 619.6 metres (2,032') of BTW core, were drilled by Falcon Drilling Ltd. of Prince George, using their F-1000 drill. Core was logged on site and split mechanically for geochemical analysis. Drill logs are attached in Appendix C. Drill sites were accessed by short spur roads built using a small cat from existing roads. A total of 125 core samples were analyzed geochemically for gold and by ICP for 32 elements by Chemex Laboratories in North Vancouver. Samples with high gold values were subsequently fire assayed. Appendix D contains analytical certificates.

Reclamation of all drill sites and drill roads was carried out in September 1997. Limited soil sampling and mapping was performed in conjunction with the reclamation, focused on the Malaput Showing area and skarn mineralization on the Fawn 6 claim. Results of this program will be reported separately.

5.0 REGIONAL GEOLOGY

The British Columbia Geological Survey carried out 1:50,000 scale regional mapping over map-sheet 93F/6 in 1992 (Green and Diakow, 1993; Diakow and Green, 1993). In 1993, this mapping was extended to the south over map-sheet 93F/3, which covers the Fawn property (Diakow and Webster, 1994; Diakow et al, 1994). Their mapping shows Jurassic Hazelton Group volcanics and sediments intruded by the Late Cretaceous Capoose Lake batholith and unconformably overlain by Eocene Ootsa Lake Group subaerial volcanics and younger plateau basalts (Figure 3).

The Early to Middle Jurassic Hazelton Group rocks in the vicinity of the Fawn property have been assigned by Diakow and Webster (1994) to their informal Naglico Formation of silica-bimodal volcanic rocks and Bajocian intravolcanic sediments which are gradationally overlain by Callovian marine sediments. The lower division of this formation consists of "crudely layered fragmental and lesser flow rocks of rhyolitic composition, and local maroon and green andesitic tuffs deposited in a subaerial environment" (Unit MJN1). The upper division is dominated by mafic and intermediate lavas (Unit MJN2), interpreted by Diakow and Webster (1994, p. 19) to be deposited in a shallow marine environment with local subaerial conditions. Green and Diakow (1993) report that a section of the upper division exceeds 1,000 metres in thickness on Tutiai Mountain, 14 kilometres north of the Fawn property. Augite porphyry plugs (Unit MJap) mapped on the Fawn claims are thought to be cogenetic with upper



LEGEND

**STRATIFIED ROCKS
MIOCENE TO PLIOCENE**

Chilcotin Group

Cv Olivine basalt

Eocene

Ootsa Lake Group

EO Rhyolite and andesite flows, quartz-bearing lapilli tuffs, tuffaceous siltstone

MIDDLE JURASSIC

Hazelton Group (Naglico Formation)

MJNs Fine to coarse-grained, fossiliferous volcanoclastics

MJN2 Basalt and andesite flows and lapilli tuffs

MJN1 Rhyolite flows, ash-flow tuffs and lapilli tuffs

INTRUSIVE ROCKS

TERTIARY

TF Felsite sills

LATE CRETACEOUS

Capoose Lake Batholith

LKqn Equigranular quartz monzonite, with lesser quartz diorite and quartz porphyry

MIDDLE JURASSIC

MJap Mafic augite-plagioclase porphyry plugs

Geology modified from Diakow and Webster (1994).

SYMBOLS

- Geological contact.....
- Fault.....
- Potential epithermal prospect.....★
- Geological section.....
- Outcrop limit.....

CASCADIA INTERNATIONAL VENTURES INC.

**FAWN PROPERTY
REGIONAL GEOLOGY**

BRITISH COLUMBIA

EQUITY ENGINEERING LTD.

DRAWN: J.W./H.A.	MINING DIV.: OMINECA	FIGURE 3
N.T.S.: 93 F/3	SCALE: 1:200,000	
DATE: OCT. 1997	REVISED:	



division Naglico Formation augite-phyric volcanics.

Wide-spread, irregularly-distributed, marine sedimentary rocks (**Unit MJNs**) are intercalated with Naglico Formation volcanics, interpreted as basins between coalescing volcanic centres. The marine sediments become dominant in the stratigraphically highest Middle Jurassic exposures. Main lithologies include feldspathic sandstone and siltstone, tuffaceous argillite, locally prominent volcanic conglomerate and scarce limestone. Fossils are common in the sedimentary rocks, with most of indeterminate or probable Middle Jurassic age and at least one early Bajocian collection (Diakow and Webster, 1994).

The Jurassic stratigraphy was intruded by the Late Cretaceous Capoose Lake Batholith (**Unit LKqm**), a 250 km² pluton which extends southwesterly for at least 20 kilometres from the Fawn property. The Hazelton volcanics of the southwestern portion of the Fawn property are thought to be underlain by the Capoose Lake Batholith at a fairly shallow depth. Its main phase consists of light coloured, medium- to coarse-grained, equigranular quartz monzonite, although its composition is locally granodioritic and a dioritic phase cuts northerly through the Fawn 2, 4 and 7 claims. Andrew (1988) reports a biotite K-Ar date of 64.3±2.4 Ma for the batholith. Miagolytic quartz porphyry dykes and plugs cut Hazelton Group sediments on the Buck property, immediately east of the Fawn claims. These were interpreted by Diakow and Webster (1994) to be subvolcanic apophyses projecting from the Capoose Lake Batholith.

Flat-lying to moderately dipping, subaerial volcanics of the Ootsa Lake Group (**Unit EO**) unconformably overlie older Mesozoic rocks. Potassium-argon dating of Ootsa Lake rocks at the Wolf prospect gave an age of 48±2 million years (mid-Eocene). The Ootsa Lake volcanics consist of calc-alkaline andesite to rhyolite. North of the Natalkuz Fault, a northeasterly trending fault which passes twenty kilometres northwest of the Fawn claims, Ootsa Lake volcanics cover an extensive area, with a 750 metre stratigraphic section. South of the fault, the Ootsa Lake Group forms thin isolated cappings on older rocks.

Miocene plateau basalts of the Chilcotin Group (**Unit Cv**) unconformably overlie all other units.

Low grade regional metamorphism and weak deformation are pervasive on the Nechako Plateau. Contact metamorphism is pronounced around intrusives. Tipper (1959) observed that the overall lack of structural features may, in part, be attributed to the abundance of often structureless volcanics in the area. The Hazelton volcanics appear more strongly deformed in comparison to other rock types, with dips of up to 70°. At the Capoose deposit, eight kilometres north of the Fawn property, bedding dips moderately (20-40°) to the southwest, with a synclinal fold axis plunging at 10° to the southeast (Andrew and Godwin, 1987). The Ootsa Lake Group volcanics were deposited in a period of extensional tectonism. Another period of deformation during the Oligocene produced broad open folds in the Ootsa Lake Group volcanics and sediments. The relatively undeformed Chilcotin Group consists of generally flat-lying to gently easterly dipping plateau lavas (Tipper, 1963).

Several styles and ages of mineralization have been documented in the vicinity of the Fawn property (Figure 3). Teck Corp.'s Tommy epithermal gold-silver prospect, 17 kilometres south of the Fawn claims, consists of several north to northeast trending veins and silicified stockwork zones hosted by Naglico Formation quartz-phyric felsic crystal lithic and ash tuffs. The veins consist of milky quartz, chalcedony, sparry calcite, ankerite and adularia, with typical epithermal textures such as druse, colloform banding, cockscomb structures and multiple brecciation/veining episodes. Only trace amounts of sulphides, mainly pyrite, chalcopyrite, sphalerite and galena, are present. The Tommy Vein, which has received the most exploration, hosts a geological resource of 478,000 tonnes grading 8.7 g/tonne Au and 82.3 g/tonne Ag across an average width of four metres (J. Pautler, pers. comm., 1997).

The Wolf epithermal gold-silver prospect, located twenty kilometres west of the Fawn property, is hosted by Eocene Ootsa Lake Group rhyolitic flows, tuffs and subvolcanic intrusives. Repeated low-sulphide silicification, brecciation and stockwork veining have been accompanied by up to 8.49 g/tonne gold and 42.2 g/tonne silver across 7.5 metres in trenching (Cann, 1984). It has been suggested that

the Wolf deposit may have been related to maar (Andrew et al, 1986), collapse caldera (Andrew, 1988) or hot-spring (Andrew, 1988) paleo-environments.

The Capoose silver deposit, located eight kilometres north of the Fawn property, is hosted by Naglico Formation mafic flows, rhyolite tuff, argillite and lithic wacke intruded by Late Cretaceous quartz-garnet rhyolite sills related to the Capoose Lake Batholith. Mineralization consists of pyrite, sphalerite, galena, chalcopyrite and arsenopyrite in disseminations, fracture-fillings and replacing garnets, and is thought to be Late Cretaceous in age (Andrew, 1988). The Capoose deposit contains 28 million tonnes grading 36 g/tonne silver and 0.9 g/tonne gold (Green and Diakow, 1993). The Capoose Lake Batholith itself has been explored for porphyry-style copper-molybdenum mineralization a few kilometres west of the Capoose deposit.

Immediately east of the Fawn property, the Buck 1-4 claims cover a 3,000 metre long zinc-arsenic-lead soil geochemical anomaly overlying Naglico Formation rocks. Proximal (vent facies) felsic volcanics change laterally to distal felsic volcanoclastics and epiclastics along with marine sedimentary and intermediate volcanic lithologies. Stratabound sphalerite-pyrrhotite mineralization, grading up to 4.69% zinc, is present in felsic ash tuffs. The Christmas Cake Showing, with a 45 centimetre chip sample grading 7.38% Zn, 2.25% Pb and 542 g/tonne Au, consists of coarse sphalerite, iron carbonate, galena, minor chalcopyrite and sugary quartz forming a matrix which supports fragments composed entirely of very fine-grained pyrite and by variably altered, angular, felsic lithic clasts (Baknes and Awmack, 1994). A northeast-trending VLF-EM conductor corresponds to a recessive zone of clay alteration with quartz-calcite veining, accompanied by 2-10% pyrite and lesser arsenopyrite and sphalerite. Although this zone returned only low gold and silver values, its similarities to the Fawn's Giver Zone suggest a genetic link (Caulfield, 1996).

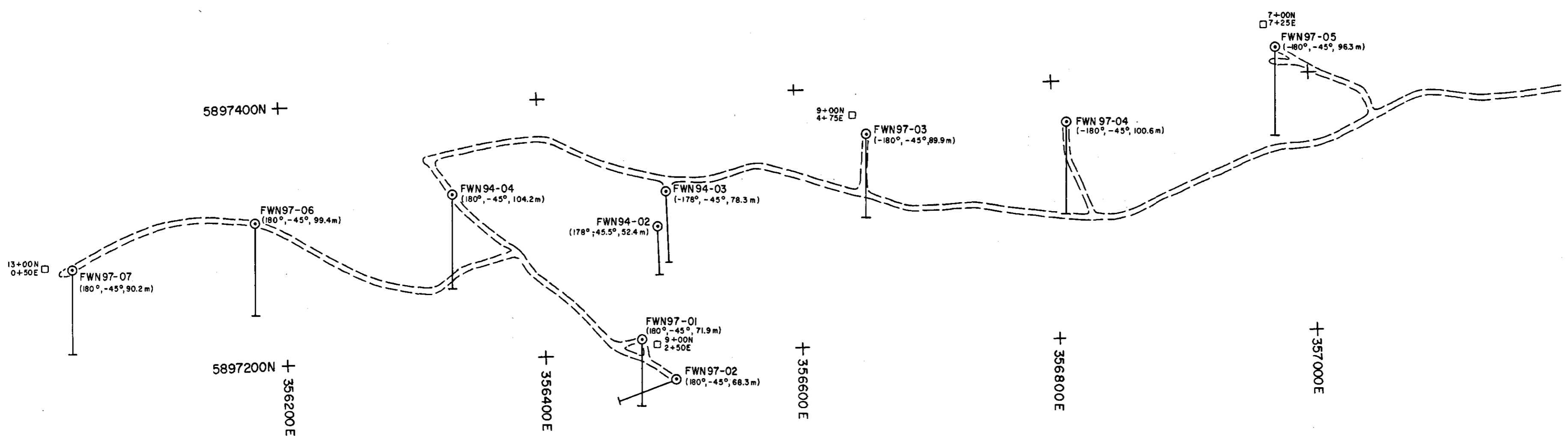
Fifteen kilometres east of the Fawn property, the PEM prospect is underlain by Naglico Formation felsic to intermediate tuffs, lapilli tuffs, breccias and flows, intercalated with argillite, siltstone and sandstone. Disseminated and shear-hosted mineralization occurs in a steeply-dipping, structurally-controlled zone of phyllic and argillic alteration at least 900 metres long, with introduction of 3-4% sphalerite and 1-2% pyrite (Schroeter and Lane, 1994). Zbitnoff (1988) reports drill intersections up to 6.3 metres grading 14.3 g/tonne gold, 27 g/tonne silver and 1.25% zinc. Textural evidence suggests that PEM mineralization may be genetically similar to that of Capoose.

6.0 PROPERTY GEOLOGY

The Fawn property is largely underlain by a sequence of Early to Middle Jurassic Hazelton Group (Naglico Formation) rhyolitic and andesitic volcanics with lesser epiclastic sediments. These have been intruded by a dioritic pluton, thought to form part of the Late Cretaceous Capoose Lake Batholith, and by later felsic dykes which are presumably feeders to the Tertiary Ootsa Lake rhyolites. No geological mapping was carried out on the Fawn property during the 1997 drill program; more detailed descriptions of geology and mineralization can be found in previous reports by Baknes and Awmack (1994a) and Awmack (1991).

7.0 DIAMOND DRILLING

Seven holes were drilled in the vicinity of the gold-bearing 1994 drill intersections on the Giver Zone. Five of these were 150-metre stepouts along the V2 (Giver Zone) VLF-EM conductor from holes FWN94-02, 03 and 04. The other two were designed to test the Giver Splay, a conductive cross-structure where epithermal veining had assayed up to 12.9 g/tonne Au in subcrop. Table 7.0.1 summarizes location, orientation and drilling depths for the 1997 holes. The holes are located in plan on Figures 4 and 5, with vertical cross-sections in Figures 6-12. Drill logs are attached in Appendix C.



CASCADIA INTERNATIONAL VENTURES INC.			
FAWN PROPERTY DRILL PLAN GIVER ZONE MAP 9			
	Date: Oct. 1997	Scale: 1:2500	FIGURE:
	U.T.M. Zone: 10 (NAD-27)	Mining Dist. OMINECA	5
	N.T.S. 93F/3E	State/Prov. B.C.	

Table 7.0.1
Drill Hole Survey Data

Hole Number	Grid Location		Elevation (m)	Azimuth (°)	Dip (°)	Length (m)
	Northing (m)	Easting (m)				
FWN97-01	9+12 N	2+45 E	1525.7	180	-45	71.9
FWN97-02	8+72 N	2+40 E	1528.7	250	-45	71.3
FWN97-03	8+82 N	4+71 E	1478.3	180	-45	89.9
FWN97-04	7+68 N	5+70 E	1457.2	180	-45	100.6
FWN97-05	6+84 N	7+18 E	1425.3	180	-45	96.3
FWN97-06	11+98 N	1+81E	1574.8	180	-45	99.4
FWN97-07	12+87 N	0+62 E	1598.5	180	-45	90.2
						619.6

FWN97-01

Hole FWN97-01 was drilled under the Giver Splay along the same section as 1994 drill holes FWN94-02 and -03. The Giver Splay is a weak VLF-EM conductor which trends 130° from conductor V2 (Giver Zone). In 1991, a sample of silicified, brecciated, pyritic subcrop cut by drusy quartz-arsenopyrite veinlets assayed 12.9 g/tonne Au; it was taken from an old backhoe pit a few metres northwest along the Giver Splay from where hole FWN97-01 cuts it. The Giver Splay had been tested previously by FWN94-04, which was targeted at its junction with the Giver Zone conductor, 180 metres to the northwest; it intersected 2.7 metres grading 2.42 g/tonne Au.

The hole was drilled through a package of andesite lapilli to ash tuff which displays zones of alteration (Figure 6). This alteration zoning progresses from propylitic to intensely sericitized andesite lapilli ash tuff. The propylitic alteration displays variable epidote, chlorite and calcite alteration with a significant magnetite content. With increasing alteration the magnetite appears to be altered to hematite with subsequent loss of magnetism and increased red to brown colour. Mixed hematite-sericite alteration then follows which results in mottled red-brown and pale green coloured lapilli-tuff. With increased alteration the colour changes to pale green-grey, then textural features of the lapilli-tuff are obliterated, commonly with increased disseminated pyrite.

FWN97-01 encountered the Giver Splay between 58.8 and 63.0 metres depth. It is a highly fractured zone of intensely sericitized and variably pyritized lapilli tuff, with local quartz+dolomite veining. A strong fracture foliation trends 10-35° to core axis, indicating that the drill hole is oriented at a low angle to the Giver Splay. This is consistent with its assumed northwesterly trend and steep dip. Assuming a 25° average angle, the true width of the Giver Splay would be about 1.8 metres; the true width of sample 197021 (2.02 g/tonne Au) would be 0.46 metres. Significant samples are summarized in Table 7.0.2.

Table 7.0.2
Significant Intercepts - FWN97-01

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197019	58.80	60.15	1.35	365	---	2.8	928	27	60	20	440
197020	60.15	61.90	1.75	<5	---	0.4	32	8	34	6	178
197021	61.90	63.00	1.10	2000	2.02	6.0	2270	9	84	10	508

FWN97-02

FWN97-02 was directed southwesterly across the same target as FWN97-01, in an effort to cut

the Giver Splay at a less acute angle. Like hole FWN97-01, it was drilled entirely within andesitic lapilli tuff (Figure 7). Strong sericite alteration was observed below the casing from 3.05 to 8.20 metres as well as paralleling a major fault intersected at 48.3 to 52.2 metres and a fault zone from 63.6 to 64.9 metres. The alteration envelopes parallel these fault zones and alteration decreases with distance from the faults. Presumably, the faults correspond to the Giver Splay. Fracturing is generally 50-55° to core axis, corresponding to the Giver Splay's assumed northwesterly trend and steep dip.

Table 7.0.3
Significant Intercepts - FWN97-02

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197024	13.40	14.70	1.30	170	---	1.2	6	28	262	<2	484
197026	50.00	52.20	2.20	130	---	3.8	264	28	44	14	154

FWN97-03

Holes FWN97-03, -04 and -05 were successive 150 metre stepouts to the east along the V2 conductor (Giver Zone) from 1994 holes FWN94-02 and -03. Hole FWN97-03 was collared in a fine-grained andesite flow, entering andesitic lapilli tuff at 9.7 metres (Figure 8). It intersected the Giver Zone from 50.5 - 83.6 metres depth, a true width of about 27 metres exhibiting intense sericite+clay alteration and heavy fracturing at 30-65° to core axis. In the core of the Giver Zone, quartz veining increases and two sections of quartz+dolomite vein breccia were intersected, roughly paralleling the core axis (67.6-74.1m and 77.0-81.5m). Slightly elevated gold and arsenic values are associated with the vein breccia (Table 7.0.3).

Table 7.0.4
Significant Intercepts - FWN97-03

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197042	70.41	74.10	3.69	575	---	2.2	1750	11	68	6	348
197044	77.00	78.90	1.90	345	---	1.4	1530	12	24	8	186
197045	78.90	81.50	2.60	345	---	1.8	1420	14	36	8	224

FWN97-04

Hole FWN 97-04 was collared 150 metres east of FWN97-03 along VLF-EM conductor V2. It displayed narrow intervals of variably hematite, sericite and clay altered andesite lapilli tuff over the upper 56.1 metres (Figure 9). From 56.1 to 90.4 metres the hole cut a zone of very intensely altered, fractured and faulted lapilli to ash tuff and a minor quartz feldspar porphyry dyke. A very strong fault zone from 59.8 to 65.0 metres displayed strongly brecciated lapilli tuff with clay gouge and alteration. The fault/fracture zone appears to continue to 81.5 metres as all rock units are broken and blocky, commonly with clay on fracture/fault slip surfaces. Broken core within the fault zone display fragments of vein quartz as well as <5mm quartz veins cross-cutting the fault breccia. This suggests multiphase quartz episodes and activity along the V2 conductor.

A quartz stringer zone from 56.3 to 57.2 metres consists of vuggy, generally 1-2 mm (to a maximum of 30 mm) grey quartz stringers with traces of arsenopyrite, chalcopyrite and sphalerite. A later dolomite stringering episode does not appear to host any significant sulphide mineralization. Minor quartz+pyrite+arsenopyrite stringers occur in the 57.2 to 59.8 metre interval and rare quartz + sulphide stringering occur from 63.3 to 79.3 metres.

A quartz-feldspar porphyry dyke intersected from 81.5 to 86.0 metres displays strong fracturing with significant alteration paralleling these fractures suggesting later movement on the fault zone.

Table 7.0.5
Significant Intercepts - FWN97-04

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197049	14.30	15.80	1.50	130	---	<.2	14	46	2	12	120
197057	56.30	57.20	0.90	125	---	1.2	282	10	16	8	70
197059	57.60	58.70	1.10	130	---	3.2	370	27	18	12	176
197061	59.40	59.80	0.40	580	---	6.6	3690	14	50	14	266
197065	63.30	65.00	1.70	140	---	1.8	622	11	30	8	184
197067	66.50	67.40	0.90	120	---	1.2	1200	17	18	10	128
197071	74.80	75.90	1.10	275	---	2.2	638	17	24	8	284

FWN97-05

Hole FWN 97-05 was the easternmost stepout along conductor V2, collared another 160 metres east of FWN97-04 in an area of anomalous arsenic soil geochemistry. The upper 50.5 metres cut a variably sericite-hematite altered interval of ash and lapilli tuff (Figure 10). Strong sericite alteration is associated with a fault at 21.9 to 22.4 metres. Minor silicified zones occur in the interval from 20.0 to 21.9 metres above the fault.

Quartz-feldspar porphyry was intersected from 50.5 to 86.6 metres. This interval was subdivided into two units based upon their groundmass colour difference as well as significant fracturing differences. The upper 50.5 to 78.0 metre interval displayed strong fracturing and weak mineralization, with white and grey quartz stringers hosting trace pyrite and arsenopyrite. An extremely rubbly sericite and clay altered zone from 60.5 to 65.9 metres is likely responsible for the V2 conductor.

Gold-bearing mineralization occurs at the footwall contact of the quartz feldspar porphyry. Throughout the mixed porphyry/tuff zone (86.6-89.1 metres) and footwall ash lapilli tuff zone (89.1-91.7 metres), millimetre-scale white and grey quartz stringers host minor amounts of pyrite, arsenopyrite, sphalerite, galena and chalcopyrite. Very fine grained black alteration associated with this zone may be the result of fine-grained sulphides in the wall rock.

Table 7.0.6
Significant Intercepts - FWN97-05

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197092	86.60	87.90	1.30	575	---	4.4	1295	8	10	6	74
197093	87.90	89.10	1.20	210	---	3.0	892	12	12	8	106
197094	89.10	90.60	1.50	1620	1.58	0.2	16	31	62	2	312
197095	90.60	91.70	1.10	300	---	0.6	62	47	56	6	368

FWN97-06

Holes FWN97-06 and FWN97-07 were stepouts west along conductor V2 from 1994 drill hole FWN94-04, which intersected 2.7 metres assaying 2.4 g/tonne Au. Hole FWN97-06 was collared 150 metres west of FWN94-04 in the vicinity of several epithermal float boulders sampled in 1991. These samples, known as the Givermore Zone, graded up to 3.4 g/tonne Au and 92 g/tonne Ag.

FWN97-06 encountered dominantly andesite flows with minor andesite lapilli tuffs (Figure 11). Alteration throughout the interval from 43.3 to 71.1 metres is dominantly sericite, clay and pyrite alteration with the most intense alteration paralleling a fault zone from 59.74 to 69.5 metres. Pyrite mineralization is dominant throughout the altered interval and likely represents pyritization of the magnetite (magnetite→hematite→pyrite). Within the fault zone is an interval of silicified breccia with dark grey chalcedonic quartz and minor white quartz from 59.74 to 61.4 metres. On either side of this silicified fault breccia occurs strongly altered and variably quartz and dolomite stringered andesite. Chalcedonic quartz stringers host varying amounts of pyrite, arsenopyrite and sphalerite but sulphides are generally present as trace amounts. These stringers occur sporadically throughout the altered zone, but are also concentrated as a stockwork zone from 53.1 to 55.3 metres. The upper, silicified, part of the fault and the overlying alteration averages 1.08 g/tonne Au and 23 g/tonne Ag along 10.2 metres of core (51.2 - 61.4 metres).

A strong fault breccia with clay gouge intersected from 61.4 to 62.5 metres, and an extremely broken and faulted clay and sericite altered zone from 62.5 to 69.5 metres are thought to be the source of the V2 conductor.

