ASSESSMENT REPORT TITLE PAGE AND SUMMARY

2007 Diamond Drilling on the JW Property

Total Cost: $589,468.11

Author: Rory Kutluoglu

NOTICE OF WORK PERMIT NUMBER(S)/DATE(S): MX-1-732, Start 2007/May/22, End 2007/JUN/15
STATEMENT OF WORK EVENT NUMBER(S)/DATE(S): 4156985, 2007/JUL/05

Year: 2007
PROPERTY NAME: JW
CLAIM NAME(S) (on which work was done): JW (Claim # 509237)

COMMODITIES SOUGHT: Gold, Copper

MINERAL INVENTORY MINFILE NUMBER(S),IF KNOWN
MINING DIVISION: Liard
NTS / BCGS: 104G012
LATITUDE 57° 11’7.98”
LONGITUDE 131° 35’9.96" (at centre of work)
UTM Zone EASTING 343770 NORTHING 6341020

OWNER(S): Romios Gold Resources Inc
MAILING ADDRESS 25 Adelade St E. Suite 1010, Toronto, ON M5C 3A1

OPERATOR(S) Romios Gold Resources Inc
MAILING ADDRESS: 25 Adelade St E. Suite 1010, Toronto, ON M5C 3A1

REPORT KEYWORDS (lithology, age, stratigraphy, structure, alteration, mineralization, size and attitude **do not use abbreviations or codes**)
Andesite Volcanics, Triassic Jurassic, Galore Creek Intrusive, Stuhini Group, propylitic potassic, gold copper low sulphidation veins

REFERENCES TO PREVIOUS ASSESSMENT WORK AND ASSESSMENT REPORT NUMBERS
18114,00501,19904,20844,18116,20843,00669

Romios Gold Resources Inc.

2007 DIAMOND DRILLING REPORT ON THE JW PROPERTY

Located in the Galore Creek Area
Liard Mining Division
NTS 104G/4E
BCGS 104G.012, 104G.013
57^{o} 11’ North Latitude
131^{o} 35' West Longitude

-prepared for-

ROMIOS GOLD RESOURCES INC.
Suite 1010, 25 Adelaide Street E.
Toronto, Ontario, Canada
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-prepared by-

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October 1, 2007

SUMMARY

The JW property consists of a single map-selection claim covering approximately 6.1 km^2 of mountainous terrain in northwestern British Columbia, 170 km northwest of Stewart. Access to the property is currently by helicopter from seasonal bases at Bob Quinn Lake airstrip on Highway 37, approximately 80 kilometres to the east, but a limited-access road is under construction to the Galore Creek deposit, about 10 kilometres to the east of the JW property. The property is owned by the Galore Creek Staking Syndicate 2003, which has granted Romios Gold Resources Inc. an option to earn a 100% interest.

Prior exploration was carried out on the JW property from 1959 to 1990. Kennco explored porphyry copper mineralization from 1959 to 1965, reporting a trench with 13.1 metres grading 0.76% Cu. Bellex and Sarabat explored the property from 1988 to 1990; Bellex concentrated on the porphyry potential of its southern portion and Sarabat explored the Au-rich veins peripheral to the porphyry system on its northern portion. Sarabat identified a number of narrow, E-W trending quartz-sulphide veins, including Jake's Vein (170m strike length averaging 25.3 g/tonne Au across 0.23m) and the Boundary Vein (11.3 g/tonne Au across 3.4m). Bellex identified a strong coincident Cu-Au soil anomaly measuring 500 x 1,400 metres and drilled five holes on its fringes. Each of the holes intersected Cu-Au mineralization, including 51 metres @ 0.24% Cu and 0.48 g/tonne Au (JW90-01) and 45 metres @ 0.24% Cu and 0.36 g/tonne Au (JW90-03). In 2007, Romios drilled 3 holes (481.50 m) within the core of the soil geochemical anomaly, forming the basis of this report.

The JW property is underlain by Upper Triassic Stuhini Group flows, volcaniclastics and wackes, which have been intruded by a microdiorite of the Late Triassic to Jurassic Galore Creek intrusions and/or Texas Creek Suite intrusions. The 2007 drilling established that the anomalous Cu and Au soil geochemical signature in the valley of Jack Wilson Ck, which appears to be made up of a very thick sequence of alluvial material, exceeding 80m in thickness and is unlikely to represent bedrock metal values. JW07-06 also intercepted 2.4 metres with a weighted average of 31.9 g/tonne Au (true width of 1.8m) from 279.80m to 282.20m. This intersection corresponds to a series of narrow quartz carbonate veinlets and a 20cm quartz carbonate vein with irregular contacts and oriented @ 50^o to core axis. Sulphide mineralization in this intersection consisted of 2-4% pyrite, with no other base metal sulphides. The Boundary Zone vein is exposed on surface approximately 170m vertically above this intercept.

Future work to develop the property further should include additional surface mapping, soil contours and drilling. The surface component of the program would be focused on better identifying the nature of the veins previously identified as well as exploring for more vein style mineralization in the southern portion of the property. Additional drilling should be designed to extend and evaluate the significance of the intersection in JW07-06 and test for similar results at depth for other veins located in the northern portion of the property.

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1.0 INTRODUCTION

The author directed a diamond drilling program on the JW property in June 2007 for Equity Engineering Ltd. (“Equity”), under contract to Romios Gold Resources Inc. (“Romios”). Data from this program has been compiled, summarized and interpreted in this report. Additional information for this report was derived from publicly-available assessment reports and government maps and publications. The author’s experience with the property consists of the core-logging done during the 2007 program.

2.0 RELIANCE ON OTHER EXPERTS

The author did not rely on other experts regarding legal, environmental, political or other such issues.

3.0 PROPERTY DESCRIPTION AND LOCATION

The JW property lies in the Coast Range Mountains of northwestern British Columbia, approximately 170 km northwest of Stewart (Figure 1). It lies within the Liard Mining Division, centred at 57º 11’ north latitude and 131º 35’ west longitude.

The JW property consists of a single Mineral Titles Online (MTO) map-selection claim, covering 6.1 km^2, as summarized in Table 1. The JW property partially overlaps legacy claims on each side, reducing its area slightly; when the legacy claims lapse, the ground covered by them will revert to the JW MTO claim. The claim is held in the name of Romios Gold Resources Inc., but separate documents indicate that it is held by the Galore Creek Staking Syndicate 2003 (“Syndicate”), which has granted Romios an option to earn a 100% interest in it.

Table 1: Claim Data

Mineral Tenure	Area (Ha)	Expiry Date
509237	613.829	December 31, 2016
Total	613.829	

Surface rights over the JW property are owned by the Province of British Columbia. No significant surface disturbance or any major environmental liabilities have been noted by the author. Exploration permits must be obtained from the British Columbia Ministry of Energy, Mines and Petroleum Resources to carry out future exploration programs.

4.0 ACCESSIBILITY, CLIMATE, LOCAL RESOURCES, INFRASTRUCTURE AND PHYSIOGRAPHY

The JW property lies in the Coast Range Mountains of northwestern British Columbia, approximately 170 km northwest of Stewart. Access to the property is currently by helicopter from seasonal bases at the Bob Quinn Lake airstrip on Highway 37