Table 7.0.7
Significant Intercepts - FWN97-06

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197100	51.20	53.10	1.90	680	---	7.2	2350	45	34	24	258
197101	53.10	54.00	0.90	1680	1.61	42.4	1835	29	36	60	288
197102	54.00	55.35	1.35	1365	1.23	56.8	2690	24	56	64	260
197103	55.35	57.00	1.65	2340	2.33	11.0	5370	52	54	22	428
197104	57.00	59.74	2.74	<5	---	3.0	232	78	26	20	244
197105	59.74	61.40	1.66	1700	1.68	50.0	2640	34	202	34	656
Avg.	51.20	61.40	10.20	1080	1.08	23.3	2316	49	66	33	349
197109	69.50	71.00	1.50	250	---	1.8	1385	22	28	8	266
197112	81.40	83.90	2.50	160	---	1.0	332	35	14	6	124

FWN97-07

Hole FWN 97-07 was the westernmost 1997 drill hole, collared 150 metres west of FWN97-06 on the V2 conductor, in an area with no soil geochemical anomalies and no surface epithermal float. It encountered a package of volcanic flows and fragmental volcanics which displayed alteration parallel to two strong faults (Figure 12). All quartz and dolomite stringering occur within the faults or alteration zone.

Pyrite mineralization occurs disseminated throughout the extremely altered zones, and within minor quartz stringers. Within the fault zones clay gouge and breccia appear to contain disseminated pyrite throughout. Fragments of quartz-feldspar porphyry within the fault breccia suggest the presence of a dyke which may have occupied the zone, but was reduced to rubble by faulting. The quartz feldspar porphyry fragments may also have been dragged into the zone from elsewhere.

Stringering throughout the hole appeared to be substantially less than previous holes along the V2 conductor. The reduction in stringering may be the result of the strong faulting which may have post-dated the quartz veining event. The strong faults from 43.7 to 50.4 metres and 63.6 to 67.0 metres may both represent the V2 conductor although the 43.7-50.4 metre fault displayed a larger interval with strong clay gouge.

Table 7.0.8
Significant Intercepts - FWN97-07

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
197115	29.30	30.60	1.30	140	---	0.6	120	21	12	6	78
197127	65.40	67.00	1.60	165	---	0.8	1890	17	20	6	90
197128	78.10	78.90	0.80	25	---	13.4	32	4060	490	4	1610

8.0 DISCUSSION

The 1997 diamond drilling program focused on the Giver Zone, an east-west trending zone of faulting and sericite-clay alteration with local quartz stringer stockworks and chalcedonic breccias. It is marked on surface by sporadic arsenic, lead and zinc soil geochemistry and a strong VLF-EM conductor (V2). Eight holes tested the Giver Zone in 1994 and 1997, on seven sections along 900 metres of its strike length. Each of these holes produced wide altered intervals and samples exceeding 100 ppb Au; five returned intersections of 1-3 g/tonne Au, including hole FWN97-06 with 10.2 metres grading 1.08 g/tonne Au.

Two 1997 holes tested the Giver Splay, a weak, northwesterly-trending VLF-EM cross-structure to the Giver Zone. Epithermal veining which subcrops between the 1997 holes and the Giver Zone had yielded assays up 12.9 g/tonne Au in 1991. However, the drilling shows the Giver Splay to be a relatively narrow zone of faulting and mineralization; the best core sample assayed 2.02 g/tonne Au across a few tens of centimetres true width.

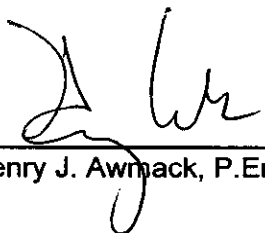
Quartz-feldspar porphyry dykes were encountered in several Giver Zone holes, locally altered and mineralized. They both follow fault zones and are locally fractured or brecciated by later faulting. They may be related to the Late Cretaceous Capoose Lake Batholith or may be feeders to Eocene Ootsa Lake Group volcanics. Their close spatial association with alteration and epithermal mineralization in drill holes and on surface suggests a genetic link as well, implying that Fawn epithermal mineralization is Cretaceous or Tertiary in age. Repeated fault, hydrothermal and igneous activity along the Giver Zone is suggested by the locally brecciated dyking and by silicified fault breccias cut by later veining and faulting..

The alteration along the Giver Zone (V2) is indicative of a strong and extensive epithermal mineralizing system. Despite widespread 'sniffs' of gold, no potentially economic gold mineralization has yet been discovered along the 900 metres of the V2 conductor which has been drill tested. However, less than 15% of the prospective strike length of the known conductors has been drilled. Conductor V2 itself has been defined over an additional 1,000 metre strike length and remains open to the east and west. Three other east-west VLF-EM conductors with associated soil geochemical anomalies were also defined by the 1991 survey, aggregating 4,600 metres of strike length and each remaining open in one or two directions; none of these has been tested by drilling. In addition, it appears likely that further parallel structures will be found to the south of the 1991 geophysical survey, including the Malaput Showing, an east-west silicified zone four kilometres south of the V2 conductor. Epithermal systems are characterised by strong vertical controls on mineralization. It may turn out that the drilled portion of V2 is too low (or too high) in the epithermal system and one of the other structures is exposed at a better level.

The Tommy prospect, located 17 kilometres south of the Fawn property, consists of epithermal quartz veins in Hazelton Group (Naglico Formation) quartz-phyric rhyolite tuffs. Teck Corp. has developed a reserve of 478,000 tonnes grading 8.7 g/tonne Au over a width of four metres at Tommy in a geological setting which is very similar to the Fawn property's. Not only does this bode well for the

possibility of discovering significant gold mineralization on the Fawn claims, but it suggests a possible rheological control on mineralization. At Tommy, the rhyolite host forms brittle fractures, along which the quartz veins are emplaced. In the Giver Zone, the less competent andesitic lapilli tuffs do not form discrete fractures, but rather wide zones of faulting, alteration and quartz stockworks, with more dispersed gold mineralization. It would be worthwhile extending the Fawn geophysical grid eastward over the Naglico Formation quartz-phyric rhyolite tuffs. If the known fault structures, and their associated hydrothermal fluids, extended over into the rhyolite, they may also be characterized by narrower but more focused, discrete quartz veins.

Respectfully submitted,
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Vancouver, British Columbia
October, 1997

APPENDIX A

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- Zbitnoff, G.W. (1988): Diamond Drilling Report on PEM Claim; British Columbia Energy and Minerals Division Assessment Report #17,032.

APPENDIX B

STATEMENT OF EXPENDITURES

STATEMENT OF EXPENDITURES

Fawn 1 Claim

March 24 - April 12, 1997

PROFESSIONAL FEES AND WAGES

Stewart Harris, Project Geologist			
30.188 days @ \$425/day	12,829.90		
Jim Lehtinen, P. Geo			
23.5 days @ \$425/day	9,987.50		
Mark Baknes, P. Geo.			
3.25 days @ \$425/day	1,381.25		
Warren Cole, Field Assistant			
20.5 days @ \$225/day	4,612.50		
Clerical			
0.25 hours @ \$25/hour	6.25	\$	28,817.40

EXPENSES

Accommodation	\$ 12,048.52		
Cat	15,263.48		
Automobile Fuel	983.59		
Bulk Fuel	3,830.87		
Water Truck	14,595.00		
Camp Supplies	162.53		
Chemical Analyses	2,166.86		
Courier	48.65		
Freight	775.00		
Airfare	4,361.70		
Materials and Supplies	3,219.92		
Meals	146.16		
Taxis and Airporters	48.77		
Printing and Reproductions	102.47		
Tolls and Airport Taxes	9.35		
Automotive Expenses	53.86		
Telephone Distance Charges	352.66		
Truck Rental (Non-Equity)	3,795.39	\$	61,964.78

SUB-CONTRACTS

Falcon Drilling Ltd:			
Mob/Demob	9,165.00		
Footage	38,375.50		
Materials	467.09		
Standby/Moves/Travel	10,290.00		
Other Drilling Expenses	1,960.00	\$	60,257.59

EQUIPMENT RENTALS

4x4 Equity Crewcab			
20 days @ \$80/manday	\$ 1,600.00		
Generator, 1kVA			
5 days @ \$10/day	50.00		
Truck Radio			
19 days @ \$5/day	95.00		
Chainsaw			

17 days @ \$15/day	255.00		
Handheld Radios			
12 days @ \$5/day	<u>60.00</u>	\$	2,060.00

REPORT (estimated)

Drafting	\$ 2,500.00		
Printing and Reproductions	900.00		
Time	<u>6,600.00</u>	\$	<u>10,000.00</u>

SUBTOTAL \$ 163,099.77

PROJECT SUPERVISION CHARGE

12% on first \$100,000	\$	12,000.00
10% on remaining \$63,099.77	\$	<u>6,309.98</u>

SUBTOTAL \$ 181,409.75

GST

7.0 % on subtotal (including project supervision charges)	\$	<u>12,698.68</u>
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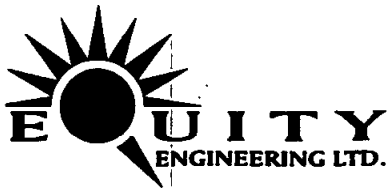
TOTAL \$ 194,108.43

APPENDIX C

DIAMOND DRILL LOGS

MINERALS AND ALTERATION TYPES

AS	arsenopyrite	BI	biotite	CA	calcite
CL	chlorite	CP	chalcopyrite	CY	clay
EP	epidote	GE	goethite	GL	galena
HE	hematite	JA	jarosite	MC	malachite
MG	magnetite	MN	Mn-oxides	MS	sericite
PO	pyrrhotite	PY	pyrite	QZ	quartz
SI	silica	SP	sphalerite	TT	tetrahedrite

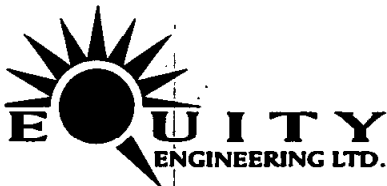


DRILL LOG

PROJECT <i>Fawn</i>				COLLAR ELEVATION <i>1525.7</i>					
HOLE <i>FWN 97-01</i>				AZIMUTH <i>180°</i>					
LOCATION <i>9+12 N 2+45 E</i>				DIP <i>-45°</i>					
LOGGED BY <i>J. Lehtinen</i>				LENGTH <i>71.9</i>					
DRILLED BY <i>Falcon Drilling</i>				HORIZONTAL PROJECTION					
ASSAYED BY				VERTICAL PROJECTION					
CORE SIZE <i>BTW</i>				ALTERATION SCALE absent slight moderate intense					
DATE STARTED <i>March 30/97</i>		DATE COMPLETED <i>March 31/97</i>							
DIP TESTS BY <i>Acid</i>				SULPHIDE SCALE traces only < 1% 1% - 3% 3% - 10% > 10%					
DEPTH	DIP	AZIM	DEPTH					DIP	AZIM
<i>71.9</i>	<i>-45</i>	<i>/</i>							
OBJECTIVE									
SUMMARY LOG									
<i>Target:</i>									
<i>0-3.05 Casing</i>									
<i>3.05-11.3 Lapilli/Ash Tuff</i>									
<i>11.3-21.8 Hematite Altered Lapilli Tuff</i>									
<i>21.8-22.9 Andesite Lapilli Tuff</i>									
<i>22.9-33.0 Variably MS-HE altered Lapilli Tuff</i>									
<i>- Fracture fol'n near base of interval</i>									
<i>33.0-40.3 Andesite Lapilli Tuff</i>									
<i>40.3-52.3 Variably MS-HE altered Fin-Lapilli Tuff</i>									
<i>52.3-53.7 - Ash Lapilli Tuff</i>									
<i>53.7-57.1 - Variable HE altered Med-Lapilli Tuff</i>									
<i>57.1-63.6 MS-FPY altered Lapilli Tuff</i>									
<i>63.6-65.5 - HE altered Lapilli Tuff</i>									
<i>65.5-71.0 - Andesite Lapilli Tuff</i>									

PAGE 1		OF 8		PROJECT FAWP		HOLE FWN 97-01					
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION				FRACTURE INTENSITY	
						Sericite	Hematite	Chy.	Dolomite		
0.0											
-1					0-3.05 Casing						
-2					3.05 - Pale green-grey to med green						
-3					11.3 Grey Lapilli Ash Tuff						
-4					- Variably altered throughout giving mottled appearance.						12
-5					- Lapilli fragments commonly pale green, also rare dark grey						
-6	66	13			- Fragments up to 4cm in diam						
-7					- Ash matrix. Rare laths of feldspar crystals up to 1.5mm.						
-8					- Strong Oxidation along fractures						
-9					Alteration - Variably sericite affct with trace hematite alteration throughout. Patchy hematite near base of interval.						
-10											50
-11					11.3 - Hematite Altered Lapilli Tuff						75
-12					21.8 - Variably altered, dominantly maroon matrix with pale green-grey fragments.						10
-13					Fragments variable composition from ash tuff, crystal tuff, porphyritic and amygdaloidal.						
-14					- Minor < 3mm dolomite stringers at 40-55° TCA + along C.A.						
-15	105	59			- Minor Clay along weak slips						15
-16					- Zones of moderate sericite alteration from 17.0 to 18.2m + 19.8 to 20.8						40
-17					- 21.2 - Minor erratic chalcocite quartz.						20
-18	101	52									15
-19											30
-20					- Very fine fracture fol'n. (crack)						10

DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					SI / FP	FRACTURE INTENSITY
						MS	HE	CY	DO	CA		
60					58.8 - Med. Gray Sarcite altered and							
61	89	82			60.15 Pyritized Lapilli Tuff.							10
62	88	36			- red grey with lighter green - grey lapilli tuff. Less PY = lighter color - Fragments remain distinct.							15
63	86	39			with chng MS alt'n + pyritization - Strong fracture fol'n at 10-30°							10
64					TCA. Fracture fall commonly of QZ-PY-vuggy Patchy creamy							
65	82	76			DO. PY. disseminated throughout Faulted (?) rubble basal contact							
66					60.15 - Pale Green Sarcite Altered Lapilli Tuff.							5
67	98	90			61.9 - Light grey-green with minor HE altered zones. Angular fragments up to 3 cm. Weak fracturing							
68					Patchy pyrite (3%) near base.							
69					61.9 - med Gray Sarcite altered and							
70					63.0 Pyritized Lapilli Tuff.							
71	111	87			(QZ per 58.8-60.15) - Strong fracture fol'n + increased vuggy quartz stringers - 15-35 TCA							
71.9	EOH				- Quartz + pyrite stringers display growth texture (very weak cockscomb) - Vein bx with QZ, PY, creamy DO.							
					63.0 - Red-brown Hematite Altered 65.5 Lapilli Tuff							
					- med to dark red brown with Lapilli fragments. up to 4 cm.							
					- Transitional contact from above with MS+PY → MS → HE±MS							
					- Minor DO bx zone 64.8-65.1 m							
					65.5 - Dark Green Andesite Lapilli 71.9 Tuff							
					- Dark-med green - Lapilli fragments up to 10 cm.							
					- Minor fragments with Muscovite? phases							
					- Strong calcite + med. epidote alteration. Cat Epidote Amygdalites							
					- Minor calcite + calcite-hematite stringers							
					EOH 71.9 m							



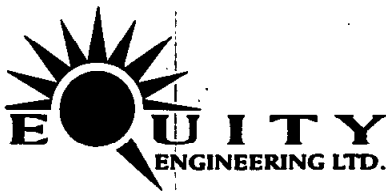
DRILL LOG

PROJECT FAWN		WKM 97-06		COLLAR ELEVATION 1528.7			
HOLE FWN 97-02				AZIMUTH 250°			
LOCATION 8+72 N 2+40 E				DIP -45°			
LOGGED BY J. Lehtinen				LENGTH 68.3			
DRILLED BY Falcon Drilling				HORIZONTAL PROJECTION			
ASSAYED BY Chemex				VERTICAL PROJECTION			
CORE SIZE BQTW				ALTERATION SCALE 			
DATE STARTED march 31/97	DATE COMPLETED April 1/97						
DIP TESTS BY				SULPHIDE SCALE 			
DEPTH	DIP	AZIM	DEPTH			DIP	AZIM
62.5	-47°	/					
OBJECTIVE							
SUMMARY LOG							
Target:							
0.0 - 3.05 - Casing							
3.05 - 8.20 - Strongly oxidized MS-HE altered Ash-Crystal Tuff							
8.20 - 14.4 - Andesite Lapilli Tuff							
14.4 - 20.85 - Variable HE-MS altered Lapilli Tuff							
20.85 - 48.3 - Andesite Lapilli Tuff ± fish Tuff. minor HE altered sections - Strong oxidation + fracturing 28.95 - 33.1							
48.3 - 52.2 Strong Fault in CY+MS altered Lapilli Tuff							
52.2 - 63.6 - MS + CY altered Lapilli Tuff							
63.6 - 64.9 - Strong Fault + Breccia Zone							
64.9 - 68.3 - Andesite Lapilli Tuff							

DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY
						MS	HE	CA	EP		
0					0.0 - Casing - Sericite alt'd						
1					3.05 Lapilli tuff in basal 40cm						
2	CASING										
3					3.05 Strongly oxidized Sericite-Hematite altered Ash-Crystal Tuff						7/6
4	96	8		X	- Extremely fractured with Limonite (Fe stained) fracture surfaces + pervasively throughout core - Rare unoxidized inclusions with MS + HE altered ash-crystal tuff.						7/6
5				X							
6	89	31		X	- Trace Pyrite - Unoxidized med-dk green Strong CA altered andesite (tuff?) from 5.8 - 6.4 m						5
7				X							7/6
8			str MS CA	75 100 100							8
9	98	73		X	8.2 - Med-dark green Andesite 14.4 Lapilli Tuff						
10					- med to dark green with minor hematite altered zones						
11			str CA	75	- Lapilli fragments less distinct than altered units.						
12	103	80		X	- Fragments with diffuse boundaries + commonly porphyritic						
13				X							
14			str MS CA QE	100 100 100 100	- Fine < 2mm CA stringers common + HE						
15	99	61		X	14.4 - Light Green-Grey to Red Brown Lapilli Tuff						
16			str MS CA	55 100 100	- Variably coloured due to varying degree of HE and MS alteration						
17				X							
18	103	69		X	Lapilli fragments angular to sub-rounded. Commonly fine grained to aphanitic + porphyritic weak DO stringering.						
19				X	Fragments up to 8 cm. HE along micro-fractures						
20				X							

DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY
						MS	HE	CA	EP	FR	
20			FE, CA		20.85- Med- Dk green Andesite	/	/	/	/	/	5
21	97	86			28.95 Lapilli Tuff.	/	/	/	/	/	
22					- Med. green with locally HE altered patches. Lapilli fragments up to 5cm. Variable composition of fragments, sub-angular to sub-rounded	/	/	/	/	/	
23			CA, HE		- Thin calcite stringers + minor HE stringers	/	/	/	/	/	
24	101	100				/	/	/	/	/	
25						/	/	/	/	/	
26						/	/	/	/	/	
27						/	/	/	/	/	
28	99	96	CA		28.95- Med- Light Green Lapilli - Ash	/	/	/	/	/	
29					33.1 Tuff.	/	/	/	/	/	
30			FR		- med-light green to extremely iron oxide stained, strongly fractured. Very fine "crack" texture @ 30.5m	/	/	/	/	/	
31	85	46				/	/	/	/	/	30
32			DO		33.1- med- Dk Green Andesite	/	/	/	/	/	
33					48.3 Lapilli Tuff.	/	/	/	/	/	
34	96	72			med-dk green. Lapilli fragments less distinct, large blocks? up to 15cm	/	/	/	/	/	8
35					- Porphyritic fragments.	/	/	/	/	/	
36					- Possibly flow + flow bx in part.	/	/	/	/	/	
37					- Weak calcite + HE. + DO.	/	/	/	/	/	
38	102	90			- Very Strong HE alteration @ base of interval.	/	/	/	/	/	
39	90	70	Stgs CA HE			/	/	/	/	/	
40						/	/	/	/	/	

PAGE 7		OF 8		PROJECT FAWN		HOLE FWN 97-02						
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY	
						MS	HE	CA	CY			
60					63.6- FAULT							
61	83	12			64.9 Gorge + Breccia Zone							90
62					of Sericite and clay in part							740
					weakly cemented with calcite							
					weak foliation @ 50° TCM							
63	8	0										
64	63	0										
	71	39										>50
65												
66	100	48			64.9- med-dark grey + green-grey							
					68.3- Andesite, Lapilli Tuff							10
67	80	46			Grey to green-grey, with							15
					lighter altered zones.							
68					Lapilli fragments indistinct							
					to rare → possibly flow?							
68.3					At contact with fault =							
E.O.H.					50° TCM foliation of fine							
					DO + PY stringering							
					Fracturing commonly 40-60°							
					TCM							
					68.3 E.O.H.							



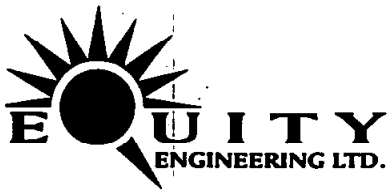
DRILL LOG

PROJECT FAWN			COLLAR ELEVATION 1478.3					
HOLE FWN 97-03			AZIMUTH 180°					
LOCATION 8+02 N 4+71 E			DIP -45°					
LOGGED BY J. LEHTINEN			LENGTH 89.9m					
DRILLED BY FALCON DRILLING			HORIZONTAL PROJECTION					
ASSAYED BY CHEMEX LABS			VERTICAL PROJECTION					
CORE SIZE BATW			ALTERATION SCALE 					
DATE STARTED April 2/97		DATE COMPLETED April 3/97						
DIP TESTS BY ACID			SULPHIDE SCALE 					
DEPTH	DIP	AZIM				DEPTH	DIP	AZIM
89.9	-44	—						
OBJECTIVE								
SUMMARY LOG								
Target:								
0-1.52 Casing								
1.52-9.7 Andesite Flow								
9.7-41.4 Andesite Lapilli Tuff								
41.4-47.8 MS I HE altered Lapilli Tuff								
47.8-50.5 Andesite Lapilli Tuff (Flow?)								
50.5-56.0 MS altered Lapilli Tuff								
56.0-63.15 MS I HE altered Lapilli Tuff								
63.15-67.6 MS altered Lapilli Tuff								
67.6-74.1 MS altered Lapilli Tuff (fault parallel to C.R.)								
74.1-77.0 MS altered Lapilli Tuff								
77.0-81.5 MS altered Lapilli Tuff. QV. Breccia parallel to C.R.								
81.5-83.6 MS altered Lapilli Tuff								
83.6-84.9 MS altered Lapilli Tuff								
84.9-89.91 Andesite Lapilli Tuff								

DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION				FRACTURE INTENSITY	
						MS	HE	CA	CY		EP
40											
41	111	73			41.4 - Light Gray-Green to Red-Brown 42.8 - Sericite altered Lapilli Tuff.						10
42			Py GZ Slip	30°	- Dominantly pale green with patchy Red brown coloring.						
43	104	83	DO	30°	- Lapilli fragments up to seen						
44			DO	30°	- Increasing fracturing and Dolomite stringering - Minor Quartz.						
45	104	74									
46											
47			DO	30°							
48					47.8 - Andesite Lapilli Tuff. 50.5 - Possibly flow.						
49	102	0	CA	20°	- Minor indistinct fragments						
50			SH DO HE	30°	- Dark green-gray - Increased dolomite stringering and alteration @ base						
51					- Strongly magnetic where unaltered						
52	87	48	Fr fol n	40°	50.5 - Light Green, Strongly Broken 56.0 - Sericite and clay altered Lapilli Tuff.						7/10
53			DO	30°							
54			DO	30°	- Pale green with minor gray (pyritic) & red-brown (HE) zones - Shandy DO stringered as fracture fill - erratic stringers						
55	101	54	DO	35°							
56			Slip	60°	56.0 - Sericite & hematite altered Lapilli tuff.						
57			DO	70°	63.15 - Increased HE alt'n. - Po-pyritic zones = x tal tuff.						
58	101	69	DO	60°							
59											
60			DO	60°							

PAGE 7 OF 9		PROJECT FAWW		HOLE FWN 97-03							
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION				FRACTURE INTENSITY	
						MS	HE	CA	CY		
60											
61	99	48			63.15- Sericite Altered Lapilli Tuff						
62					67.6 - Pale Green + with -hematitic mottled colour.						
63					- Entire unit is strongly fractured with intense DO stringering, all < .3mm.						
64	100	71			- Minor quartz stringering						
65											
66											
67	99	44			67.6- Sericite Altered Lapilli Tuff						
68					74.1 - Entire interval follows a fault which hosts light to dark grey brecciated vein quartz which has been cemented with ① Grey quartz and ② a later stage of creamy white dolomite.						
69	105	74			Pyrite appears related to ① the cementing quartz.						
70					- Clay along fracture surfaces						
71											
72	59	30									
73											
74					74.1 - Sericite Altered Lapilli Tuff						
75					77.0 - Very Strong MS alteration of Lap. Tuff						
76	129	85			- Med. → light grey-green						
77					- Relatively unfractured with little stringering. Fracturing increased from 76.2 - 77.0m						
78	121	0			77.0 - Sericite Altered Lapilli Tuff with						
79	72	26			81.5 Quartz vein Breccia along Corec Axis.						
80					- Core axis follows fracturing + Quartz vein BX						
					- Vein BX from 77.6 - 79.7m						
					Vein BX fragments med-dk grey. Angular to sub-rounded. - Silica cemented then DO stringered & cemented.						

DEPTH (M)	% CORE REC	% ROD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY
						MS	HE	CA	CY	EX	
80											
81	123	0			81.5 - Sericite Altered Lapilli + Ash Tuff						750
82					83.6 - Dominantly pale green with patches up to 10 cm of HE.						10
83	94	63	DO	150	- Angular fragments up to 1 cm - Zones of ash tuff :-						
84					83.6 - Hornblende Altered Lapilli Tuff						
85			DO, HE	120	84.9 - Fragmental + Amygdaloidal (Frags)						
86	103	51	Slip	350	- Clay altered zone at base paralleling 1 cm dolomite stringer @ ~5° TCA						
87					84.9 - Andesite Lapilli Tuff						
88	100	66	DO	175	89.91 - Dominantly Propylitic altered Lapilli Tuff						
89			DO	46	- Minor HE altered zones - Fragments up to 12 cm - Minor ~ 5mm QZ or QZ-DO stringers						
89.91					- Moderately magnetic						
					E.O.H 89.91						



DRILL LOG

PROJECT FAWN				COLLAR ELEVATION 1457.2					
HOLE FWN 97-04				AZIMUTH 180°					
LOCATION 7+68N S+70E				DIP -45°					
LOGGED BY J. LEHTINEN				LENGTH 100.6m					
DRILLED BY FALCON DRILLING				HORIZONTAL PROJECTION					
ASSAYED BY CHEMEX LABS				VERTICAL PROJECTION					
CORE SIZE BQTW				ALTERATION SCALE absent slight moderate intense					
DATE STARTED April 4/97		DATE COMPLETED April 5/97		SULPHIDE SCALE traces only < 1% 1% - 3% 3% - 10% > 10%					
DIP TESTS BY Acid → Faulty Test									
DEPTH	DIP	AZIM	DEPTH					DIP	AZIM
OBJECTIVE									
SUMMARY LOG									
Target:				55.4-56.1 Andesite Lapilli Tuff					
0-3.05 Casing				*56.1-59.8 - Strongly fractured/Stringerised Zone with QZ stringers PY, AS, & CP					
3.05-4.57 HE+MS altered Lapilli Tuff				59.8-65.0 - Fault Zone in MS, CY + Brecciated Lapilli Tuff					
4.57-5.79 MS altered Lapilli Tuff				65.0-81.5 MS altered Lapilli Tuff - Blocky & strongly fractured					
5.79-10.1 HE altered Lapilli Tuff				81.5-86.0 Quartz Feldspar Porphyry Dyke					
10.1-16.76 MS altered Lapilli Tuff				86.0-90.0 ms+CY altered Ash Tuff					
16.76-18.7 HE altered Crystal Lapilli Tuff				90.4-91.1 Andesite Ash Lapilli Tuff					
18.7-19.81 - Andesite Flow				91.1-94.1 Andesite Flow					
19.81-24.90 - HE altered Andesite Flow & Ash Lapilli Tuff				94.1-98.0 Andesite - Lapilli Ash Tuff					
24.90-28.30 - Andesite Flow Breccia									
28.30-31.8 HE altered Ash/Lapilli Tuff									
31.8-33.4 MS/CY altered Lapilli Tuff									
33.4-40.7 MS+PY altered Lapilli Tuff (QZ stringers 37.9-40.2)									
40.7-43.9 - HE altered Lapilli Tuff									
43.9-52.50 MS+CY altered Lapilli - Ash Tuff									
52.50-55.4 MS-HE altered Lapilli Tuff									

PAGE		OF 13		PROJECT		FAWN		HOLE		FWN 97-04		
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION		ALTERATION					FRACTURE INTENSITY
							MS	HE	CA	CY	EP	
0					0-	Casing						
1					3.05							
2					2.05-	Strongly Oxidized HE + MS altered						
3					4.57	Lapilli Tuff						
4	89	14				- Extremely iron stained & strongly fractured.						
5					4.57-	Sericite Altered Lapilli tuff						
6	89	38			5.79	mod grey matrix, light grey-beige fragments up to 5cm. MS + CY Altered. Tr diss. PY						
7	97	68			5.79-	Hematite altered Lapilli Tuff						
8					10.1	- Variable alteration decreasing at the interval boundaries. Strong HE = dark red-brown. Fragments up to 5cm, angular to rounded.						
9	97	69				Fragment composition from optically med. grained						
10					90	0						
11					84	25						
12					16.76	- Med Grey matrix with light creamy-green fragments						
13					116	80						
14					11.45-11.60m	@ 20 TCA. Tr. GWT SP. - Weakly Silicified.						
15	101	81			16-76-	Hematite altered Crystal-Lapilli Tuff						
16					18.7	- Variably HE altered with zones of MS alteration.						
17					105	47						
18					18.7-	Dark Green Andesite Flow.						
19					19.81	- Porphyritic alt'd flow with minor flow lx frags. - Corroded frag. boundaries. Porphyritic frags. - weakly magnetic						

MINERALIZATION DESCRIPTION	TOTAL % SULPHIDE	SAMPLES			SAMPLE NUMBER	ASSAYS									
		FROM	TO	WIDTH											
20															
21															
22															
23															
24															
25															
26															
27															
28															
29															
30															
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32															
33															
34															
35															
36															
37															
38															
39															
40															

* 34.5m - Sample for thin section (QZ stringer + alteration banding)

32 - Banded & diss. pyrite strongly pyritized with PY surrounding
 33 21mm ash fragments + also Py as fine 21mm stringers. The pyrite alteration appears with pale orange-brown mineral which may be sericite or gypsum. (scratched with thumbnail - with difficulty)
 34 or some mixture of both - Banding appears to be
 35 Pyrite alteration - Minor QZ stringers

36 - 32.1 - 39.2 - As per 1970S1

37 Also includes vuggy QZ-PY stringers 38.7m → 35.7m
 38 38.85 → 75.7m, 39.2m to 75.7m
 39 - Strongly fractured MS/CY altered

40 Zoned - Weak DO. Stringers Trace Diss. PY

Handwritten notes in the 'TOTAL % SULPHIDE' column, including a large 'X' and some illegible scribbles.

33.4 35.1 1.7 1970S0
 37.1 39.2 2.1 1970S1
 39.2 40.7 1.5 1970S2

PAGE 5 OF 13		PROJECT FAWN		HOLE FWN 97-04								
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY	
						MS	HE	CA	CY	FP		
40.7					Hematite Altered Lapilli Tuff	/			/			5
43.9					- Mottled red-brown to pale green color due to Hematite content.	/			/			5
43	67	66			- Lapilli fragments commonly remaining pale color.	/			/			5
44					- Minor QZ stringer (2cm) 95° TCN with 10% basal PY @ 41.5m	/			/			5
44					- Minor Pyrite paralleling <1mm fractures	/			/			5
45					43.9 - Sericite and Clay altered	/			/			5
46	66	47	Red	65	57-50 Lapilli - Ash Tuff	/			/			5
47			DO	4cm	- Overall light green to grey green with minor zones of HE alteration and pyrite alteration	/			/			5
48					- mixed Ash & lapilli tuff beds	/			/			5
49	101	70	QZ	2cm	- @ 45.8m -> bed = 65° TCN	/			/			5
49			slip		strongly Pyritized zone over 35cm marginal to QZ stringer @ 48.8m	/			/			5
50						/			/			5
51	96	42	QZ	2cm		/			/			5
52			DO	2cm		/			/			5
53					52-50 Hematite/Sericite altered Lapilli Tuff.	/			/			5
54			QZ	2cm	55.4 Patchy red-brown to light green. Fragments are indistinct due to alteration surrounding significant stringering and fracturing. Call < 1cm	/			/			10
55	94	51	DO	2cm		/			/			5
56					55.4 - Andesite Lapilli Tuff	/			/			5
57					56.1 - Dark green propylitic Alt'n	/			/			5
57					- Minor hematitic fractures	/			/			5
58	160	105			* 56.1 - Strongly Fractured Zone	/			/			5
59					59.8 - Fractured and stringered with ~ 50% Grey Quartz and Quartz breccia.	/			/			5
60						/			/			5
* See detail log next page for mineralized zone												

	MINERALIZATION DESCRIPTION	TOTAL SULPHIDE	SAMPLES			SAMPLE NUMBER	ASSAYS								
			FROM	TO	WIDTH ^{RE}										
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42															
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46 - Fracture paralleling C.A. + QZ
47 Stringer @ 30° TCA Tr. PY

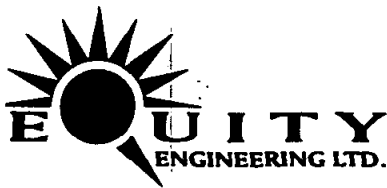
48 - Pyrite alteration margin surrounding
49 3cm fault slip @ 30° with vfg. PY

50 - DO stringers + CY gauge Alt
Zone. Trace PY

52 - PY alteration of sericite altered,
mod. fractured + DO stringered
53 Tuff. Irregular veining, minor
QZ, Vuggy fractures
54 (Alteration similar to sample 197051)

PAGE 7		OF 13		PROJECT FAWN		HOLE FWN 97-04					
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION				FRACTURE INTENSITY	
						MS	HE	CA	CY		
55					Detail of mineralized zone						
55	7	51			56.1 - Sericite and Clay alt'd Lapilli Tuff						
56					56.3 - Pale green, weak DO filled fractures						
56					56.3 - Quartz Stringer Zone						
57			DO		57.2 Strongly broken Lapilli Tuff						
57			QZ		with quartz stringering & minor DO stringers						
57					QZ @ 60 TCA @ 56.6						
57			QZ		15° - 57.0m vuggy AS, PY						
57			QZ		25° 57.15m vuggy SP, PY						
57			QZ		50° - 57.35m vuggy SP, PY						
57			QZ		Strong basal contact @ 50 TCA						
58	160	105			57.2 - Sericite alt'd Lapilli Tuff						
58			QZ		57.6 minor DO stringers						
58			QZ		57.6 - Sericite altered Lapilli Tuff						
58			QZ		58.7 - Moderately vuggy quartz stringered & mod. fractures foliated & pyritized						
59					58.7 - Sericite & Hematite altered Lapilli Tuff						
59			QZ		59.4 - Strong fractures fol'n @ 50° TCA						
59			SS								
59			SX								
60					59.4 - Sericite alt'd Lapilli Tuff						
60	93	22			59.8 - Mod grey ductile Sulfide Content						
60					- Strong fracture fol'n @ 55° TCA						
60					4.1mm QZ stringers, dark grey						
60					hairline SX stringers - AS stringer						
60					(very fine)						
61					59.8 - Sericite / Clay / Hematite altered						
61					61.7 Lapilli Tuff						
62	109	16	DC		Light grey-green to mottled red-brown						
62					- Fracture foliated @ 50° TCA						
62					- Becoming increasingly fault brecciated & clay gouge / altered toward base of interval						
62					faulting @ 50 - 55° TCA						

PAGE 9 OF 13		PROJECT FAWN		HOLE FWJ 73-04						
DEPTH (M)	% CORE REC	% ROD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION				FRACTURE INTENSITY
						MS	HE	CA	CY	
60					59.8- Fault Zone in Lapilli Tuff	/	/	/	/	745
61	93	22			65.0 - Light Gray-green, Extremely MS-Clay altered & variably brecciated throughout. Competent zones with DO stringing and occasional QZ stringing.	/	/	/	/	
62	107	16			Some QZ stringing appears to cut the fault zone	/	/	/	/	
63					- Sphalerite assoc. with QZ strgs.	/	/	/	/	
64	108	12			65.0 - Sericite Altered Lapilli Tuff.	/	/	/	/	740
65					81.5 - Entire zone is strongly brecciated blocky with numerous minor slips, generally clay coated.	/	/	/	/	
66	109	5			- DO & to lesser extent QZ veining occurs as thin stringers & vuggy zones. Stringers commonly < 2mm. Pyrite Halos surround zones of stringer	/	/	/	/	
67					QZ/DO stringing.	/	/	/	/	
68	95	61			65.0 - 67.8 Numerous strong clay slips	/	/	/	/	
69						/	/	/	/	
70	109	42				/	/	/	/	
71						/	/	/	/	
72						/	/	/	/	
73	102	98				/	/	/	/	
74						/	/	/	/	
75	118	45				/	/	/	/	
76						/	/	/	/	
77						/	/	/	/	740
78	99	47				/	/	/	/	
79						/	/	/	/	
80						/	/	/	/	



DRILL LOG

PROJECT FAWN		COLLAR ELEVATION 1425.3			
HOLE FVN 97-05		AZIMUTH 180°			
LOCATION 6+84N 7+10E		DIP -45			
LOGGED BY J. LEHTINEN		LENGTH 96.3			
DRILLED BY FALCON DRILLING		HORIZONTAL PROJECTION			
ASSAYED BY CHEMEX LABS		VERTICAL PROJECTION			
CORE SIZE BATW		ALTERATION SCALE 			
DATE STARTED APRIL 5/97	DATE COMPLETED APRIL 7/97				
DIP TESTS BY Acid		SULPHIDE SCALE 			
DEPTH	DIP			AZIM	DEPTH
96.3	-44.5				
OBJECTIVE					

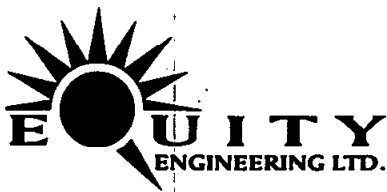
SUMMARY LOG	
Target:	
0 - 3.66 Casing	91.7 - 96.51 Ash + Lapilli Tuff ± minor QZ stringers + TR SP + CP
3.66 - 17.2 Andesite Lapilli Tuff	
17.2 - 18.7 MS-HE altered Lapilli Tuff	
18.7 - 21.9 MS altered Ash Tuff/Lapilli Tuff	
21.9 - 22.4 Fault in Ash/Lapilli Tuff	
22.4 - 30.2 MS+HE altered Lapilli Tuff	
30.2 - 46.35 Andesite Lapilli Tuff	
46.35 - 50.50 HE altered Lapilli Tuff	
50.50 - 78.0 Quartz Feldspar Perphyry with Quartz Stringers ± PY, AS	
78.0 - 89.6 Quartz Feldspar Perphyry	
89.6 - 91.7 Mixed QFP + Ash/Lapilli Tuff = Footwall Zone ± QZ ± PY AS	
91.7 - 96.3 MS altered Ash Lapilli Tuff + QZ stringers ± PY + AS	

PAGE 1 OF 13			PROJECT FAWN		HOLE FWN 97-05					
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION				FRACTURE INTENSITY
						MS	HE	CA	CY	
0					0- Casing					
1					3.66					
2					CASING					
3					3.66- Andesite Lapilli Tuff					
4					13.7 med-dark green					
5	37	0			- Fragments vary in size up to 5cm and are matrix supported.					750
6	46	0			Zone with coarse ash and rare fragments					
7	117	50			- Fragment composition trachytic to perphyritic					
8	78	15			- Alteration restricted to propylitic, weak CA alteration.					
9	83	0			Base of interval = gradational contact with 1st appearance of HE then MS. Magnetics lost through HE zone.					8
10	161	32								
11	113	57								
12										
13	103	33								
14										
15	98	39								
16										
17					17.20- Hematite & Sericite Altered Lapilli Tuff					
18					18.70 - Minor HE near top of interval grading into MS alt'm + weak CY in bottom of interval					
19	99	47			- Numerous angular fragments.					
20					- Gradational into crystal tuff @ base.					
					- Minor Quartz stringers @ base.					
					18.70- Sericite Altered Ash Tuff					
					21.0 - Light to med drab green.					
					- Fine grained ash grading to crystal Tuff at upper + lower contacts. Alteration appears to					

PAGE 3 OF 10		PROJECT FAWN		HOLE FWN 97-05							
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY
						MS	HE	CA	CY	EP	
20			DO	30°	have obliterated most textural features.	/	/	/	/	/	
21			DO	35°	- Upper Contact = clay altered.	/	/	/	/	/	10
22	104	64	DO	50°	* - Vuggy Silicified Zones	/	/	/	/	/	200
23			DO	40°	@ 20.03 - 20.23	/	/	/	/	/	
24			DO	45°	20.50 - 20.75	/	/	/	/	/	
25	98	43	DO	45°	21.13 - 21.90	/	/	/	/	/	8
26			slip		- Zones = Black silicified stringers with trace PY. Black = Vlg sulphides? or Carbon?	/	/	/	/	/	
27					21.0 - Sericite Altered Lapilli Tuff.	/	/	/	/	/	
28					21.9 - Light-med grey green - Silicified Zones (Vuggy) see Above	/	/	/	/	/	5
29					21.9 - Fault.	/	/	/	/	/	
30	93	77	DO	45°	22.4 Clay altered & Brecciated Sericite Lapilli Tuff	/	/	/	/	/	
31			DO	45°	22.4 - Sericite + Hematite altered Lapilli Tuff	/	/	/	/	/	
32			DO	45°	30.2 - Very Distinct Lapilli Tuff.	/	/	/	/	/	
33					- Light - Med green - grey with zones of dark fragments	/	/	/	/	/	
34	103	79			- Fragment supported. Angular to sub-angled, heterolithic fragments	/	/	/	/	/	10
35					- Clay altered near top of interval	/	/	/	/	/	
36			DO		- Fragments 0.5 to 1.0 cm only	/	/	/	/	/	
37	97	41	CA	45°	30.2 - Andesite Lapilli, tuff	/	/	/	/	/	
38			CA	45°	46.35 - Dark green	/	/	/	/	/	
39			CA	45°	- Compositionally identical to unit above (without sericite alteration)	/	/	/	/	/	
40	95	45	CA	45°	- Fragments are less distinct in unaltered version	/	/	/	/	/	
41			DO, HE		- Mod - strongly magnetic	/	/	/	/	/	
42			CA		- minor HE fragments - Minor HE alt in paralleling stringers.	/	/	/	/	/	
43					- Fragment Composition = ophanitic to porphyritic	/	/	/	/	/	
44	108	43			- Angular to sub-rounded	/	/	/	/	/	

DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION				FRACTURE INTENSITY	
						MS	HE	CA	CY		
60	18	16				/					
61	6	0				/					75
62	63	0				/					
63	117	0				/					
64	25	0				/					
65						/					
66	49	13				/					
67	63	0				/					
68	98	26				/					
69	85	0			78.0 - Quartz Feldspar Perphyry	/					
70	81	0			86.6 - Similar to previous interval	/					
	82	0			but colour change of groundmass	/					
71	62	0			to pink-grey.	/					
					- Fracturing appears to be	/					
72	89	0			significantly less with less	/					
	82	0			rubby core - Fractures with	/					
73					Sericite and Clay.	/					
					- Fracturing common along C.P.	/					
74	100	18			and @ 50-65° TC/P	/					
75						/					
76	76	6	81			/					
			66			/					
77			+			/					
			00			/					
78	92	0				/					
79						/					
80						/					

DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY
						MS	HE	CA	CY		
80											
81	126	0									
82	180	0									
82	74	0									45
83					89.6 - Quartz Feldspar Periphyry (Fadwall Zone)						
84	87	0	QEP		89.1 Mixed Zone						
84	113	0			Xenolith(?) inclusions of altered lapilli						
85	57	0			tuff in strongly fractured and stringered QEP.						
86					Gray & white (2 phase) quartz						
87	79	0	QEP		stringers up to 1cm as erratic						
87			QZ		stockwork with PY + AS.						
88	109	0	Fr		Vuggy QZ. Also DO & QZ with GN, SP						
88			Fr		sericite and clay on fracture						
89			Fr		surfaces.						
89	74	7	Fr, QZ, AT		89.1 - Sericite Alteration Ash-Lapilli						
90					Tuff.						
91	101	44	QZ, PY, GY		- Pale green-buff. darker grey						45
91			DO		zones with increased						
92					diss PY. + Patchy PY.						
92					- QZ stringers paralleling						
93					dominant fracture direction						
93					@ 40-55°						
94	78	43	Red Slip		- Alteration(?) banding of PY						
94					+ CY + SE @ 12° TCA @ 90.6m						
95					90.7 = fault Bx + Diss PY						
95					91.6 Banded PY + GY? (50° TCA) 5cm						
96	75	23	Red Slip		91.7 - Interbedded Ash and Lapilli Tuff.						
96	31		QZ, DO, Sp.		96.31 Pale green-beige + buff coloured						
E. O. H.					minor fine ash tuff, inter-bedded						
					with coarse ash & Lapilli tuff.						
					minor QZ stringers (ICP + SP) crosscut						
					& offset by DO stringers @ 20-30° TCA						
					- 92.2 - 93.2 - Black v.l.g. alt'n						
					py = Carbon? = Evaporite + patchy.						

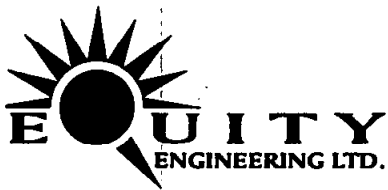


DRILL LOG

PROJECT FAWN				COLLAR ELEVATION 1574.8			
HOLE FWN 97-06				AZIMUTH 180°			
LOCATION 11+98 N 1+81 E				DIP -45			
LOGGED BY J. LEHTINEN				LENGTH 99.4			
DRILLED BY FALCON DRILLING				HORIZONTAL PROJECTION			
ASSAYED BY CHEMEX LABS				VERTICAL PROJECTION			
CORE SIZE				<div style="display: flex; align-items: center;"> <div style="margin-right: 10px;"> </div> <div> <p>ALTERATION SCALE</p> <p>absent</p> <p>slight</p> <p>moderate</p> <p>intense</p> </div> </div>			
DATE STARTED APRIL 7/97		DATE COMPLETED April 9/97					
DIP TESTS BY Acid				<div style="display: flex; align-items: center;"> <div style="margin-right: 10px;"> </div> <div> <p>SULPHIDE SCALE</p> <p>traces only</p> <p>< 1%</p> <p>1% - 3%</p> <p>3% - 10%</p> <p>> 10%</p> </div> </div>			
DEPTH	DIP	AZIM	DEPTH				
90.2	-44.5						
OBJECTIVE							
SUMMARY LOG							
Target:				81.4 - 99.4 Andesite Lapilli Tuff			
0-3.05 Casing							
3.05-43.3 Andesite Flows							
43.3-50.8 MS+CY altered porphyritic andesite							
50.8-53.1 MS+PY altered Andesite + Minor QZ stringers							
53.1-55.35 Quartz Stockwork in Andesite, Stockwork + Q.V. Breccia							
55.35-59.74 MS altered porphyritic andesite							
59.74-61.4 Silicified fault breccia, PY+Tr SP							
61.4-62.5 Fault Breccia							
62.5-69.5 Broken faulted MS altered andesite							
69.5-71.1 MS+PY altered andesite							
71.1-72.5 Andesite Lapilli Tuff							
72.5-81.4 Andesite Flow Breccia							

PAGE 5 OF 10		PROJECT FAWN		HOLE FWN 97-06								
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY	
						MS	HE	CA	CY	SI		PY
40					43.7 - Sericite and Clay altered Porphyritic Andesite							
41	99	36			50.0 Pale green + grey-green with increased PY content -							5
42					- Original porphyritic texture still visible							
43	83	78			- Pyritization throughout interval, increasing along vein margins.							
44					- Pale creamy-buff colored 100 stringers							20
45	88	11	DO	SO	- Minor QZ stringer ~1cm @ 47.5m							
45	145	39			- Fault @ 47.8m - 25° TCA - Gouge + BSX							
46	71	11			- Basal contact sharp @ 40° TCA							
47	99	0			50.0 Strongly Sericite and Pyrite							
47	139	0	QZ	25%	53.1 Altered Andesite with							
48	131	49			minor Quartz veins							20
48	71	6			Increased pyritization with							
49	102	49			5cm banded/foliated DO-QZ							
49					vein at top of interval							
50					- 1cm grey QZ stringers with							
50	92	39			PY - (Dark grey = F.G. AS?)							20
51					QZ = Chalcadonic							
52					53.1 - Quartz Stockwork Zone							
52					55.35 - Dark grey to tan-grey							
53	107	41			- Zone comprised of vuggy							
53					banded QZ + DO veining with							
54					dominantly grey chalcadonic							20
54					& white quartz							
55	102	25			- Stringer stockwork and							
55					minor silicified zones							
56					- Minor vein breccia							
56	105	20			- Mineralization = PY ± AS?							
57					Secondary = Realgar, Tr. SP							8
57	95	18			55.35 - Strong Sericite Altered Porphyritic							
58					59.74 Andesite with interbedded Ash Tuft							
58					Pale green with a dark							
59	112	44			grainy appearance due to							
59					concentrations of PY as							
60					alteration - Minor Tuft (Basal 1.0m)							
60					- Minor QZ-DO-PY stringers							
60					- Sharp basal Contact @ 60°							

PAGE 9 OF 10		PROJECT FANN		HOLE FWN 97-06								
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY	
						MS	HE	CA	CY	EP		PY
80			CA		81.4 - Andesite Lapilli Tuff							
81	117	115			99.4 Similar to unit above but alteration zones produce weakly bleached, lighter zones with distinct lapilli fragments							
82			10cm 100% + PY + AS	20	Fragment's up to 10cm with varying textures from aphanitic to porphyritic							
83	96	73	CA	40	82.2 - 10cm, grey white qz with ≈ 5% PY. Trace AS							
84					Alteration Zone 81.3 - 83.9							
85	110	87		35°	85.6 - 86.2							
86			large mass		Weak alt'n 88.1 - 90.2							
87					- Numerous rounded lapilli fragments with corroded boundaries = (lapilli incorporated into flow?) + fragments with diffuse boundaries							
88			DO	40								
89	104	30	DO + QZ	55								
90			QZ	70 60								
91												
92	110	38										
93	99	240										
94												
95	103	67										
96				35								
97												
98	103	67										
99				50								
100					99.4 F.O.H							



DRILL LOG

PROJECT FAWN				COLLAR ELEVATION 1598.5					
HOLE Fawn 99-07				AZIMUTH 180°					
LOCATION 12+87N 0+62 E				DIP -45°					
LOGGED BY J. LEHTINEN				LENGTH 90.2m					
DRILLED BY FALCON DRILLING				HORIZONTAL PROJECTION					
ASSAYED BY CHEMEX LABS				VERTICAL PROJECTION					
CORE SIZE BQTW				<div style="text-align: right;">ALTERATION SCALE</div> <div style="display: flex; justify-content: space-between; width: 100px; margin-left: 50px;"> 0123 </div> <div style="margin-left: 100px;"> <p>absent</p> <p>slight</p> <p>moderate</p> <p>intense</p> </div>					
DATE STARTED APRIL 9/99		DATE COMPLETED April 11/99							
DIP TESTS BY Acid				<div style="text-align: right;">SULPHIDE SCALE</div> <div style="display: flex; justify-content: space-between; width: 100px; margin-left: 50px;"> 01234 </div> <div style="margin-left: 100px;"> <p>traces only</p> <p>< 1%</p> <p>1% - 3%</p> <p>3% - 10%</p> <p>> 10%</p> </div>					
DEPTH	DIP	AZIM	DEPTH					DIP	AZIM
90.2	-44.5	-							
OBJECTIVE									
SUMMARY LOG									
Target:									
0 - 1.52 Casing									
1.52 - 29.4 Andesite Porphyry									
29.4 - 30.6 MS/HE/CY altered Andesite Porphyry									
30.6 - 35.35 MS/HE altered Lapilli Tuff									
35.35 - 43.7 MS altered Andesite Porphyry									
43.7 - 50.4 Fault Zone in QEP? Lapilli Tuff									
50.4 - 63.6 Andesite Lapilli Tuff									
63.6 - 67.0 Stringer / Fault Zone in Lapilli Tuff									
67.0 - 69.3 Andesite Lapilli Tuff									
69.3 - 86.9 Andesite Flow									
86.9 - 90.22 Andesite Feldspar Porphyry									

PAGE 5 OF 10		PROJECT FAWN		HOLE 97-07								
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY	
						MS	HE	CA	CY	CL		
40					35-35- Sericite altered Andesite(?)	/	/	/	/	/		
41	52	86			43.7 Porphyry	/	/	/	/	/		
42					- Light green-buff, with white to pale green phenocrysts up to 3-4 mm.	/	/	/	/	/		10
43	96	33			- Uniform texture = porphyritic.	/	/	/	/	/		
44	123	0			- Matrix commonly with dissem PY	/	/	/	/	/		
45	64	0			- Pale green phenos = MS ± CH?	/	/	/	/	/		
45	98	0			- Weak clay alteration	/	/	/	/	/		
46	77	0			- Basal 2.0 m = patchy HE alteration + increased CY	/	/	/	/	/		
47	100	6		34°	alteration + increased DO + Chalcidonic Qz stringering	/	/	/	/	/		
48	93	0			43.7- Fault Zone	/	/	/	/	/		
49	101	44			50.4 Extremely broken, dark grey with clay gouge and fault breccia	/	/	/	/	/		
50	121	55		50°	- Fragments are very strongly MS altered and PY is dissem throughout.	/	/	/	/	/		
51					- Fragments of QFP are incorporated into the bx indicating possible narrow dyke or the fragments have been dragged in.	/	/	/	/	/		
52	86	37				/	/	/	/	/		
53					- Possible fault orientation from slips within the zone are Avg = 34° TCA	/	/	/	/	/		
54	103	35				/	/	/	/	/		
55	113	0			- Basal Contact @ 50° TCA	/	/	/	/	/		
56	66	0			- Host Lithology = Lapilli Tuff?	/	/	/	/	/		
56	83	0				/	/	/	/	/		
57	41	0				/	/	/	/	/		
58	44	0				/	/	/	/	/		
59	82	0				/	/	/	/	/		
59	57	58				/	/	/	/	/		
60	19	0				/	/	/	/	/		

PAGE 7 OF 10		PROJECT FAWN		HOLE FWN 97-07							
DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY
						MS	HE	CA	CY	CL	
60	100	0				/	/	/	/	/	
61	80	0	Andesite Core		50.4- Andesite Lapilli Tuff 63.6 - Dark to med green - Fragments up to 3cm. - Extremely broken	/	/	/	/	/	5
62	71	0			- Strong chlorite alteration with zones of strong clay alteration.	/	/	/	/	/	10
63	100	0			- Extremely broken core from 53.70 - 63.6m.	/	/	/	/	/	
64	100	35			47	/	/	/	/	/	
65					- Strong Py alteration in MS alt'd Rx + PY in fractures	/	/	/	/	/	
66			AS, PY CPT PY		- Strong CA alt'n in lighter colored zones.	/	/	/	/	/	
67	100	52				/	/	/	/	/	
68			CA		63.6- Stringer Fault Zone in 67.0 Altered Lapilli Tuff	/	/	/	/	/	
69					- Med grey-green - Very strongly fractured and stringered with DO stringers throughout. Stringing erratic = weak stockwork.	/	/	/	/	/	
70	99	65	DO DO PY		Minor 2cm QZ stringer along C.B. with PY + AS. - Truncated by fault slips 123 @ 66.0-66.1	/	/	/	/	/	
71	99	42			Pyrite finally diss. + stringers	/	/	/	/	/	
72						/	/	/	/	/	
73	86	51	EP.			/	/	/	/	/	
74					67.0- Andesite Lapilli Tuff. 69.3 Dark green with sub-rounded fragments up to 5cm.	/	/	/	/	/	
75	120	33			- Cross-cut by CA, DO ± HE Stringers 2-3mm.	/	/	/	/	/	
76					- Weak epidote patches = amygdules? - Weakly magnetic	/	/	/	/	/	
77	89	49				/	/	/	/	/	
78			CPT PY			/	/	/	/	/	
79	90	47				/	/	/	/	/	
80						/	/	/	/	/	

DEPTH (M)	% CORE REC	% RQD	LITHOLOGY	STRUCTURE	GEOLOGICAL DESCRIPTION	ALTERATION					FRACTURE INTENSITY	
						MS	HE	CA	CY	CL		
80			CA + EP + CH	28°	69.3 - Andesite							
81	70	47		30°	86.9 Dark green-grey to grey - Patchy amygdaloidal zones - Generally aphanitic to fine grained - minor CA + EP + PY stringers throughout interval - Strong Magnetic							5
82												
83	64	6			78.1 - 78.8 = Fractures (Discrete with PY + CP fill overall = 21% CP) <small>in this sample section (1.5m)</small> - Basal contact = so TCA							
84	16	52		38°								
85												
86	77	70		50°	86.9 - Andesite Feldspar Porphyry 90.22 Light grey phenas in dark grey matrix. - Phen's up to 2mm, moderately packed - Very weak CH alt'n. - PY erratic as discs + fine fill - Strongly Magnetic							
87												
88												
89	100	71										
90					90.22 = E.O.H.							

1997 DIAMOND DRILL CORE SAMPLES

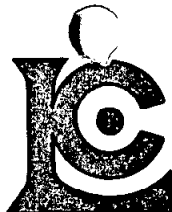
Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
FWN97-01											
197001	3.05	4.55	1.50	30	---	1.0	122	3	86	4	880
197002	10.10	11.00	0.90	<5	---	0.8	14	23	60	8	570
197003	11.00	12.50	1.50	<5	---	<.2	10	<1	10	4	286
197004	16.00	17.00	1.00	<5	---	<.2	4	5	8	4	194
197005	17.00	18.20	1.20	5	---	1.0	6	57	16	<2	350
197006	18.20	19.83	1.63	<5	---	<.2	8	13	10	2	140
197007	19.83	21.60	1.77	<5	---	<.2	10	16	12	2	108
197008	21.60	22.90	1.30	<5	---	<.2	14	11	2	<2	78
197009	22.90	24.20	1.30	10	---	0.4	6	169	8	2	180
197010	24.20	25.50	1.30	<5	---	<.2	<2	1	2	<2	136
197011	25.50	26.20	0.70	<5	---	<.2	4	<1	4	2	246
197012	26.20	27.40	1.20	<5	---	<.2	12	<1	10	2	396
197013	27.40	28.80	1.40	<5	---	0.6	6	13	4	<2	420
197014	31.90	33.00	1.10	<5	---	0.4	14	44	10	14	142
197015	44.20	44.70	0.50	5	---	3.8	74	112	76	42	258
197016	47.60	49.60	2.00	<5	---	<.2	6	10	4	6	80
197017	56.00	57.10	1.10	<5	---	<.2	14	10	6	2	240
197018	57.10	58.80	1.70	<5	---	<.2	14	5	2	2	92
197019	58.80	60.15	1.35	365	---	2.8	928	27	60	20	440
197020	60.15	61.90	1.75	<5	---	0.4	32	8	34	6	178
197021	61.90	63.00	1.10	2000	2.02	6.0	2270	9	84	10	508
197022	63.00	64.00	1.00	10	---	<.2	26	2	12	8	112
FWN97-02											
197023	3.05	4.50	1.45	<5	---	1.0	36	23	28	4	372
197024	13.40	14.70	1.30	170	---	1.2	6	28	262	<2	484
197025	48.00	50.00	2.00	<5	---	<.2	16	50	16	12	122
197026	50.00	52.20	2.20	130	---	3.8	264	28	44	14	154
197027	52.20	53.20	1.00	<5	---	<.2	22	9	8	<2	56
197028	56.90	58.50	1.60	20	---	0.4	166	5	14	6	142
197029	58.50	59.30	0.80	10	---	0.6	52	25	54	8	320
197030	60.70	62.50	1.80	<5	---	0.4	20	15	36	4	292
197031	62.50	64.00	1.50	20	---	2.6	12	18	60	<2	388
197032	64.00	64.90	0.90	15	---	1.8	20	81	56	6	398
FWN97-03											
197033	6.40	8.23	1.83	10	---	<.2	8	49	2	<2	96
197034	41.80	43.10	1.30	<5	---	<.2	88	44	6	2	88
197035	43.10	44.70	1.60	<5	---	0.2	44	46	2	8	88
197036	44.70	46.20	1.50	<5	---	<.2	2	12	<2	2	80
197037	46.20	47.30	1.10	<5	---	<.2	12	6	2	<2	94
197038	51.70	54.00	2.30	5	---	0.4	56	25	4	2	100
197039	63.15	65.53	2.38	10	---	0.2	306	23	8	2	138
197040	65.53	67.60	2.07	<5	---	0.4	28	19	10	4	122
197041	67.60	70.41	2.81	60	---	0.6	254	7	38	2	246
197042	70.41	74.10	3.69	575	---	2.2	1750	11	68	6	348
197043	76.20	77.00	0.80	25	---	0.6	206	2	18	2	246
197044	77.00	78.90	1.90	345	---	1.4	1530	12	24	8	186
197045	78.90	81.50	2.60	345	---	1.8	1420	14	36	8	224

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
FWN97-04											
197046	9.70	11.30	1.60	<5	---	0.6	24	42	6	8	114
197047	11.30	12.90	1.60	<5	---	0.6	16	37	38	6	206
197048	12.90	14.30	1.40	<5	---	0.2	18	31	2	6	86
197049	14.30	15.80	1.50	130	---	<.2	14	46	2	12	120
197050	33.40	35.10	1.70	80	---	2.4	508	16	14	16	122
197051	37.10	39.20	2.10	<5	---	3.2	214	17	14	12	128
197052	39.20	40.70	1.50	<5	---	0.8	24	27	2	6	46
197053	46.50	47.50	1.00	<5	---	0.6	58	28	6	6	94
197054	48.50	49.00	0.50	30	---	1.4	172	11	12	12	96
197055	49.50	50.80	1.30	<5	---	0.4	36	28	6	8	64
197056	50.80	52.10	1.30	70	---	2.8	212	28	16	14	156
197057	56.30	57.20	0.90	125	---	1.2	282	10	16	8	70
197058	57.20	57.60	0.40	<5	---	0.2	36	8	10	6	104
197059	57.60	58.70	1.10	130	---	3.2	370	27	18	12	176
197060	58.70	59.40	0.70	<5	---	<.2	18	4	2	2	84
197061	59.40	59.80	0.40	580	---	6.6	3690	14	50	14	266
197062	59.80	61.50	1.70	<5	---	<.2	20	7	<2	2	82
197063	61.50	62.40	0.90	10	---	0.2	176	7	10	4	142
197064	62.40	63.30	0.90	30	---	0.6	500	12	12	6	224
197065	63.30	65.00	1.70	140	---	1.8	622	11	30	8	184
197066	65.00	66.50	1.50	<5	---	0.2	16	8	6	<2	100
197067	66.50	67.40	0.90	120	---	1.2	1200	17	18	10	128
197068	69.20	70.80	1.60	<5	---	<.2	26	13	<2	<2	98
197069	72.20	73.00	0.80	30	---	0.6	150	7	20	4	618
197070	73.90	74.80	0.90	50	---	0.4	220	29	8	8	142
197071	74.80	75.90	1.10	275	---	2.2	638	17	24	8	284
197072	75.90	77.80	1.90	10	---	1.0	54	38	24	6	908
197073	77.80	79.30	1.50	50	---	1.0	152	34	6	10	86
197074	79.30	81.50	2.20	30	---	0.6	98	34	<2	10	102
197080	83.82	85.04	1.22	<5	---	<.2	8	<1	6	<2	8
197081	86.00	87.30	1.30	10	---	2.4	32	47	8	8	76
FWN97-05											
197075	18.00	20.00	2.00	<5	---	0.2	26	103	4	24	80
197076	20.00	21.00	1.00	<5	---	<.2	52	18	14	6	140
197077	21.00	21.90	0.90	<5	---	<.2	32	8	30	12	124
197078	21.90	22.40	0.50	10	---	0.2	246	42	12	14	138
197079	22.40	24.20	1.80	<5	---	<.2	80	39	2	6	72
197082	50.00	50.50	0.50	<5	---	0.2	70	18	10	6	112
197083	50.50	52.10	1.60	95	---	1.8	752	2	12	8	170
197084	52.10	53.03	0.93	60	---	1.4	142	1	8	<2	12
197085	53.03	54.50	1.47	25	---	3.8	46	10	10	4	18
197086	54.50	56.00	1.50	10	---	3.2	20	10	10	4	14
197087	56.00	57.50	1.50	5	---	0.8	36	5	12	2	14
197088	57.50	59.00	1.50	10	---	1.4	34	4	8	<2	12
197089	59.00	60.50	1.50	10	---	0.8	30	1	10	2	8
197090	60.50	62.80	2.30	55	---	0.6	120	1	14	<2	18
197091	62.80	65.90	3.10	20	---	<.2	28	<1	2	<2	2
197092	86.60	87.90	1.30	575	---	4.4	1295	8	10	6	74

Sample Number	From (m)	To (m)	Length (m)	Au (ppb)	Au (g/t)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
FWN97-05											
197093	87.90	89.10	1.20	210	---	3.0	892	12	12	8	106
197094	89.10	90.60	1.50	1620	1.58	0.2	16	31	62	2	312
197095	90.60	91.70	1.10	300	---	0.6	62	47	56	6	368
197096	91.70	93.30	1.60	<5	---	0.8	114	35	6	8	38
197097	94.60	96.31	1.71	<5	---	0.2	114	62	2	16	58
FWN97-06											
197098	47.30	49.00	1.70	30	---	0.4	282	30	30	8	370
197111	49.00	50.00	1.00	15	---	<.2	32	13	20	2	344
197099	50.00	51.20	1.20	30	---	0.8	240	37	40	12	376
197100	51.20	53.10	1.90	680	---	7.2	2350	45	34	24	258
197101	53.10	54.00	0.90	1680	1.61	42.4	1835	29	36	60	288
197102	54.00	55.35	1.35	1365	1.23	56.8	2690	24	56	64	260
197103	55.35	57.00	1.65	2340	2.33	11.0	5370	52	54	22	428
197104	57.00	59.74	2.74	<5	---	3.0	232	78	26	20	244
197105	59.74	61.40	1.66	1700	1.68	50.0	2640	34	202	34	656
197106	61.40	62.50	1.10	65	---	3.0	246	76	84	22	460
197107	62.50	64.00	1.50	<5	---	0.2	16	31	64	2	314
197108	68.00	69.50	1.50	10	---	0.6	66	48	54	4	380
197109	69.50	71.00	1.50	250	---	1.8	1385	22	28	8	266
197110	71.00	72.40	1.40	<5	---	<.2	14	26	2	2	116
197112	81.40	83.90	2.50	160	---	1.0	332	35	14	6	124
197113	85.60	86.20	0.60	25	---	1.0	32	27	10	2	96
197114	88.00	89.90	1.90	<5	---	<.2	6	64	6	<2	104
FWN97-07											
197115	29.30	30.60	1.30	140	---	0.6	120	21	12	6	78
197116	30.60	32.30	1.70	<5	---	<.2	12	59	2	<2	92
197117	32.30	33.40	1.10	<5	---	<.2	32	33	6	4	78
197118	33.40	35.35	1.95	<5	---	<.2	30	33	6	6	314
197119	35.35	37.50	2.15	<5	---	0.2	32	68	6	14	98
197120	42.10	43.70	1.60	<5	---	<.2	10	60	8	14	66
197121	43.70	46.00	2.30	30	---	0.4	188	35	12	14	98
197122	46.00	47.85	1.85	10	---	0.2	20	54	8	16	60
197123	47.85	48.70	0.85	<5	---	<.2	22	22	8	4	38
197124	48.70	49.68	0.98	30	---	<.2	12	9	6	<2	30
197125	49.68	50.40	0.72	30	---	0.6	94	76	12	16	84
197126	63.60	65.40	1.80	<5	---	<.2	8	26	8	<2	68
197127	65.40	67.00	1.60	165	---	0.8	1890	17	20	6	90
197128	78.10	78.90	0.80	25	---	13.4	32	4060	490	4	1610
197129	87.17	88.20	1.03	<5	---	<.2	16	50	8	<2	90

APPENDIX D

CERTIFICATES OF ANALYSIS



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

A9720423

Comments: ATTN:S.HARRIS/J.LEHTINEN

CERTIFICATE

A9720423

(EIA) - EQUITY ENGINEERING LTD.

Project: WKM 97-06
P.O. #:

Samples submitted to our lab in Vancouver, BC.
This report was printed on 21-AUG-97.

SAMPLE PREPARATION

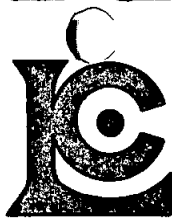
CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION
205	129	Geochem ring to approx 150 mesh
226	129	0-3 Kg crush and split
3202	129	Rock - save entire reject
229	129	ICP - AQ Digestion charge

* NOTE 1:

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.

ANALYTICAL PROCEDURES

CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION	METHOD	DETECTION LIMIT	UPPER LIMIT
983	129	Au ppb: Fuse 30 g sample	FA-AAS	5	10000
997	6	Au g/t: 1 assay ton, grav.	FA-GRAVIMETRIC	0.07	1000.0
2118	129	Ag ppm: 32 element, soil & rock	ICP-AES	0.2	100.0
2119	129	Al %: 32 element, soil & rock	ICP-AES	0.01	15.00
2120	129	As ppm: 32 element, soil & rock	ICP-AES	2	10000
2121	129	Ba ppm: 32 element, soil & rock	ICP-AES	10	10000
2122	129	Be ppm: 32 element, soil & rock	ICP-AES	0.5	100.0
2123	129	Bi ppm: 32 element, soil & rock	ICP-AES	2	10000
2124	129	Ca %: 32 element, soil & rock	ICP-AES	0.01	15.00
2125	129	Cd ppm: 32 element, soil & rock	ICP-AES	0.5	100.0
2126	129	Co ppm: 32 element, soil & rock	ICP-AES	1	10000
2127	129	Cr ppm: 32 element, soil & rock	ICP-AES	1	10000
2128	129	Cu ppm: 32 element, soil & rock	ICP-AES	1	10000
2150	129	Fe %: 32 element, soil & rock	ICP-AES	0.01	15.00
2130	129	Ga ppm: 32 element, soil & rock	ICP-AES	10	10000
2131	129	Hg ppm: 32 element, soil & rock	ICP-AES	1	10000
2132	129	K %: 32 element, soil & rock	ICP-AES	0.01	10.00
2151	129	La ppm: 32 element, soil & rock	ICP-AES	10	10000
2134	129	Mg %: 32 element, soil & rock	ICP-AES	0.01	15.00
2135	129	Mn ppm: 32 element, soil & rock	ICP-AES	5	10000
2136	129	Mo ppm: 32 element, soil & rock	ICP-AES	1	10000
2137	129	Na %: 32 element, soil & rock	ICP-AES	0.01	5.00
2138	129	Ni ppm: 32 element, soil & rock	ICP-AES	1	10000
2139	129	P ppm: 32 element, soil & rock	ICP-AES	10	10000
2140	129	Pb ppm: 32 element, soil & rock	ICP-AES	2	10000
2141	129	Sb ppm: 32 element, soil & rock	ICP-AES	2	10000
2142	129	Sc ppm: 32 elements, soil & rock	ICP-AES	1	10000
2143	129	Sr ppm: 32 element, soil & rock	ICP-AES	1	10000
2144	129	Ti %: 32 element, soil & rock	ICP-AES	0.01	5.00
2145	129	Tl ppm: 32 element, soil & rock	ICP-AES	10	10000
2146	129	U ppm: 32 element, soil & rock	ICP-AES	10	10000
2147	129	V ppm: 32 element, soil & rock	ICP-AES	1	10000
2148	129	W ppm: 32 element, soil & rock	ICP-AES	10	10000
2149	129	Zn ppm: 32 element, soil & rock	ICP-AES	2	10000



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers
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 British Columbia, Canada V7J 2C1
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To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
 VANCOUVER, BC
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Project: WKM 97-06
 Comments: ATTN:S.HARRIS/J.LEHTINEN

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CERTIFICATE OF ANALYSIS A9720423

SAMPLE	PREP CODE	Au ppb FA+AA	Au FA g/t	Ag ppm	Al %	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K %	La ppm	Mg %
197001	205 226	30	-----	1.0	0.60	122	20	0.5	2	0.77	3.5	22	24	3	5.21	< 10	1	0.28	< 10	0.38
197002	205 226	< 5	-----	0.8	0.75	14	30	0.5	< 2	0.66	1.0	19	16	23	4.94	< 10	1	0.28	10	0.40
197003	205 226	< 5	-----	< 0.2	0.99	10	30	0.5	< 2	0.86	< 0.5	19	12	< 1	5.79	< 10	< 1	0.31	10	0.48
197004	205 226	< 5	-----	< 0.2	0.74	4	30	0.5	< 2	1.83	< 0.5	17	12	5	4.37	< 10	< 1	0.26	< 10	0.75
197005	205 226	5	-----	1.0	0.46	6	20	0.5	< 2	2.99	0.5	19	9	57	4.53	< 10	< 1	0.20	< 10	1.09
197006	205 226	< 5	-----	< 0.2	0.82	8	30	1.0	< 2	1.89	< 0.5	17	11	13	4.55	< 10	< 1	0.27	10	0.68
197007	205 226	< 5	-----	< 0.2	0.67	10	30	1.0	< 2	4.33	< 0.5	20	12	16	4.59	< 10	< 1	0.20	10	0.87
197008	205 226	< 5	-----	< 0.2	2.22	14	60	0.5	< 2	4.17	< 0.5	19	15	11	4.57	< 10	< 1	0.18	10	1.49
197009	205 226	10	-----	0.4	0.86	6	30	0.5	< 2	3.31	0.5	21	15	169	5.06	< 10	< 1	0.20	< 10	1.00
197010	205 226	< 5	-----	< 0.2	0.93	< 2	30	0.5	< 2	1.02	< 0.5	18	7	1	5.17	< 10	< 1	0.23	< 10	0.51
197011	205 226	< 5	-----	< 0.2	0.94	4	20	0.5	< 2	1.04	0.5	22	10	< 1	6.82	< 10	< 1	0.21	10	0.62
197012	205 226	< 5	-----	< 0.2	1.01	12	20	0.5	< 2	0.72	0.5	21	7	< 1	4.73	< 10	< 1	0.23	< 10	0.36
197013	205 226	< 5	-----	0.6	0.91	6	40	0.5	< 2	2.22	0.5	22	10	13	4.30	< 10	1	0.23	< 10	0.82
197014	205 226	< 5	-----	0.4	0.91	14	30	0.5	< 2	3.61	< 0.5	19	11	44	4.70	< 10	< 1	0.20	< 10	1.17
197015	205 226	5	-----	3.8	0.74	74	30	0.5	< 2	0.52	1.5	19	22	112	2.45	< 10	1	0.40	< 10	0.25
197016	205 226	< 5	-----	< 0.2	0.82	6	20	0.5	< 2	3.00	< 0.5	16	9	10	4.41	< 10	< 1	0.20	10	0.83
197017	205 226	< 5	-----	< 0.2	0.81	14	30	0.5	< 2	1.16	0.5	16	8	10	4.07	< 10	< 1	0.26	10	0.44
197018	205 226	< 5	-----	< 0.2	0.72	14	20	0.5	< 2	0.99	< 0.5	21	8	5	5.53	< 10	< 1	0.25	10	0.42
197019	205 226	365	-----	2.8	0.55	928	30	0.5	< 2	1.15	3.0	29	14	27	5.09	< 10	< 1	0.32	< 10	0.30
197020	205 226	< 5	-----	0.4	0.79	32	30	1.0	< 2	0.59	0.5	27	7	8	5.92	< 10	< 1	0.32	10	0.31
197021	205 226	2000	2.02	6.0	0.77	2270	40	1.0	< 2	2.03	3.5	25	20	9	5.70	< 10	1	0.39	< 10	0.50
197022	205 226	10	-----	< 0.2	0.87	26	20	1.0	< 2	1.98	< 0.5	20	9	2	6.03	< 10	< 1	0.22	< 10	0.72
197023	205 226	< 5	-----	1.0	0.99	36	30	1.0	< 2	0.58	2.5	27	10	23	6.10	< 10	< 1	0.20	10	0.12
197024	205 226	170	-----	1.2	1.70	6	30	< 0.5	< 2	3.83	2.0	20	15	28	4.85	< 10	1	0.23	< 10	1.29
197025	205 226	< 5	-----	< 0.2	1.87	16	30	1.5	< 2	2.64	0.5	22	15	50	5.77	< 10	< 1	0.29	< 10	0.84
197026	205 226	130	-----	3.8	0.79	264	30	1.5	< 2	0.72	0.5	28	10	28	5.21	< 10	< 1	0.36	< 10	0.16
197027	205 226	< 5	-----	< 0.2	1.00	22	30	1.5	< 2	1.04	< 0.5	23	10	9	6.16	< 10	1	0.26	10	0.40
197028	205 226	20	-----	0.4	0.69	166	40	1.0	< 2	2.00	< 0.5	29	14	5	5.35	< 10	1	0.32	< 10	0.55
197029	205 226	10	-----	0.6	0.64	52	40	0.5	< 2	0.43	1.5	21	11	25	4.05	< 10	< 1	0.36	10	0.12
197030	205 226	< 5	-----	0.4	0.68	20	30	0.5	< 2	0.51	1.0	23	12	15	5.01	< 10	< 1	0.31	10	0.34
197031	205 226	20	-----	2.6	0.65	12	30	0.5	< 2	1.52	1.5	30	15	18	6.07	< 10	< 1	0.29	< 10	0.53
197032	205 226	15	-----	1.8	0.76	20	30	0.5	< 2	2.81	2.5	18	13	81	4.37	< 10	< 1	0.29	< 10	0.75
197033	205 226	10	-----	< 0.2	2.46	8	40	< 0.5	< 2	4.53	< 0.5	15	21	49	4.31	< 10	< 1	0.18	< 10	0.97
197034	205 226	< 5	-----	< 0.2	0.87	88	30	1.0	< 2	1.82	< 0.5	19	12	44	4.37	< 10	< 1	0.36	10	0.60
197035	205 226	< 5	-----	0.2	0.85	44	40	0.5	< 2	2.82	< 0.5	18	17	46	4.50	< 10	< 1	0.31	< 10	0.86
197036	205 226	< 5	-----	< 0.2	0.87	2	160	0.5	< 2	3.42	< 0.5	17	12	12	4.66	< 10	< 1	0.17	< 10	1.14
197037	205 226	< 5	-----	< 0.2	0.91	12	30	0.5	< 2	2.12	< 0.5	21	13	6	5.32	< 10	< 1	0.14	< 10	0.89
197038	205 226	5	-----	0.4	0.86	56	110	0.5	< 2	3.53	< 0.5	17	13	25	4.31	< 10	< 1	0.32	< 10	1.16
197039	205 226	10	-----	0.2	1.03	306	40	1.0	< 2	1.37	< 0.5	16	14	23	4.48	< 10	1	0.39	10	0.44
197040	205 226	< 5	-----	0.4	1.19	28	30	1.5	< 2	0.95	< 0.5	14	12	19	4.48	< 10	< 1	0.33	10	0.37

CERTIFICATION: *Heath Bechler*



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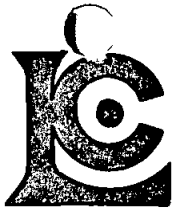
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SAMPLE	PREP CODE		Mn	Mo	Na	Ni	P	Pb	Sb	Sc	Sr	Ti	Tl	U	V	W	Zn
			ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm
197001	205	226	6010	< 1	< 0.01	3	1130	86	4	4	15	< 0.01	< 10	< 10	30	< 10	880
197002	205	226	6880	< 1	< 0.01	2	1500	60	8	4	14	< 0.01	< 10	< 10	32	< 10	570
197003	205	226	4430	< 1	< 0.01	1	1420	10	4	8	15	< 0.01	< 10	< 10	61	< 10	286
197004	205	226	2060	< 1	< 0.01	3	1090	8	4	5	33	< 0.01	< 10	< 10	49	< 10	194
197005	205	226	3360	< 1	< 0.01	2	1300	16	< 2	4	38	< 0.01	< 10	< 10	35	< 10	350
197006	205	226	2390	< 1	< 0.01	3	1320	10	2	7	31	< 0.01	< 10	< 10	63	< 10	140
197007	205	226	1900	< 1	< 0.01	3	1370	12	2	8	75	< 0.01	< 10	< 10	60	< 10	108
197008	205	226	1395	< 1	< 0.02	3	1280	2	< 2	6	77	< 0.01	< 10	< 10	62	< 10	78
197009	205	226	3090	< 1	< 0.01	4	1280	8	2	8	40	< 0.01	< 10	< 10	72	< 10	180
197010	205	226	5110	< 1	< 0.01	< 1	1310	2	< 2	7	15	< 0.01	< 10	< 10	66	< 10	136
197011	205	226	4510	< 1	< 0.01	3	1380	4	2	8	16	< 0.01	< 10	< 10	94	< 10	246
197012	205	226	3790	< 1	< 0.01	2	1450	10	2	11	15	< 0.01	< 10	< 10	69	< 10	396
197013	205	226	3900	< 1	< 0.01	3	1210	4	< 2	6	20	< 0.01	< 10	< 10	54	< 10	420
197014	205	226	2780	< 1	< 0.01	4	1200	10	14	7	59	< 0.01	< 10	< 10	83	< 10	142
197015	205	226	1965	< 1	< 0.01	3	910	76	42	6	12	< 0.01	< 10	< 10	38	< 10	258
197016	205	226	1635	< 1	< 0.01	3	1380	4	6	6	36	< 0.01	< 10	< 10	65	< 10	80
197017	205	226	3700	< 1	< 0.01	1	1540	6	2	6	17	< 0.01	< 10	< 10	42	< 10	240
197018	205	226	5790	< 1	< 0.01	1	1410	2	2	8	12	< 0.01	< 10	< 10	68	< 10	92
197019	205	226	4800	< 1	< 0.01	3	1190	60	20	6	14	< 0.01	< 10	< 10	31	< 10	440
197020	205	226	4990	< 1	< 0.01	2	1400	34	6	9	11	< 0.01	< 10	< 10	54	< 10	178
197021	205	226	4630	< 1	< 0.01	3	1330	84	10	7	20	< 0.01	< 10	< 10	34	< 10	508
197022	205	226	4590	< 1	< 0.01	< 1	1370	12	8	10	34	< 0.01	< 10	< 10	107	< 10	112
197023	205	226	3740	< 1	< 0.01	3	1420	28	4	10	20	< 0.01	< 10	< 10	89	< 10	372
197024	205	226	5230	< 1	< 0.01	1	1430	262	< 2	3	62	< 0.01	< 10	< 10	38	< 10	484
197025	205	226	1885	< 1	< 0.01	3	1210	16	12	6	33	< 0.01	< 10	< 10	67	< 10	122
197026	205	226	2180	< 1	< 0.01	4	1260	44	14	4	12	< 0.01	< 10	< 10	21	< 10	154
197027	205	226	>10000	< 1	< 0.01	5	1230	8	< 2	8	14	< 0.01	< 10	< 10	56	< 10	56
197028	205	226	4050	< 1	< 0.01	5	1200	14	6	10	17	< 0.01	< 10	< 10	30	< 10	142
197029	205	226	1595	< 1	< 0.01	4	1380	54	8	5	11	< 0.01	< 10	< 10	23	< 10	320
197030	205	226	3290	< 1	< 0.01	1	1380	36	4	7	14	< 0.01	< 10	< 10	36	< 10	292
197031	205	226	5340	5	0.01	2	1390	60	< 2	5	25	< 0.01	< 10	< 10	36	< 10	388
197032	205	226	8250	1	0.01	4	1170	56	6	3	40	< 0.01	< 10	< 10	35	< 10	398
197033	205	226	1135	1	0.08	2	2080	2	< 2	2	95	< 0.01	< 10	< 10	49	< 10	96
197034	205	226	2570	< 1	< 0.01	1	1480	6	2	4	55	< 0.01	< 10	< 10	39	< 10	88
197035	205	226	2110	< 1	< 0.01	1	1320	2	8	4	54	< 0.01	< 10	< 10	40	< 10	88
197036	205	226	1385	< 1	< 0.01	1	1420	< 2	2	5	66	< 0.01	< 10	< 10	64	< 10	80
197037	205	226	1900	< 1	< 0.01	1	1430	2	< 2	6	36	< 0.01	< 10	< 10	75	< 10	94
197038	205	226	1745	< 1	< 0.01	2	1140	4	2	6	71	< 0.01	< 10	< 10	49	< 10	100
197039	205	226	4280	< 1	< 0.01	1	1690	8	2	6	40	< 0.01	< 10	< 10	35	< 10	138
197040	205	226	4590	< 1	< 0.01	< 1	1670	10	4	5	34	< 0.01	< 10	< 10	43	< 10	122

CERTIFICATION: _____



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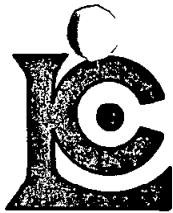
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SAMPLE	PREP CODE	Au ppb FA+AA	Au FA g/t	Ag ppm	Al %	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K %	La ppm	Mg %
197041	205 226	60	-----	0.6	1.09	254	70	1.0	< 2	0.60	1.0	16	34	7	3.14	< 10	1	0.51	10	0.16
197042	205 226	575	-----	2.2	0.40	1750	20	0.5	< 2	1.18	1.5	18	42	11	2.63	< 10	< 1	0.27	< 10	0.27
197043	205 226	25	-----	0.6	0.87	206	30	1.5	< 2	1.13	0.5	32	12	2	5.44	< 10	< 1	0.51	< 10	0.23
197044	205 226	345	-----	1.4	0.56	1530	20	1.0	< 2	0.62	0.5	26	24	12	3.76	< 10	< 1	0.36	< 10	0.15
197045	205 226	345	-----	1.8	1.02	1420	30	1.0	< 2	0.83	2.0	22	24	14	3.54	< 10	< 1	0.58	10	0.20
197046	205 226	< 5	-----	0.6	0.72	24	30	0.5	< 2	0.65	< 0.5	27	10	42	5.55	< 10	< 1	0.23	10	0.53
197047	205 226	< 5	-----	0.6	1.16	16	60	0.5	< 2	0.52	2.0	22	14	37	5.32	< 10	< 1	0.41	10	0.38
197048	205 226	< 5	-----	0.2	0.86	18	50	0.5	< 2	0.71	< 0.5	24	12	31	5.22	< 10	< 1	0.32	10	0.45
197049	205 226	130	-----	< 0.2	1.14	14	90	0.5	< 2	1.88	< 0.5	18	15	46	5.00	< 10	< 1	0.40	10	0.91
197050	205 226	80	-----	2.4	0.97	508	30	2.0	< 2	1.27	< 0.5	27	8	16	5.67	< 10	1	0.43	< 10	0.31
197051	205 226	< 5	-----	3.2	1.32	214	40	1.5	< 2	0.51	< 0.5	24	16	17	4.82	< 10	< 1	0.70	< 10	0.10
197052	205 226	< 5	-----	0.8	0.99	24	100	1.5	< 2	0.69	< 0.5	26	10	27	5.46	< 10	< 1	0.27	< 10	0.38
197053	205 226	< 5	-----	0.6	1.36	58	30	1.5	< 2	0.80	< 0.5	21	14	28	3.89	< 10	1	0.74	10	0.26
197054	205 226	30	-----	1.4	1.17	172	20	2.0	< 2	0.64	< 0.5	29	7	11	5.99	< 10	< 1	0.46	< 10	0.22
197055	205 226	< 5	-----	0.4	1.29	36	30	2.0	< 2	0.85	< 0.5	22	8	28	5.36	< 10	1	0.52	< 10	0.35
197056	205 226	70	-----	2.8	0.75	212	10	1.5	< 2	0.62	< 0.5	26	7	28	5.36	< 10	< 1	0.44	< 10	0.12
197057	205 226	125	-----	1.2	0.94	282	40	1.0	< 2	0.61	< 0.5	21	17	10	4.32	< 10	1	0.52	< 10	0.16
197058	205 226	< 5	-----	0.2	0.70	36	110	1.0	< 2	0.63	< 0.5	18	7	8	5.42	< 10	< 1	0.33	< 10	0.31
197059	205 226	130	-----	3.2	1.14	370	40	1.5	< 2	0.50	0.5	16	24	27	3.78	< 10	< 1	0.53	10	0.12
197060	205 226	< 5	-----	< 0.2	0.89	18	30	1.5	< 2	0.99	< 0.5	16	7	4	4.67	< 10	1	0.30	10	0.46
197061	205 226	580	-----	6.6	1.40	3690	30	2.0	< 2	0.61	< 0.5	29	14	14	4.26	< 10	< 1	0.59	10	0.13
197062	205 226	< 5	-----	< 0.2	0.92	20	100	1.0	< 2	1.33	< 0.5	15	7	7	3.94	< 10	< 1	0.32	10	0.53
197063	205 226	10	-----	0.2	1.27	176	40	1.5	< 2	0.80	< 0.5	15	11	7	3.86	< 10	< 1	0.52	10	0.29
197064	205 226	30	-----	0.6	0.93	500	30	1.5	< 2	1.07	0.5	15	8	12	4.05	< 10	< 1	0.44	10	0.30
197065	205 226	140	-----	1.8	1.02	622	80	1.0	< 2	0.94	< 0.5	17	10	11	4.46	< 10	< 1	0.43	10	0.30
197066	205 226	< 5	-----	0.2	0.85	16	40	0.5	< 2	1.29	< 0.5	20	9	8	4.54	< 10	< 1	0.36	10	0.51
197067	205 226	120	-----	1.2	1.10	1200	40	1.0	< 2	0.81	< 0.5	24	17	17	4.54	< 10	1	0.53	10	0.26
197068	205 226	< 5	-----	< 0.2	0.83	26	30	0.5	< 2	1.26	< 0.5	20	10	13	4.39	< 10	< 1	0.29	10	0.50
197069	205 226	30	-----	0.6	1.03	150	60	1.5	< 2	2.91	2.0	21	16	7	5.35	< 10	< 1	0.55	< 10	0.82
197070	205 226	50	-----	0.4	0.69	220	80	1.0	< 2	1.79	< 0.5	21	9	29	4.82	< 10	< 1	0.34	10	0.60
197071	205 226	275	-----	2.2	0.88	638	60	1.0	< 2	1.94	0.5	21	19	17	4.80	< 10	< 1	0.39	< 10	0.60
197072	205 226	10	-----	1.0	0.60	54	80	0.5	< 2	1.74	3.0	37	10	38	5.52	< 10	< 1	0.37	< 10	0.59
197073	205 226	50	-----	1.0	1.07	152	100	1.0	< 2	2.27	< 0.5	26	12	34	5.76	< 10	1	0.45	< 10	0.95
197074	205 226	30	-----	0.6	0.59	98	50	1.0	< 2	4.00	< 0.5	30	16	34	5.08	< 10	1	0.35	< 10	1.33
197075	205 226	< 5	-----	0.2	1.44	26	70	1.0	< 2	0.47	< 0.5	16	16	103	3.33	< 10	< 1	0.31	10	0.21
197076	205 226	< 5	-----	< 0.2	0.87	52	40	0.5	< 2	0.45	< 0.5	18	14	18	3.68	< 10	< 1	0.45	< 10	0.20
197077	205 226	< 5	-----	< 0.2	0.62	32	30	0.5	< 2	0.34	< 0.5	13	15	8	2.41	< 10	< 1	0.26	< 10	0.18
197078	205 226	10	-----	0.2	0.66	246	30	0.5	< 2	0.45	< 0.5	20	12	42	3.92	< 10	< 1	0.32	< 10	0.19
197079	205 226	< 5	-----	< 0.2	1.41	80	60	1.5	< 2	1.72	< 0.5	18	12	39	4.73	< 10	< 1	0.43	10	0.66
197080	205 226	< 5	-----	< 0.2	0.48	8	< 10	0.5	< 2	0.27	< 0.5	< 1	26	< 1	0.38	< 10	< 1	0.27	< 10	0.04

CERTIFICATION:



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers
 212 Brooksbank Ave., North Vancouver
 British Columbia, Canada V7J 2C1
 PHONE: 604-984-0221 FAX: 604-984-0218

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
 VANCOUVER, BC
 V6B 1N2

Project: WKM 97-06
 Comments: ATTN:S.HARRIS/J.LEHTINEN

Page Number :2-B
 Total Pages :4
 Certificate Date: 21-AUG-97
 Invoice No. :19720423
 P.O. Number :
 Account :EIA

CERTIFICATE OF ANALYSIS A9720423

SAMPLE	PREP CODE		Mn	Mo	Na	Ni	P	Pb	Sb	Sc	Sr	Ti	Tl	U	V	W	Zn
			ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm
197041	205	226	2100	2 < 0.01	1	1280	38	2	4	33 < 0.01	< 10	< 10	23	< 10	246		
197042	205	226	1650	4 < 0.01	3	840	68	6	2	19 < 0.01	< 10	< 10	13	< 10	348		
197043	205	226	3830	< 1 < 0.01	5	1510	18	2	9	25 < 0.01	< 10	< 10	28	< 10	246		
197044	205	226	1410	1 < 0.01	3	1110	24	8	3	20 < 0.01	< 10	< 10	17	< 10	186		
197045	205	226	1510	4 < 0.01	2	1170	36	8	4	24 < 0.01	< 10	< 10	20	< 10	224		
197046	205	226	2550	1 < 0.01	3	1590	6	8	6	12 < 0.01	< 10	< 10	43	< 10	114		
197047	205	226	2810	1 0.01	1	1510	38	6	6	11 < 0.01	< 10	< 10	42	< 10	206		
197048	205	226	2510	< 1 < 0.01	1	1550	2	6	7	12 < 0.01	< 10	< 10	56	< 10	86		
197049	205	226	1860	< 1 0.01	1	1310	2	12	6	25 < 0.01	< 10	< 10	39	< 10	120		
197050	205	226	2130	< 1 < 0.01	3	1370	14	16	4	28 < 0.01	< 10	< 10	22	< 10	122		
197051	205	226	1205	1 < 0.01	4	1350	14	12	6	22 < 0.01	< 10	< 10	30	< 10	128		
197052	205	226	6030	< 1 < 0.01	3	1340	2	6	9	22 < 0.01	< 10	< 10	56	< 10	46		
197053	205	226	6070	1 < 0.01	1	1390	6	6	8	28 < 0.01	< 10	< 10	58	< 10	94		
197054	205	226	4240	< 1 < 0.01	1	1600	12	12	6	34 < 0.01	< 10	< 10	40	< 10	96		
197055	205	226	6070	1 < 0.01	1	1450	6	8	8	31 < 0.01	< 10	< 10	49	< 10	64		
197056	205	226	2840	1 < 0.01	3	1450	16	14	6	28 < 0.01	< 10	< 10	27	< 10	156		
197057	205	226	3750	3 < 0.01	2	1230	16	8	7	19 < 0.01	< 10	< 10	38	< 10	70		
197058	205	226	6690	1 < 0.01	< 1	1300	10	6	7	21 < 0.01	< 10	< 10	43	< 10	104		
197059	205	226	2640	1 < 0.01	1	1300	18	12	5	21 < 0.01	< 10	< 10	27	< 10	176		
197060	205	226	2830	< 1 < 0.01	1	1650	2	2	7	22 < 0.01	< 10	< 10	55	< 10	84		
197061	205	226	1490	1 < 0.01	2	1550	50	14	2	32 < 0.01	< 10	< 10	22	< 10	266		
197062	205	226	2180	< 1 < 0.01	< 1	1490	< 2	2	6	28 < 0.01	< 10	< 10	46	< 10	82		
197063	205	226	3680	< 1 < 0.01	1	1480	10	4	6	30 < 0.01	< 10	< 10	34	< 10	142		
197064	205	226	3220	< 1 < 0.01	1	1570	12	6	4	34 < 0.01	< 10	< 10	21	< 10	224		
197065	205	226	2850	2 < 0.01	1	1430	30	8	5	29 < 0.01	< 10	< 10	29	< 10	184		
197066	205	226	3370	< 1 < 0.01	1	1150	6	< 2	7	30 < 0.01	< 10	< 10	43	< 10	100		
197067	205	226	2280	4 < 0.01	4	1320	18	10	5	25 < 0.01	< 10	< 10	32	< 10	128		
197068	205	226	3220	< 1 < 0.01	3	1550	< 2	< 2	6	23 < 0.01	< 10	< 10	42	< 10	98		
197069	205	226	3830	4 < 0.01	3	1450	20	4	9	36 < 0.01	< 10	< 10	43	< 10	618		
197070	205	226	3690	1 < 0.01	3	1410	8	8	8	24 < 0.01	< 10	< 10	41	< 10	142		
197071	205	226	2820	3 < 0.01	4	1110	24	8	7	26 < 0.01	< 10	< 10	54	< 10	284		
197072	205	226	4000	6 < 0.01	6	1030	24	6	8	22 < 0.01	< 10	< 10	45	< 10	908		
197073	205	226	2750	1 < 0.01	4	1170	6	10	8	36 < 0.01	< 10	< 10	50	< 10	86		
197074	205	226	3270	4 < 0.01	9	850	< 2	10	9	51 < 0.01	< 10	< 10	44	< 10	102		
197075	205	226	1955	2 < 0.01	< 1	1530	4	24	5	15 < 0.01	< 10	< 10	21	< 10	80		
197076	205	226	1465	143 < 0.01	1	1390	14	6	5	14 < 0.01	< 10	< 10	28	< 10	140		
197077	205	226	1580	791 < 0.01	< 1	1090	30	12	3	9 < 0.01	< 10	< 10	15	< 10	124		
197078	205	226	1300	12 < 0.01	3	1210	12	14	4	20 < 0.01	< 10	< 10	22	< 10	138		
197079	205	226	2370	1 < 0.01	1	1630	2	6	7	30 < 0.01	< 10	< 10	59	< 10	72		
197080	205	226	380	< 1 < 0.01	< 1	20	6	< 2	< 1	10 < 0.01	< 10	< 10	< 1	< 10	8		

CERTIFICATION: _____



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207 - 675 W. HASTINGS ST.
 VANCOUVER, BC
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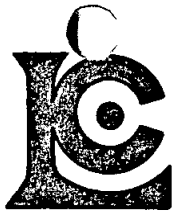
Project: WKM 97-06
 Comments: ATTN:S.HARRIS/J.LEHTINEN

Page Number : 3-A
 Total Pages : 4
 Certificate Date: 21-AUG-97
 Invoice No. : I9720423
 P.O. Number :
 Account : EIA

CERTIFICATE OF ANALYSIS A9720423

SAMPLE	PREP CODE	Au ppb FA+AA	Au FA g/t	Ag ppm	Al %	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K %	La ppm	Mg %
197081	205 226	10	-----	2.4	1.14	32	80	1.5	< 2	3.19	< 0.5	20	15	47	4.20	< 10	< 1	0.48	< 10	0.89
197082	205 226	< 5	-----	0.2	1.32	70	10	1.5	< 2	0.76	< 0.5	29	11	18	3.62	< 10	< 1	0.30	10	0.25
197083	205 226	95	-----	1.8	0.89	752	20	1.0	< 2	0.58	< 0.5	11	20	2	2.40	< 10	< 1	0.42	< 10	0.17
197084	205 226	60	-----	1.4	0.65	142	< 10	0.5	< 2	0.09	< 0.5	< 1	37	1	0.31	< 10	< 1	0.33	< 10	0.03
197085	205 226	25	-----	3.8	0.55	46	< 10	0.5	< 2	0.22	< 0.5	< 1	34	10	0.39	< 10	< 1	0.30	< 10	0.07
197086	205 226	10	-----	3.2	0.58	20	< 10	0.5	< 2	0.08	< 0.5	< 1	42	10	0.27	< 10	< 1	0.31	< 10	0.02
197087	205 226	5	-----	0.8	0.64	36	< 10	0.5	< 2	0.08	< 0.5	< 1	41	5	0.27	< 10	< 1	0.34	< 10	0.02
197088	205 226	10	-----	1.4	0.67	34	< 10	0.5	< 2	0.07	< 0.5	< 1	54	4	0.27	< 10	< 1	0.34	< 10	0.03
197089	205 226	10	-----	0.8	0.54	30	< 10	0.5	< 2	0.06	< 0.5	< 1	39	1	0.25	< 10	< 1	0.30	< 10	0.01
197090	205 226	55	-----	0.6	0.66	120	< 10	0.5	< 2	0.07	< 0.5	< 1	68	1	0.30	< 10	< 1	0.35	< 10	0.02
197091	205 226	20	-----	< 0.2	0.55	28	< 10	0.5	< 2	0.09	< 0.5	< 1	56	< 1	0.28	< 10	< 1	0.31	< 10	0.03
197092	205 226	575	-----	4.4	0.80	1295	< 10	0.5	< 2	0.40	< 0.5	6	39	8	1.03	< 10	< 1	0.48	< 10	0.13
197093	205 226	210	-----	3.0	0.66	892	< 10	0.5	< 2	0.70	< 0.5	10	29	12	2.06	< 10	< 1	0.36	< 10	0.21
197094	205 226	1620	1.58	0.2	1.09	16	30	0.5	< 2	2.16	1.0	22	28	31	5.21	< 10	< 1	0.26	< 10	0.80
197095	205 226	300	-----	0.6	1.40	62	40	0.5	< 2	1.19	1.5	22	13	47	5.85	< 10	< 1	0.47	< 10	0.57
197096	205 226	< 5	-----	0.8	1.46	114	10	2.0	< 2	0.80	< 0.5	27	17	35	7.18	< 10	< 1	0.27	< 10	0.51
197097	205 226	< 5	-----	0.2	1.07	114	30	1.0	< 2	1.85	< 0.5	20	14	62	3.89	< 10	< 1	0.53	10	0.62
197098	205 226	30	-----	0.4	1.04	282	40	0.5	< 2	0.44	1.5	23	20	30	5.34	< 10	< 1	0.54	< 10	0.29
197099	205 226	30	-----	0.8	0.92	240	50	0.5	< 2	0.31	2.0	21	20	37	4.70	< 10	< 1	0.47	< 10	0.08
197100	205 226	680	-----	7.2	0.88	2350	40	0.5	< 2	0.35	2.0	23	25	45	5.87	< 10	< 1	0.43	< 10	0.06
197101	205 226	1680	1.61	42.4	0.69	1835	50	0.5	< 2	0.25	1.5	19	35	29	4.33	< 10	< 1	0.34	< 10	0.04
197102	205 226	1365	1.23	56.8	0.87	2690	40	0.5	< 2	0.26	1.5	18	40	24	5.08	< 10	< 1	0.39	< 10	0.04
197103	205 226	2340	2.33	11.0	0.69	5370	30	0.5	< 2	0.25	2.0	23	26	52	5.00	< 10	< 1	0.40	< 10	0.06
197104	205 226	< 5	-----	3.0	1.13	232	50	1.5	< 2	0.51	0.5	23	21	78	5.26	< 10	< 1	0.57	< 10	0.29
197105	205 226	1700	1.68	50.0	0.63	2640	10	0.5	< 2	0.15	5.0	16	49	34	3.65	< 10	< 1	0.27	< 10	0.03
197106	205 226	65	-----	3.0	1.03	246	30	1.0	< 2	0.30	2.0	24	20	76	4.94	< 10	< 1	0.44	< 10	0.10
197107	205 226	< 5	-----	0.2	1.10	16	30	0.5	< 2	2.19	1.0	22	29	31	5.27	< 10	< 1	0.26	< 10	0.81
197108	205 226	10	-----	0.6	1.43	66	40	0.5	< 2	1.23	1.5	22	12	48	5.96	< 10	< 1	0.47	< 10	0.58
197109	205 226	250	-----	1.8	0.82	1385	30	1.0	< 2	2.70	1.5	17	14	22	5.23	< 10	< 1	0.40	< 10	0.81
197110	205 226	< 5	-----	< 0.2	1.21	14	170	0.5	< 2	2.02	< 0.5	18	10	26	4.96	< 10	< 1	0.37	< 10	0.66
197111	205 226	15	-----	< 0.2	0.87	32	50	0.5	< 2	0.42	1.5	19	15	13	5.49	< 10	< 1	0.50	< 10	0.39
197112	205 226	160	-----	1.0	1.32	332	20	1.0	< 2	2.51	< 0.5	21	9	35	5.28	< 10	< 1	0.39	< 10	0.56
197113	205 226	25	-----	1.0	2.34	32	160	0.5	< 2	2.89	< 0.5	19	15	27	4.70	< 10	< 1	0.39	< 10	1.20
197114	205 226	< 5	-----	< 0.2	1.78	6	190	0.5	< 2	2.20	< 0.5	23	26	64	4.81	< 10	< 1	0.51	10	0.88
197115	205 226	140	-----	0.6	0.94	120	50	0.5	< 2	2.91	< 0.5	18	12	21	4.83	< 10	< 1	0.39	< 10	0.74
197116	205 226	< 5	-----	< 0.2	1.11	12	70	0.5	< 2	0.75	< 0.5	29	17	59	5.38	< 10	< 1	0.47	< 10	0.38
197117	205 226	< 5	-----	< 0.2	0.79	32	50	0.5	< 2	0.38	< 0.5	20	15	33	4.66	< 10	< 1	0.31	< 10	0.33
197118	205 226	< 5	-----	< 0.2	0.97	30	50	0.5	< 2	0.38	1.5	23	19	33	5.44	< 10	< 1	0.42	10	0.29
197119	205 226	< 5	-----	0.2	0.87	32	20	0.5	< 2	0.46	0.5	24	18	68	5.69	< 10	< 1	0.31	10	0.22
197120	205 226	< 5	-----	< 0.2	1.01	10	10	1.0	< 2	0.51	< 0.5	18	17	60	4.75	< 10	< 1	0.29	10	0.31

CERTIFICATION: _____



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CERTIFICATE OF ANALYSIS A9720423

SAMPLE	PREP CODE	Mn ppm	Mo ppm	Na %	Ni ppm	P ppm	Pb ppm	Sb ppm	Sc ppm	Sr ppm	Ti %	Tl ppm	U ppm	V ppm	W ppm	Zn ppm
197081	205 226	2210	3 < 0.01		4	1210	8	8	8	49 < 0.01	< 10	< 10	55	< 10		76
197082	205 226	2710	4 < 0.01		3	1440	10	6	6	36 < 0.01	< 10	10	44	< 10		112
197083	205 226	790	9 < 0.01		1	680	12	8	2	28 < 0.01	< 10	< 10	12	< 10		170
197084	205 226	65	< 1 < 0.01		< 1	20	8	< 2	< 1	24 < 0.01	< 10	< 10	< 1	< 10		12
197085	205 226	185	1 < 0.01		< 1	10	10	4	< 1	24 < 0.01	< 10	< 10	< 1	< 10		18
197086	205 226	35	< 1 < 0.01		< 1	10	10	4	< 1	24 < 0.01	< 10	< 10	< 1	< 10		14
197087	205 226	35	< 1 < 0.01		< 1	10	12	2	< 1	25 < 0.01	< 10	< 10	< 1	< 10		14
197088	205 226	30	< 1 < 0.01		< 1	< 10	8	< 2	< 1	25 < 0.01	< 10	< 10	< 1	< 10		12
197089	205 226	30	< 1 < 0.01		< 1	< 10	10	2	< 1	22 < 0.01	< 10	< 10	< 1	< 10		8
197090	205 226	25	< 1 < 0.01		< 1	< 10	14	< 2	< 1	24 < 0.01	< 10	< 10	< 1	< 10		18
197091	205 226	80	< 1 < 0.01		< 1	< 10	2	< 2	< 1	21 < 0.01	< 10	< 10	< 1	< 10		2
197092	205 226	545	6 < 0.01		3	220	10	6	2	21 < 0.01	< 10	20	12	< 10		74
197093	205 226	790	3 < 0.01		5	160	12	8	5	35 < 0.01	< 10	10	15	< 10		106
197094	205 226	3410	< 1 < 0.01		9	920	62	2	13	20 < 0.01	< 10	< 10	97	< 10		312
197095	205 226	3730	< 1 < 0.01		3	1210	56	6	7	18 < 0.01	< 10	< 10	63	< 10		368
197096	205 226	9440	< 1 < 0.01		11	820	6	8	13	60 < 0.01	< 10	< 10	90	< 10		38
197097	205 226	2850	1 < 0.01		8	910	2	16	7	46 < 0.01	< 10	< 10	35	< 10		58
197098	205 226	3810	1 < 0.01		10	950	30	8	9	8 < 0.01	< 10	< 10	53	< 10		370
197099	205 226	1395	3 < 0.01		7	980	40	12	5	8 < 0.01	< 10	< 10	34	< 10		376
197100	205 226	2890	2 < 0.01		9	1120	34	24	4	8 < 0.01	< 10	< 10	26	< 10		258
197101	205 226	1560	3 < 0.01		7	830	36	60	3	7 < 0.01	< 10	< 10	19	< 10		288
197102	205 226	635	4 < 0.01		8	910	56	64	3	9 < 0.01	< 10	< 10	20	< 10		260
197103	205 226	1735	3 < 0.01		9	810	54	22	5	7 < 0.01	< 10	< 10	26	< 10		428
197104	205 226	5060	3 < 0.01		9	1240	26	20	12	11 < 0.01	< 10	< 10	60	< 10		244
197105	205 226	310	4 < 0.01		8	440	202	34	2	6 < 0.01	< 10	< 10	25	< 10		656
197106	205 226	1920	8 < 0.01		10	610	84	22	6	18 < 0.01	< 10	< 10	45	< 10		460
197107	205 226	3450	1 < 0.01		10	910	64	2	13	20 < 0.01	< 10	< 10	97	< 10		314
197108	205 226	3800	< 1 < 0.01		1	1200	54	4	7	18 < 0.01	< 10	< 10	64	< 10		380
197109	205 226	4330	< 1 < 0.01		< 1	1320	28	8	8	21 < 0.01	< 10	< 10	32	< 10		266
197110	205 226	2200	< 1 < 0.01		< 1	1210	2	2	8	45 < 0.01	< 10	< 10	66	< 10		116
197111	205 226	4690	1 < 0.01		7	940	20	2	12	8 < 0.01	< 10	< 10	60	< 10		344
197112	205 226	2050	< 1 < 0.01		3	1480	14	6	7	47 < 0.01	< 10	< 10	52	< 10		124
197113	205 226	1860	< 1 0.05		1	1430	10	2	6	78 0.03	< 10	< 10	81	< 10		96
197114	205 226	1685	< 1 < 0.01		10	1320	6	< 2	7	47 < 0.01	< 10	< 10	78	< 10		104
197115	205 226	3190	< 1 < 0.01		< 1	1470	12	6	6	43 < 0.01	< 10	< 10	32	< 10		78
197116	205 226	2010	2 < 0.01		12	860	2	< 2	4	12 < 0.01	< 10	< 10	47	< 10		92
197117	205 226	2360	< 1 < 0.01		9	1010	6	4	6	9 < 0.01	< 10	< 10	44	< 10		78
197118	205 226	2870	< 1 < 0.01		8	900	6	6	8	8 < 0.01	< 10	< 10	51	< 10		314
197119	205 226	2780	1 < 0.01		8	1210	6	14	8	10 < 0.01	< 10	< 10	45	< 10		98
197120	205 226	2560	< 1 < 0.01		6	1190	8	14	11	11 < 0.01	< 10	< 10	69	< 10		66

CERTIFICATION:

Heidi Buchler



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers
 212 Brooksbank Ave., North Vancouver
 British Columbia, Canada V7J 2C1
 PHONE: 604-984-0221 FAX: 604-984-0218

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
 VANCOUVER, BC
 V6B 1N2

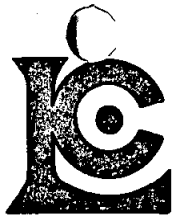
Project: WKM 97-06
 Comments: ATTN:S.HARRIS/J.LEHTINEN

Page Number: 4-A
 Total Pages: 4
 Certificate Date: 21-AUG-97
 Invoice No.: 19720423
 P.O. Number:
 Account: EIA

CERTIFICATE OF ANALYSIS A9720423

SAMPLE	PREP CODE	Au ppb FA+AA	Au FA g/t	Ag ppm	Al %	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K %	La ppm	Mg %
197121	205 226	30	-----	0.4	0.83	188	50	0.5	< 2	0.22	< 0.5	11	24	35	2.84	< 10	< 1	0.32	10	0.04
197122	205 226	10	-----	0.2	0.80	20	30	0.5	< 2	0.26	< 0.5	19	10	54	5.89	< 10	< 1	0.18	10	0.08
197123	205 226	< 5	-----	< 0.2	0.72	22	30	0.5	< 2	0.19	< 0.5	18	12	22	7.06	< 10	< 1	0.20	< 10	0.13
197124	205 226	30	-----	< 0.2	0.53	12	10	< 0.5	< 2	0.40	< 0.5	13	19	9	3.66	< 10	< 1	0.18	< 10	0.21
197125	205 226	30	-----	0.6	0.75	94	40	0.5	< 2	1.13	< 0.5	31	11	76	5.80	< 10	< 1	0.18	< 10	0.50
197126	205 226	< 5	-----	< 0.2	1.29	8	40	< 0.5	< 2	2.39	< 0.5	23	22	26	5.92	< 10	< 1	0.19	< 10	0.91
197127	205 226	165	-----	0.8	0.82	1890	50	0.5	< 2	2.71	< 0.5	25	18	17	5.42	< 10	< 1	0.29	< 10	0.82
197128	205 226	25	-----	13.4	7.72	32	190	< 0.5	< 2	2.78	16.0	25	28	4060	6.97	10	< 1	1.86	< 10	2.05
197129	205 226	< 5	-----	< 0.2	2.91	16	90	< 0.5	< 2	1.61	< 0.5	16	28	50	4.81	< 10	< 1	0.28	< 10	1.00

CERTIFICATION: _____



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver
 British Columbia, Canada V7J 2C1
 PHONE: 604-984-0221 FAX: 604-984-0218

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
 VANCOUVER, BC
 V6B 1N2

Project: WKM 97-06
 Comments: ATTN:S.HARRIS/J.LEHTINEN

Page No. : 4-B
 Total Pages : 4
 Certificate Date: 21-AUG-97
 Invoice No. : I9720423
 P.O. Number :
 Account : EIA

CERTIFICATE OF ANALYSIS A9720423

SAMPLE	PREP CODE		Mn	Mo	Na	Ni	P	Pb	Sb	Sc	Sr	Ti	Tl	U	V	W	Zn
			ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
197121	205	226	690	1	0.01	3	500	12	14	1	14	< 0.01	< 10	< 10	9	< 10	98
197122	205	226	585	1	0.08	5	670	8	16	2	27	< 0.01	< 10	< 10	37	< 10	60
197123	205	226	920	1	0.04	8	220	8	4	4	18	< 0.01	< 10	< 10	27	< 10	38
197124	205	226	1040	1	0.04	3	170	6	< 2	5	15	< 0.01	< 10	< 10	31	< 10	30
197125	205	226	1500	< 1	0.03	10	790	12	16	7	27	< 0.01	< 10	< 10	40	< 10	84
197126	205	226	1220	1	0.04	5	870	8	< 2	4	51	< 0.01	< 10	< 10	30	< 10	68
197127	205	226	1870	1	< 0.01	4	1280	20	6	6	36	< 0.01	< 10	< 10	40	< 10	90
197128	205	226	2250	2	0.42	< 1	1060	490	4	7	400	0.15	< 10	< 10	152	< 10	1610
197129	205	226	645	< 1	0.18	< 1	2150	8	< 2	5	334	0.10	< 10	< 10	91	< 10	90

CERTIFICATION: _____

APPENDIX E

GEOLOGIST'S AND ENGINEER'S CERTIFICATES

GEOLOGIST'S CERTIFICATE

I, Jim Lehtinen, of 4317 Briardale Road, Royston in the Province of British Columbia, DO HEREBY CERTIFY:

1. THAT I am a Consulting Geologist with Equity Engineering Ltd. with offices at Suite 207, 675 West Hastings Street, Vancouver, British Columbia.
2. THAT I am a graduate of the University of British Columbia with a Bachelor of Science degree in Geology.
3. THAT I am a Professional Geoscientist registered in good standing with the Association of Professional Engineers and Geoscientists of the Province of British Columbia.
4. THAT this report is based on a diamond drilling program I supervised in March and April of 1997, and on publicly available reports.

DATED at Vancouver, British Columbia, this ____ day of October, 1997.

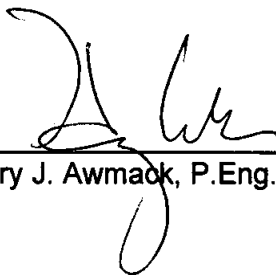
Jim Lehtinen, P.Geol.

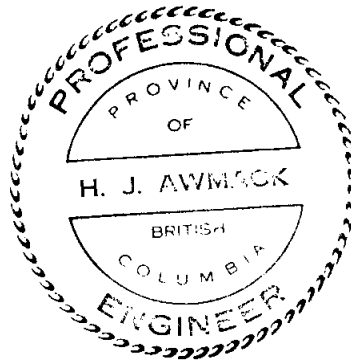
ENGINEER'S CERTIFICATE

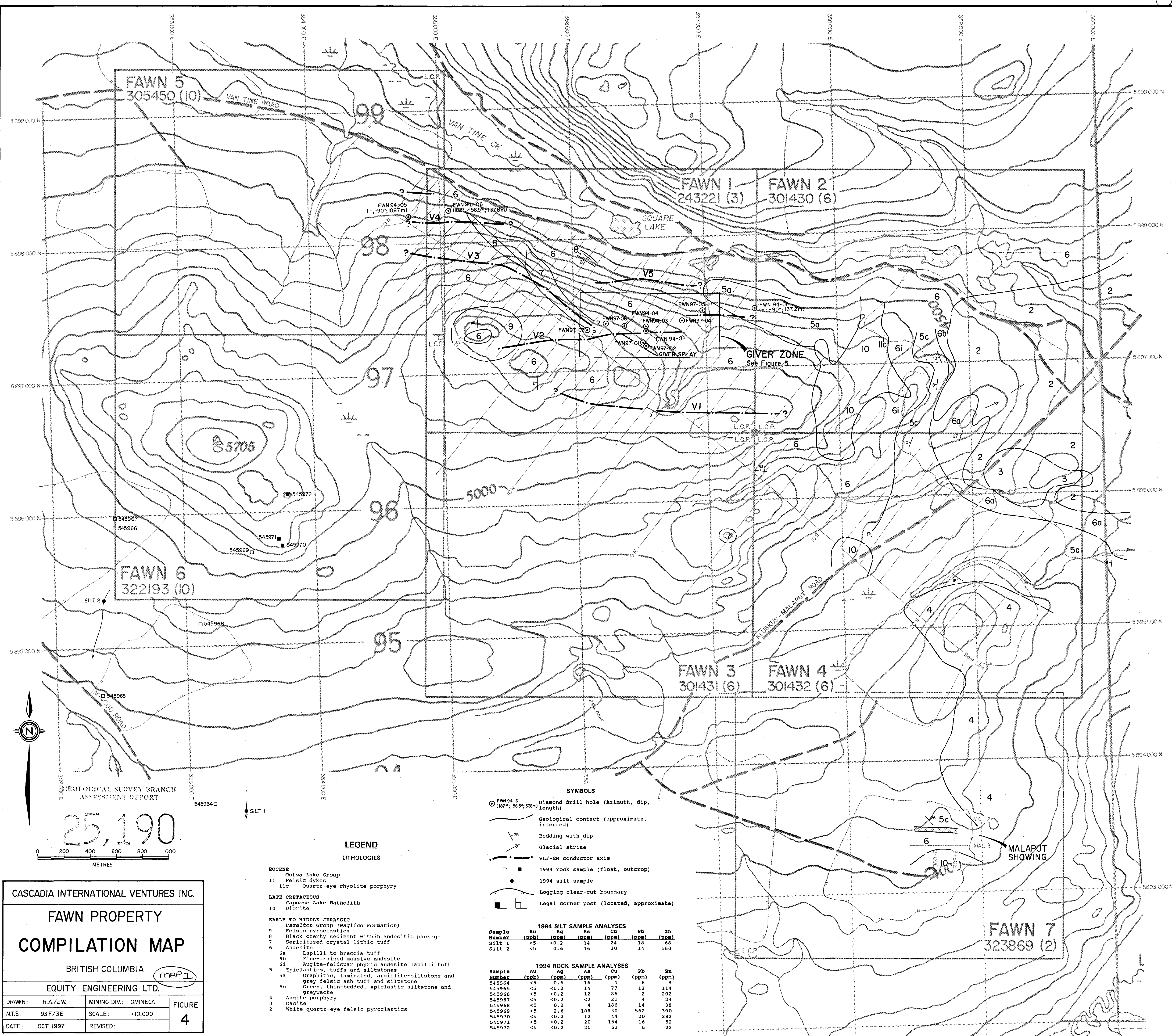
I, Henry J. Awmack, of 1735 Larch Street, Vancouver, in the Province of British Columbia, DO HEREBY CERTIFY:

1. THAT I am a Consulting Geological Engineer with offices at Suite 207, 675 West Hastings Street, Vancouver, British Columbia.
2. THAT I am a graduate of the University of British Columbia with an Honours Bachelor of Applied Science degree in Geological Engineering.
3. THAT I am a Professional Engineer registered in good standing with the Association of Professional Engineers and Geoscientists of the Province of British Columbia.
4. THAT this report is based on fieldwork carried out under my direction during March and April of 1997, and on publicly available reports. I have examined the property in the field.

DATED at Vancouver, British Columbia, this 20th day of October, 1997.


Henry J. Awmack, P.Eng.





FAWN 5
305450 (10)

FAWN 1
243221 (3)

FAWN 2
301430 (6)

FAWN 6
322193 (10)

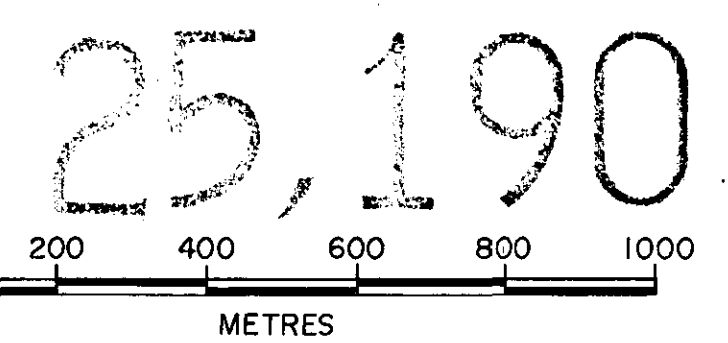
FAWN 3
301431 (6)

FAWN 4
301432 (6)

FAWN 7
323869 (2)



GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT



LEGEND
LITHOLOGIES

- EOCENE**
Ootsa Lake Group
11 Felsic dykes
11c Quartz-eye rhyolite porphyry
- LATE CRETACEOUS**
Capoose Lake Batholith
10 Diorite
- EARLY TO MIDDLE JURASSIC**
Hazelton Group (Naglico Formation)
9 Felsic pyroclastics
8 Black cherty sediment within andesitic package
7 Sericitized crystal lithic tuff
6 Andesite
6a Lapilli to breccia tuff
6b Fine-grained massive andesite
6i Augite-feldspar phryc andesite lapilli tuff
5 Epilastics, tufts and siltstones
5a Graphitic, laminated, argillite-siltstone and grey felsic ash tuff and siltstone
5c Green, thin-bedded, epiclastic siltstone and greywacke
4 Augite porphyry
3 Dacite
2 White quartz-eye felsic pyroclastics

- SYMBOLS**
- FWN 94-8 (182°, -56.5°, 137.8m) Diamond drill hole (Azimuth, dip, length)
 - Geological contact (approximate, inferred)
 - ↘ 25° Bedding with dip
 - Glacial striae
 - VLF-EM conductor axis
 - 1994 rock sample (float, outcrop)
 - 1994 silt sample
 - Logging clear-cut boundary
 - Legal corner post (located, approximate)

1994 SILT SAMPLE ANALYSES

Sample Number	Au (ppb)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Zn (ppm)
Silt 1	<5	<0.2	14	24	18	68
Silt 2	<5	0.6	16	30	14	160

1994 ROCK SAMPLE ANALYSES

Sample Number	Au (ppb)	Ag (ppm)	As (ppm)	Cu (ppm)	Pb (ppm)	Zn (ppm)
545964	<5	0.6	14	4	6	8
545965	<5	<0.2	14	77	12	114
545966	<5	<0.2	12	86	2	202
545967	<5	<0.2	<2	21	4	24
545968	<5	0.2	4	186	14	38
545969	<5	2.6	108	30	562	390
545970	<5	<0.2	12	44	20	282
545971	<5	<0.2	20	154	16	52
545972	<5	<0.2	20	62	6	22

CASCADIA INTERNATIONAL VENTURES INC.

FAWN PROPERTY

COMPILATION MAP

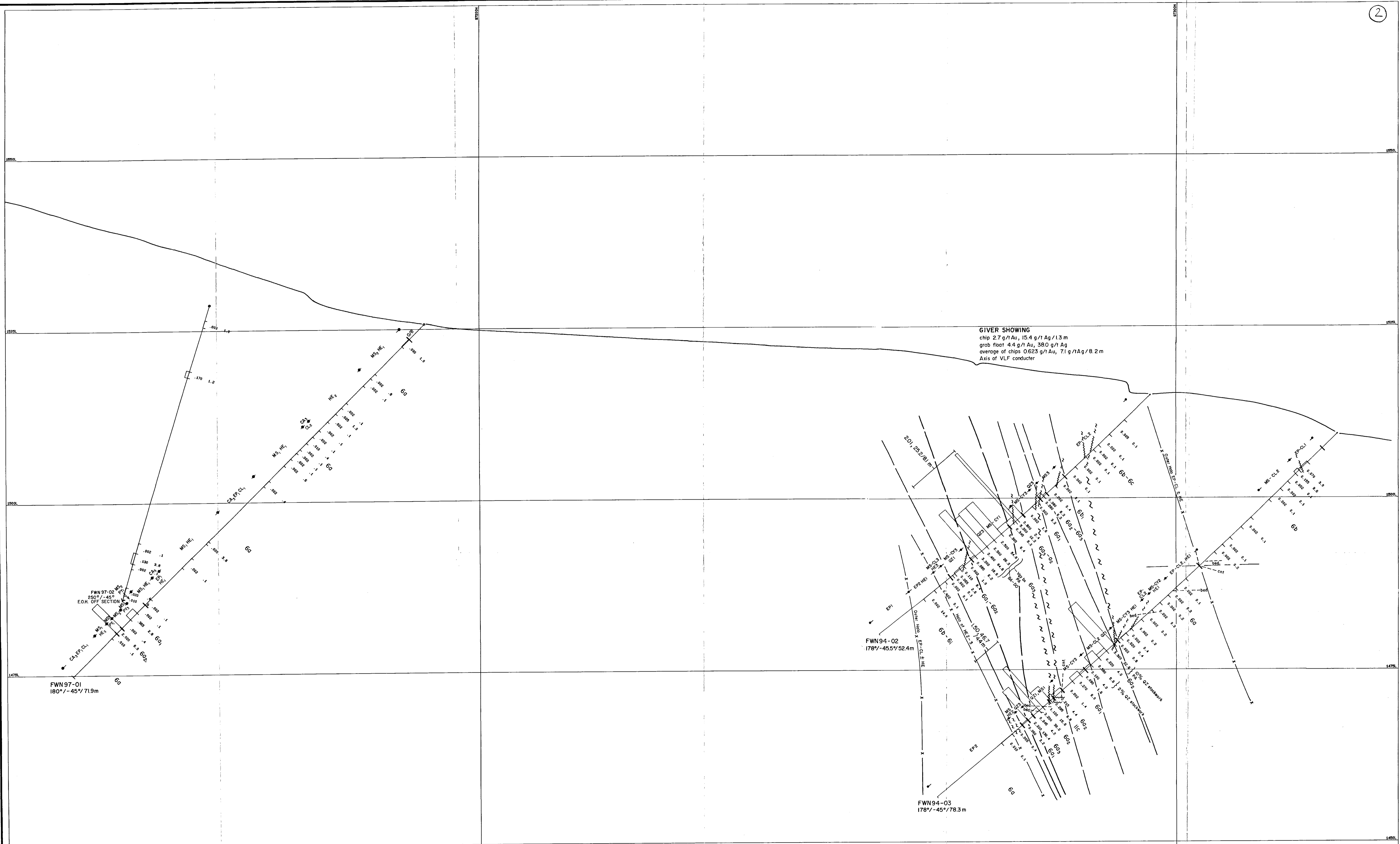
BRITISH COLUMBIA

EQUITY ENGINEERING LTD.

DRAWN: H.A./J.W. MINING DIV: OMINECA FIGURE 4

N.T.S.: 93 F/3E SCALE: 1:10,000

DATE: OCT. 1997 REVISED:



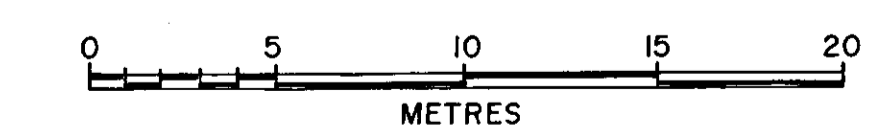
GIVER SHOWING
 chip 2.7 g/t Au, 15.4 g/t Ag / 1.3 m
 grab float 4.4 g/t Au, 380 g/t Ag
 average of chips 0.623 g/t Au, 71 g/t Ag / 8.2 m
 Axis of VLF conductor

- LITHOLOGIES**
- Eocene
 - Ootaa Lake Group
 - 11c Quartz-feldspar rhyolite porphyry
 - EARLY TO MIDDLE JURASSIC
 - Revelon Group (Magnifico Formation)
 - 6 Andesites
 - 6a Lapilli to breccia tuff
 - 6a strong sericite clay altered
 - 6a strong sericite-clay altered with chalcodony stockwork
 - 6a vuggy chalcodony breccia
 - 6a strong sericite-clay altered andesite with jasper matrix crackle breccia
 - 6a chlorite pyrite breccia
 - 6a intense oligoclase-sericite-biotite-pyrite altered
 - 6b Fine-grained massive andesite
 - 6b strong sericite-clay altered
 - 6b strong sericite-clay altered with chalcodony stockwork
 - 6b chlorite pyrite breccia
 - 6b plagioclase (oligoclase)-muscovite-pyrite chalcodony breccia
 - 6b chalcodony-epidote breccia
 - 6b intense oligoclase-sericite-biotite-pyrite altered
 - 6c Feldspar porphyry
 - 6c strong sericite-clay altered
 - 6c strong sericite-clay altered with chalcodony stockwork
 - 6c Maroon feldspar porphyry
 - 6c Amygdaloidal andesite
 - 6c Augite-feldspar phyric andesite lapilli tuff
 - 6c Maroon massive-bedded siltstone-coarse-grained sandstone
 - 5 Epitaxial, Tuffs and Siltstones
 - 5a Fine (sharp) laminated argillite-siltstone and gray felsic ash tuff and siltstone
 - 5b Fine laminated argillite-siltstone and gray felsic ash tuff and siltstone

- ALTERATION & MINERALIZATION**
- AS arsenopyrite
 - CL chlorite
 - EP epidote
 - MG magnetite
 - PO pyrrhotite
 - QZ chalcedony
 - BA barite
 - CL clay
 - GL galena
 - MS sericite
 - PR pyrrhotite
 - SP sphalerite
 - BI biotite
 - DO dolomite
 - HE hematite
 - PI plagioclase
 - PY pyrite
- ALTERATION INTENSITY**
- 1 weak
 - 2 moderate
 - 3 strong
- SYMBOLS**
- 180°/-45°/71.3m Diamond drill hole (azimuth, dip, length)
 - 3.200 39.0 Assay interval: Au (g/t) Ag (g/t)
 - 2.02, 252/8.1m Composite assay interval: Au (g/t) Ag (g/t) / Length (m)
 - ~ ~ ~ Fault/Shear
 - Lithological contact
 - • • Alteration contact
 - CL Alteration interval
- Histograms to left of drill holes correspond to gold grades

GEOLOGICAL SURVEY BRANCH
 ASSESSMENT REPORT

25,190

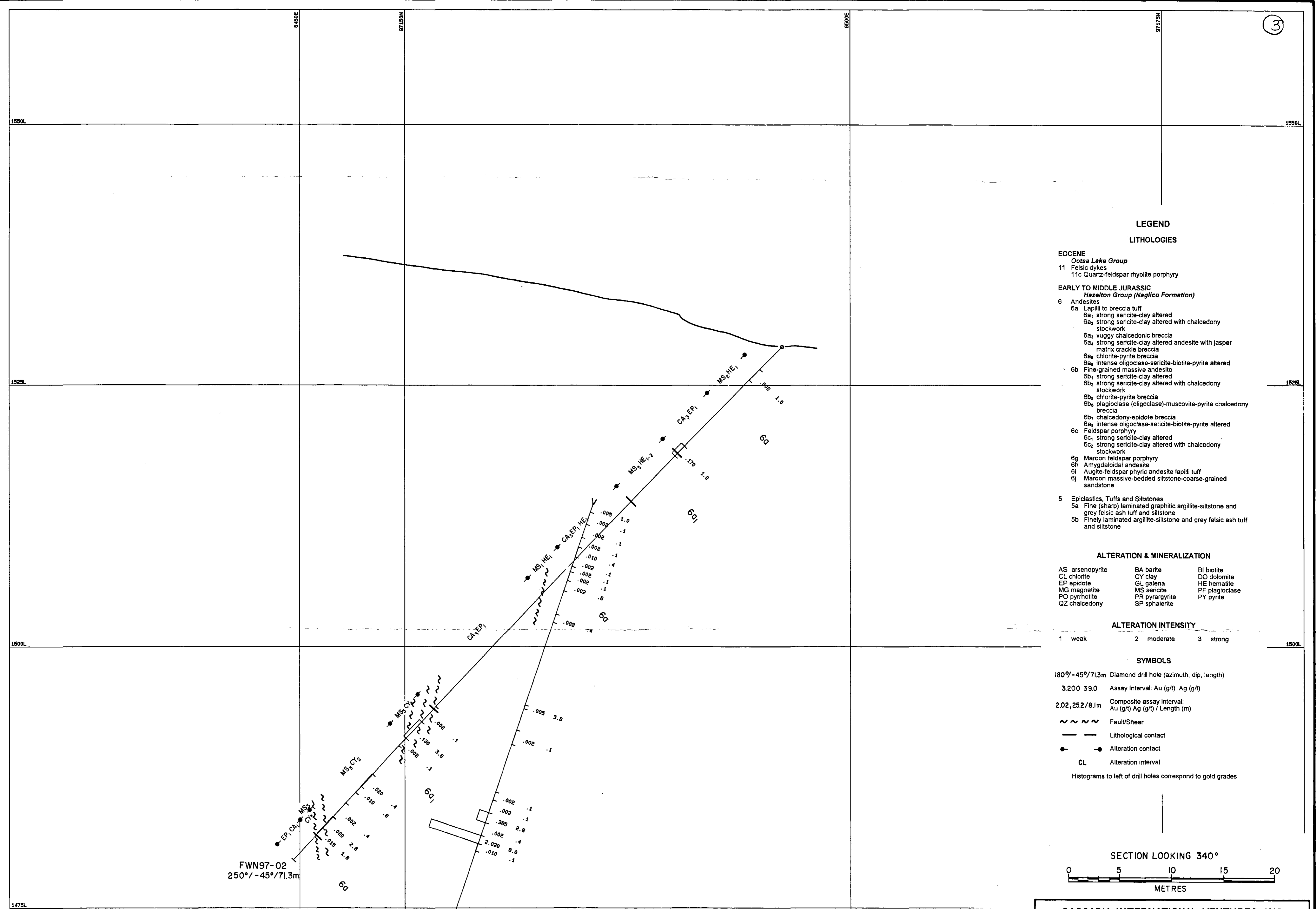


CASCADIA INTERNATIONAL VENTURES INC.

FAWN PROPERTY
FWN94-02,94-03,97-01,97-02
CROSS-SECTION
GIVER ZONE

NOTE: SECTION LOOKING TOWARDS 268°

Date	OCT. 1997	Scale	1:200	FIGURE	6
UTM Zone ID (NAD-27)	NTS	Mining District	OMINECA		
NTS	93 F / 3E	State/Province	B.C.		



LEGEND
LITHOLOGIES

- EOCENE**
Ootsa Lake Group
11 Felsic dykes
11c Quartz-feldspar rhyolite porphyry
- EARLY TO MIDDLE JURASSIC**
Hazelton Group (Naglico Formation)
- 6 Andesites**
6a Lapilli to breccia tuff
6a₁ strong sericite-clay altered
6a₂ strong sericite-clay altered with chalcodony stockwork
6a₃ vuggy chalcodonic breccia
6a₄ strong sericite-clay altered andesite with jasper matrix crackle breccia
6a₅ chlorite-pyrite breccia
6a₆ intense oligoclase-sericite-biotite-pyrite altered
6b Fine-grained massive andesite
6b₁ strong sericite-clay altered
6b₂ strong sericite-clay altered with chalcodony stockwork
6b₃ chlorite-pyrite breccia
6b₄ plagioclase (oligoclase)-muscovite-pyrite chalcodony breccia
6b₅ chalcodony-epidote breccia
6b₆ intense oligoclase-sericite-biotite-pyrite altered
6c Feldspar porphyry
6c₁ strong sericite-clay altered
6c₂ strong sericite-clay altered with chalcodony stockwork
6g Maroon feldspar porphyry
6h Amygdaloidal andesite
6i Augite-feldspar phytic andesite lapilli tuff
6j Maroon massive-bedded siltstone-coarse-grained sandstone
- 5 Epiclastics, Tuffs and Siltstones**
5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone
5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone

ALTERATION & MINERALIZATION

- | | | |
|-----------------|-----------------|----------------|
| AS arsenopyrite | BA barite | BI biotite |
| CL chlorite | CY clay | DO dolomite |
| EP epidote | GL galena | HE hematite |
| MG magnetite | MS sericite | PF plagioclase |
| PO pyrrhotite | PR pyrrargyrite | PY pyrite |
| QZ chalcodony | SP sphalerite | |

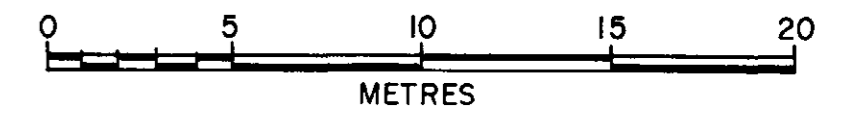
ALTERATION INTENSITY

- 1 weak 2 moderate 3 strong

SYMBOLS

- 180°/-45°/71.3m Diamond drill hole (azimuth, dip, length)
3.200 39.0 Assay interval: Au (g/t) Ag (g/t)
2.02, 252/8.1m Composite assay interval: Au (g/t) Ag (g/t) / Length (m)
- ~ ~ ~ ~ Fault/Shear
— — — Lithological contact
● — ● Alteration contact
CL Alteration interval
- Histograms to left of drill holes correspond to gold grades

SECTION LOOKING 340°



GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT

25,190

FWN97-02
250°/-45°/71.3m

FWN 97-01
180°/-45°/71.9m

CASCADIA INTERNATIONAL VENTURES INC.

FAWN PROPERTY
FWN 97-02 (MAP 3)
CROSS-SECTION
GIVER ZONE

EQUITY	Date	OCT. 1997	Scale	1: 200	FIGURE 7
	U.T.M. Zone	10 (NAD-27)	Mining District	OMINECA	
	N.T.S.	93F/3E	State/Province	B. C.	

LEGEND
LITHOLOGIES

- EOCENE**
Ootsa Lake Group
11 Felsic dykes
11c Quartz-feldspar rhyolite porphyry
- EARLY TO MIDDLE JURASSIC**
Hazelton Group (Naglico Formation)
6 Andesites
6a Lapilli to breccia tuff
6a₁ strong sericite-clay altered
6a₂ strong sericite-clay altered with chalcedony stockwork
6a₃ vuggy chalcidonic breccia
6a₄ strong sericite-clay altered andesite with jasper matrix crackle breccia
6a₅ chlorite-pyrite breccia
6a₆ intense oligoclase-sericite-biotite-pyrite altered
6b Fine-grained massive andesite
6b₁ strong sericite-clay altered
6b₂ strong sericite-clay altered with chalcedony stockwork
6b₃ chlorite-pyrite breccia
6b₄ plagioclase (oligoclase)-muscovite-pyrite chalcedony breccia
6b₅ chalcedony-epidote breccia
6a₆ intense oligoclase-sericite-biotite-pyrite altered
6c Feldspar porphyry
6c₁ strong sericite-clay altered
6c₂ strong sericite-clay altered with chalcedony stockwork
6g Maroon feldspar porphyry
6h Amygdaloidal andesite
6i Augite-feldspar phyric andesite lapilli tuff
6j Maroon massive-bedded siltstone-coarse-grained sandstone
- 5 Epiclastics, Tuffs and Siltstones
5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone
5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone

ALTERATION & MINERALIZATION

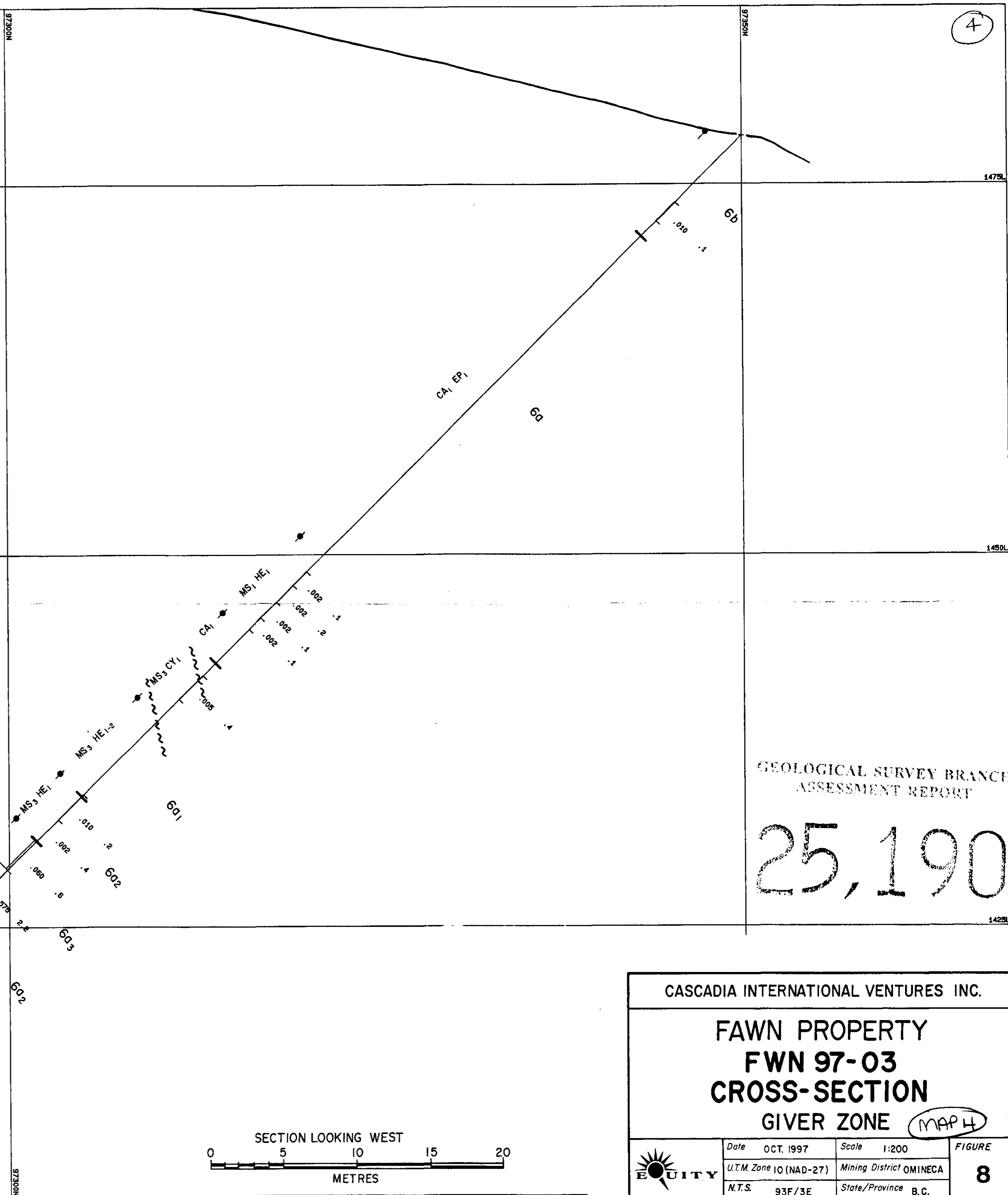
- | | | |
|-----------------|-----------------|----------------|
| AS arsenopyrite | BA barite | BI biotite |
| CL chlorite | CY clay | DO dolomite |
| EP epidote | GL galena | HE hematite |
| MG magnetite | MS sericite | PF plagioclase |
| PO pyrrhotite | PR pyrrargyrite | PY pyrite |
| QZ chalcedony | SP sphalerite | |

ALTERATION INTENSITY

- 1 weak 2 moderate 3 strong

SYMBOLS

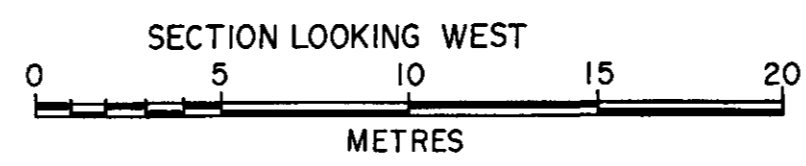
- 180°/-45°/71.3m Diamond drill hole (azimuth, dip, length)
3.200 39.00 Assay interval: Au (g/t) Ag (g/t)
2.02, 25.2/8.1m Composite assay interval:
Au (g/t) Ag (g/t) / Length (m)
- ~ ~ ~ ~ Fault/Shear
— — — — Lithological contact
● — — — — Alteration contact
CL Alteration interval
- Histograms to left of drill holes correspond to gold grades



GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT

25,190

FWN 97-03
180°/-45°/89.9m



CASCADIA INTERNATIONAL VENTURES INC.				
FAWN PROPERTY				
FWN 97-03				
CROSS-SECTION				
GIVER ZONE MAP 4				
	Date	OCT. 1997	Scale	1:200
	U.T.M. Zone	10 (NAD-27)	Mining District	OMINECA
	N.T.S.	93F/3E	State/Province	B.C.
			FIGURE	8

LEGEND
LITHOLOGIES

- EOCENE**
Ootsa Lake Group
11 Felsic dykes
11c Quartz-feldspar rhyolite porphyry
- EARLY TO MIDDLE JURASSIC**
Hazelton Group (Naglico Formation)
- 6 Andesites
6a Lapilli to breccia tuff
6a₁ strong sericite-clay altered
6a₂ strong sericite-clay altered with chalcedony stockwork
6a₃ vuggy chalcedonic breccia
6a₄ strong sericite-clay altered andesite with jasper matrix crackle breccia
6a₅ chlorite-pyrite breccia
6a₆ intense oligoclase-sericite-biotite-pyrite altered
6b Fine-grained massive andesite
6b₁ strong sericite-clay altered
6b₂ strong sericite-clay altered with chalcedony stockwork
6b₃ chlorite-pyrite breccia
6b₄ plagioclase (oligoclase)-muscovite-pyrite chalcedony breccia
6b₇ chalcedony-epidote breccia
6a₄ intense oligoclase-sericite-biotite-pyrite altered
6c Feldspar porphyry
6c₁ strong sericite-clay altered
6c₂ strong sericite-clay altered with chalcedony stockwork
6g Maroon feldspar porphyry
6h Amygdaloidal andesite
6i Augite-feldspar phyric andesite lapilli tuff
6j Maroon massive-bedded siltstone-coarse-grained sandstone
- 5 Epiclastics, Tufts and Siltstones
5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone
5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone

ALTERATION & MINERALIZATION

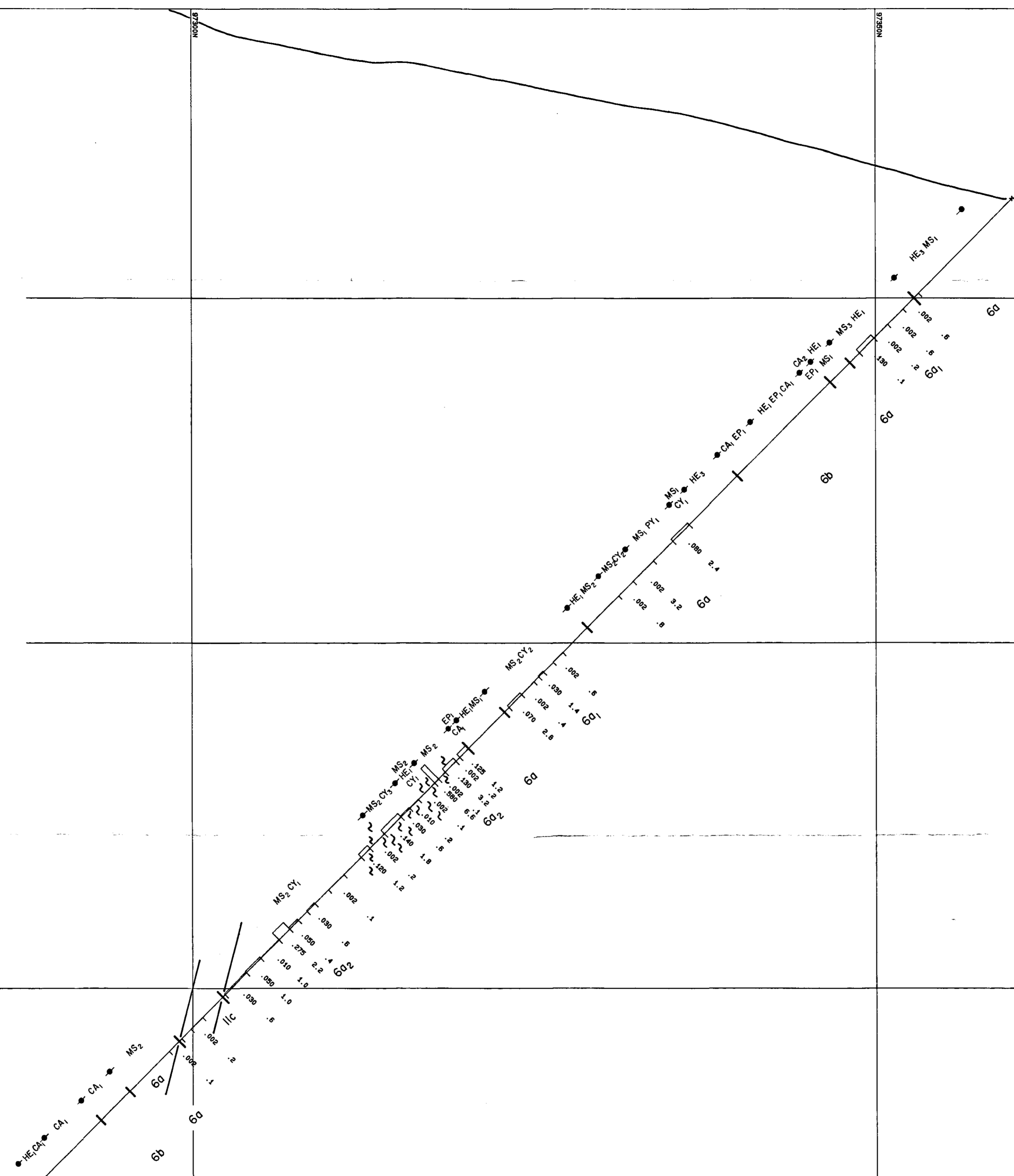
- | | | |
|-----------------|-----------------|----------------|
| AS arsenopyrite | BA barite | BI biotite |
| CL chlorite | CY clay | DO dolomite |
| EP epidote | GL galena | HE hematite |
| MG magnetite | MS sericite | PF plagioclase |
| PO pyrrhotite | PR pyrrargyrite | PY pyrite |
| QZ chalcedony | SP sphalerite | |

ALTERATION INTENSITY

- 1 weak 2 moderate 3 strong

SYMBOLS

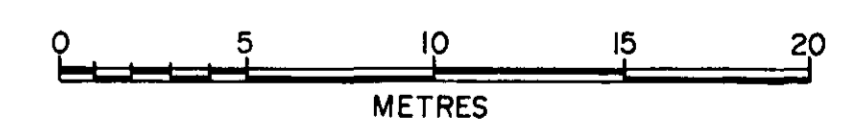
- 180°/-45°/71.3m Diamond drill hole (azimuth, dip, length)
3.200 39.0 Assay Interval: Au (g/t) Ag (g/t)
2.02, 25.2/8.1m Composite assay interval:
Au (g/t) Ag (g/t) / Length (m)
- ~ ~ ~ Fault/Shear
— Lithological contact
● Alteration contact
CL Alteration interval
- Histograms to left of drill holes correspond to gold grades



GEOLOGICAL SURVEY BRANCH
ASSESSMENT REPORT

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SECTION LOOKING WEST



CASCADIA INTERNATIONAL VENTURES INC.

FAWN PROPERTY
FWN 97-04
CROSS-SECTION
GIVER ZONE MAPS

EQUITY	Date	OCT. 1997	Scale	1:200	FIGURE 9
	U.T.M. Zone	10 (NAD-27)	Mining District	OMINECA	
	N.T.S.	93F/3E	State/Province	B.C.	

1450L

1450L

1425L

1425L

1440L

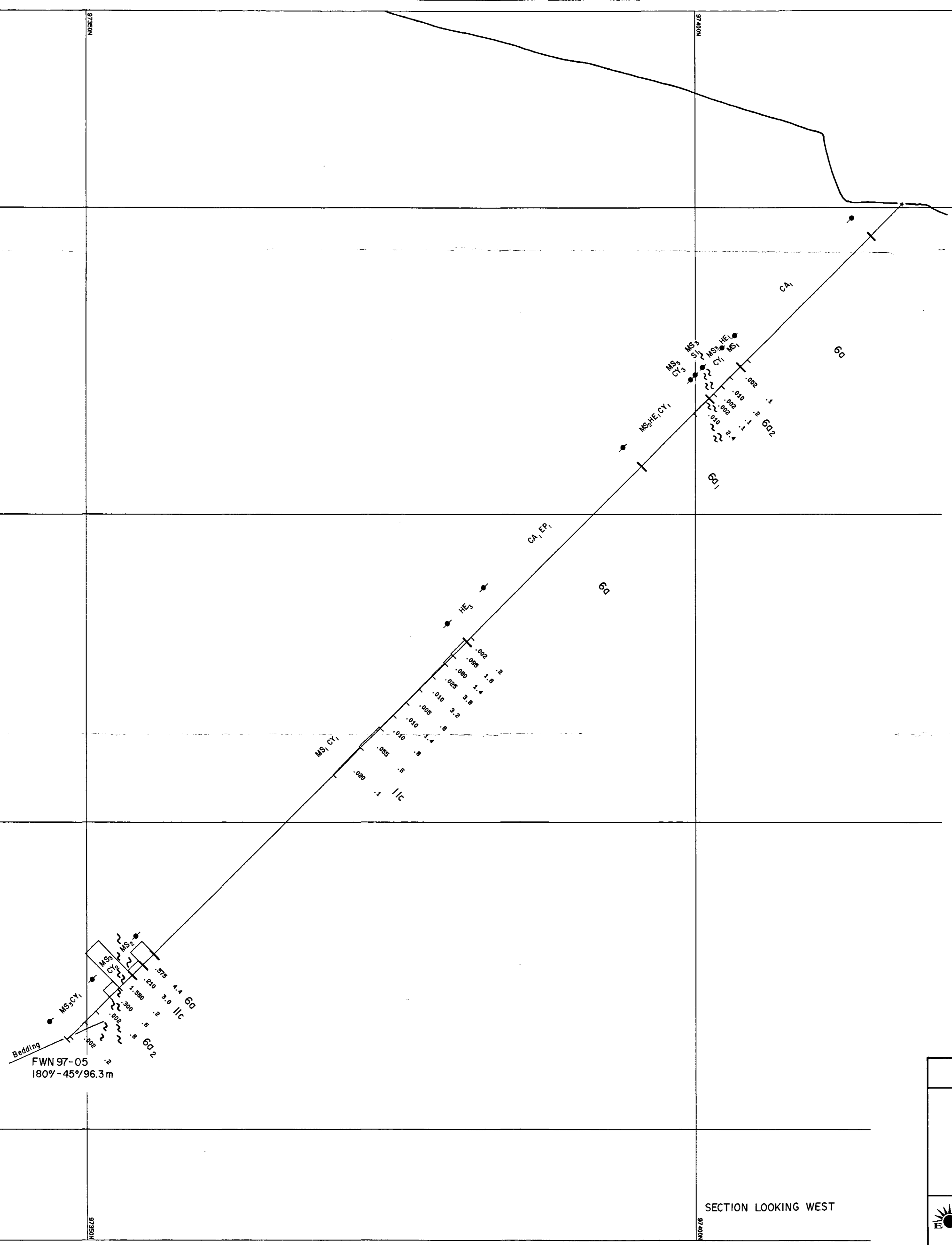
1440L

1375E

1395E

1385E

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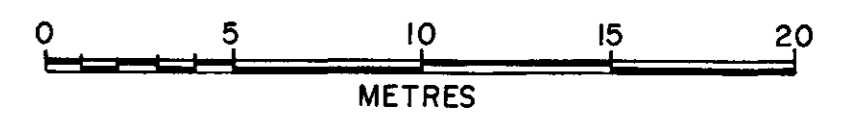


- LEGEND**
LITHOLOGIES
- EOCENE**
Ootsa Lake Group
11 Felsic dykes
11c Quartz-feldspar rhyolite porphyry
- EARLY TO MIDDLE JURASSIC**
Hazelton Group (Naglico Formation)
6 Andesites
6a Lapilli to breccia tuff
6a1 strong sericite-clay altered
6a2 strong sericite-clay altered with chalcedony stockwork
6a3 vuggy chalcedonic breccia
6a4 strong sericite-clay altered andesite with jasper matrix crackle breccia
6a5 chlorite-pyrite breccia
6a6 intense oligoclase-sericite-biotite-pyrite altered
6b Fine-grained massive andesite
6b1 strong sericite-clay altered
6b2 strong sericite-clay altered with chalcedony stockwork
6b3 chlorite-pyrite breccia
6b4 plagioclase (oligoclase)-muscovite-pyrite chalcedony breccia
6b5 chalcedony-epidote breccia
6b6 intense oligoclase-sericite-biotite-pyrite altered
6c Feldspar porphyry
6c1 strong sericite-clay altered
6c2 strong sericite-clay altered with chalcedony stockwork
6g Maroon feldspar porphyry
6h Amygdaloidal andesite
6i Augite-feldspar phytic andesite lapilli tuff
6j Maroon massive-bedded siltstone-coarse-grained sandstone
- 5 Epiclastics, Tuffs and Siltstones
5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone
5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone

- ALTERATION & MINERALIZATION**
- | | | |
|-----------------|-----------------|----------------|
| AS arsenopyrite | BA barite | BI biotite |
| CL chlorite | CY clay | DO dolomite |
| EP epidote | GA galena | HE hematite |
| MG magnetite | MS sericite | PF plagioclase |
| PO pyrrhotite | PR pyrrargyrite | PY pyrite |
| QZ chalcedony | SP sphalerite | |

- ALTERATION INTENSITY**
- 1 weak 2 moderate 3 strong

- SYMBOLS**
- 180° - 45° / 71.3m Diamond drill hole (azimuth, dip, length)
3.200 39.0 Assay Interval: Au (g/t) Ag (g/t)
2.02, 25.2 / 8.1m Composite assay interval: Au (g/t) Ag (g/t) / Length (m)
~ ~ ~ ~ Fault/Shear
— — — — Lithological contact
● ● ● ● Alteration contact
CL Alteration interval
Histograms to left of drill holes correspond to gold grades



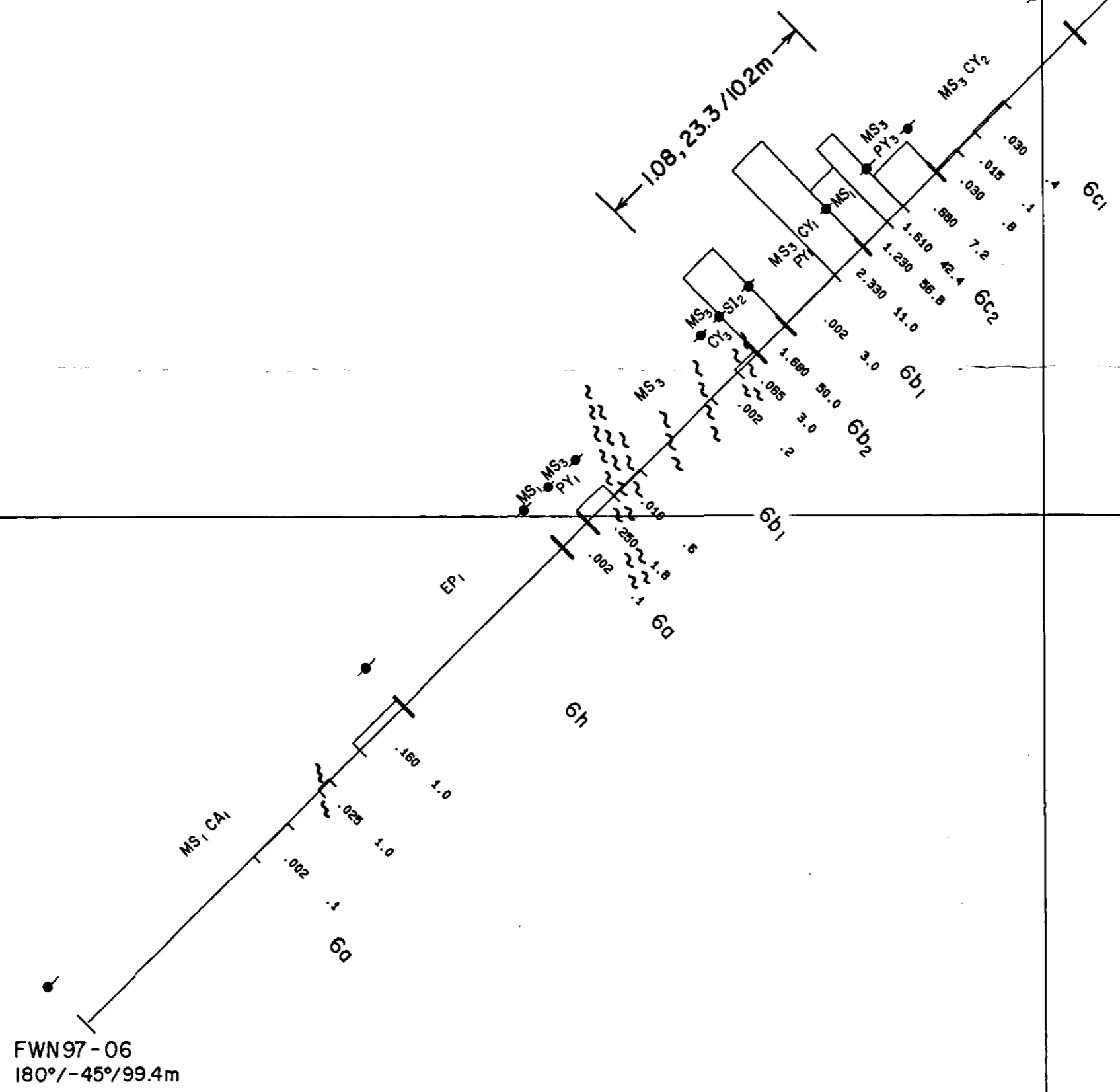
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FAWN PROPERTY
FWN 97-05
CROSS-SECTION
GIVER ZONE MAP 6

EQUITY	Date	OCT, 1997	Scale	1:200	FIGURE
	U.T.M. Zone	10 (NAD-27)	Mining District	OMINECA	10
	N.T.S.	93F/3E	State/Province	B.C.	

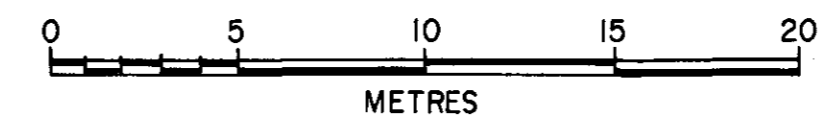
SECTION LOOKING WEST

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FWN97-06
180°/-45°/99.4m

SECTION LOOKING WEST



- LEGEND**
LITHOLOGIES
- EOCENE**
Ootsa Lake Group
11 Felsic dykes
11c Quartz-feldspar rhyolite porphyry
- EARLY TO MIDDLE JURASSIC**
Hazelton Group (Naglicco Formation)
- 6 Andesites
6a Lapilli to breccia tuff
6a₁ strong sericite-clay altered
6a₂ strong sericite-clay altered with chalcedony stockwork
6a₃ vuggy chalcadonic breccia
6a₄ strong sericite-clay altered andesite with jasper matrix crackle breccia
6a₅ chlorite-pyrite breccia
6a₆ intense oligoclase-sericite-biotite-pyrite altered
6b Fine-grained massive andesite
6b₁ strong sericite-clay altered
6b₂ strong sericite-clay altered with chalcedony stockwork
6b₃ chlorite-pyrite breccia
6b₄ plagioclase (oligoclase)-muscovite-pyrite chalcadony breccia
6b₅ chalcadony-epidote breccia
6b₆ intense oligoclase-sericite-biotite-pyrite altered
6c Feldspar porphyry
6c₁ strong sericite-clay altered
6c₂ strong sericite-clay altered with chalcadony stockwork
6g Maroon feldspar porphyry
6h Amygdaloidal andesite
6i Augite-feldspar phync andesite lapilli tuff
6j Maroon massive-bedded siltstone-coarse-grained sandstone
- 5 Epiclastics, Tuffs and Siltstones
5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone
5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone

- ALTERATION & MINERALIZATION**
- | | | |
|-----------------|-----------------|----------------|
| AS arsenopyrite | BA barite | BI biotite |
| CL chlorite | CY clay | DO dolomite |
| EP epidote | GL galena | HE hematite |
| MG magnetite | MS sericite | PF plagioclase |
| PO pyrrhotite | PR pyrrargyrite | PY pyrite |
| QZ chalcadony | SP sphalerite | |

- ALTERATION INTENSITY**
- 1 weak 2 moderate 3 strong

- SYMBOLS**
- 180°/-45°/71.3m Diamond drill hole (azimuth, dip, length)
3.200 390 Assay Interval: Au (g/t) Ag (g/t)
2.02, 25.2/8.1m Composite assay interval: Au (g/t) Ag (g/t) / Length (m)
- ~ ~ ~ ~ Fault/Shear
— — — — Lithological contact
• • • • Alteration contact
CL Alteration interval
- Histograms to left of drill holes correspond to gold grades

CASCADIA INTERNATIONAL VENTURES INC.

FAWN PROPERTY
FWN 97-06 MAP 7
CROSS-SECTION
GIVER ZONE

EQUITY	Date	0 CT. 1997	Scale	1: 200	FIGURE 11
	U.T.M. Zone	10 (NAD-27)	Mining District	OMINECA	
	N.T.S.	93 F/3E	State/Province	B.C.	

LEGEND
LITHOLOGIES

- EOCENE**
Ootsa Lake Group
- 11 Felsic dykes
11c Quartz-feldspar rhyolite porphyry
- EARLY TO MIDDLE JURASSIC**
Hazelton Group (Naglic Formation)
- 6 Andesites
- 6a Lapilli to breccia tuff
6a₁ strong sericite-clay altered
6a₂ strong sericite-clay altered with chalcidony stockwork
6a₃ vuggy chalcidony breccia
6a₄ strong sericite-clay altered andesite with jasper matrix crackle breccia
6a₅ chlorite-pyrite breccia
6a₆ intense oligoclase-sericite-biotite-pyrite altered
- 6b Fine-grained massive andesite
6b₁ strong sericite-clay altered
6b₂ strong sericite-clay altered with chalcidony stockwork
6b₃ chlorite-pyrite breccia
6b₄ plagioclase (oligoclase)-muscovite-pyrite chalcidony breccia
6b₅ chalcidony-epidote breccia
6b₆ intense oligoclase-sericite-biotite-pyrite altered
- 6c Feldspar porphyry
6c₁ strong sericite-clay altered
6c₂ strong sericite-clay altered with chalcidony stockwork
- 6g Maroon feldspar porphyry
6h Amygdaloidal andesite
6i Augite-feldspar phyric andesite lapilli tuff
6j Maroon massive-bedded siltstone-coarse-grained sandstone
- 5 Epilastics, Tuffs and Siltstones
5a Fine (sharp) laminated graphitic argillite-siltstone and grey felsic ash tuff and siltstone
5b Finely laminated argillite-siltstone and grey felsic ash tuff and siltstone

ALTERATION & MINERALIZATION

- | | | |
|-----------------|-----------------|----------------|
| AS arsenopyrite | BA barite | BI biotite |
| CL chlorite | CY clay | DO dolomite |
| EP epidote | GL galena | HE hematite |
| MG magnetite | MS sericite | PF plagioclase |
| PO pyrrhotite | PR pyrrargyrite | PY pyrite |
| QZ chalcidony | SP sphalerite | |

ALTERATION INTENSITY

- 1 weak 2 moderate 3 strong

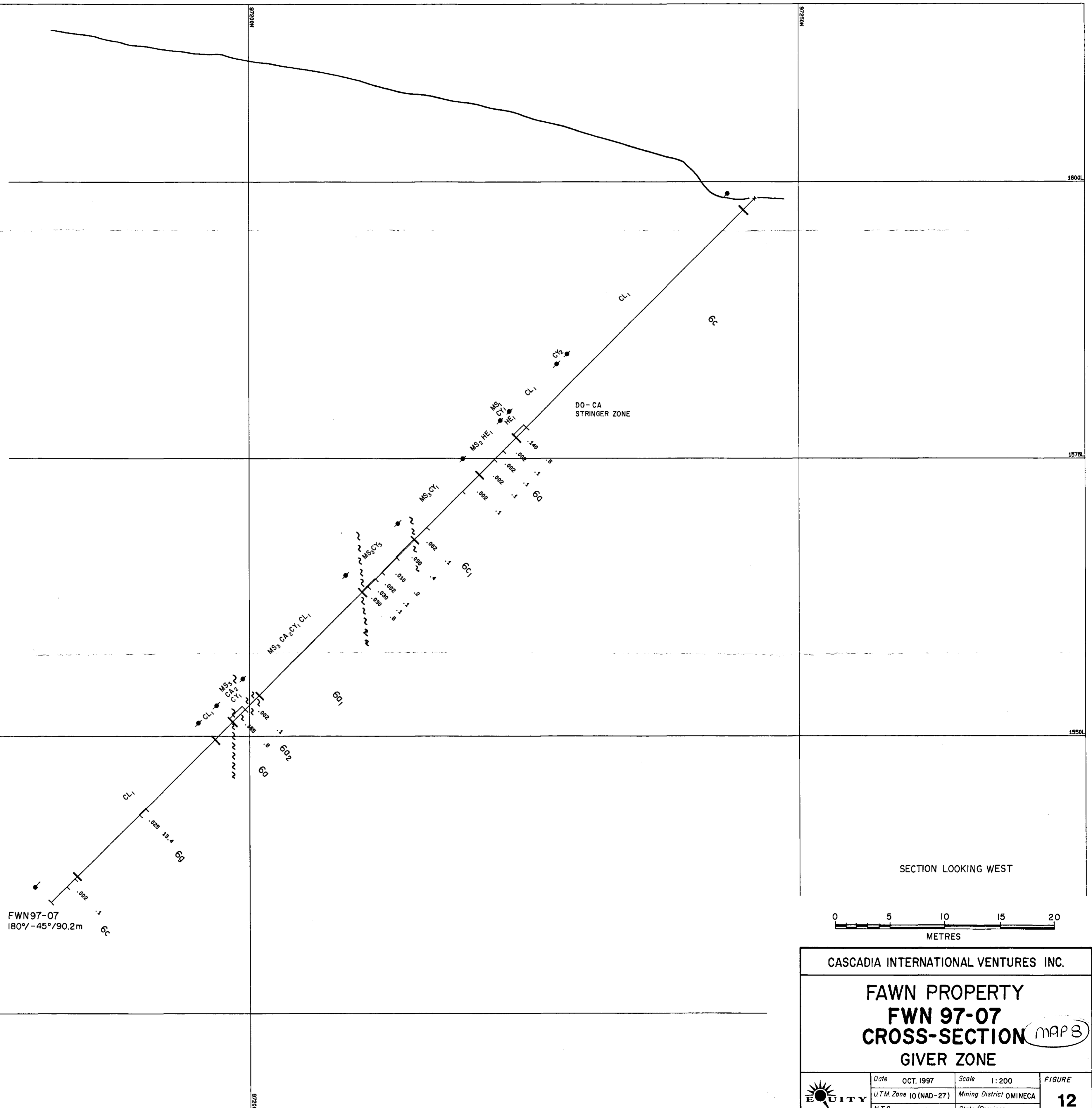
SYMBOLS

- 180°-45°/71.3m Diamond drill hole (azimuth, dip, length)
3.200 39.0 Assay interval: Au (g/t) Ag (g/t)
- 2.02, 25.2/8.1m Composite assay interval:
Au (g/t) Ag (g/t) / Length (m)
- ~ ~ ~ ~ Fault/Shear
- — — Lithological contact
- • • • • Alteration contact
- CL Alteration interval
- Histograms to left of drill holes correspond to gold grades

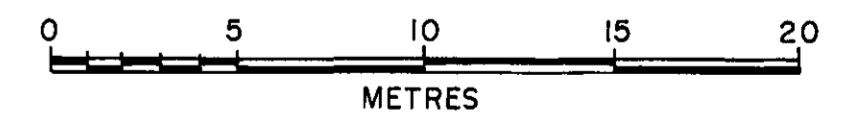
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FWN97-07
180°-45°/90.2m



SECTION LOOKING WEST



CASCADIA INTERNATIONAL VENTURES INC.

FAWN PROPERTY
FWN 97-07
CROSS-SECTION MAPS
GIVER ZONE



Date	OCT. 1997	Scale	1: 200	FIGURE
U.T.M. Zone	10 (NAD-27)	Mining District	OMINECA	12
N.T.S.	93 F/3E	State/Province	B.C.	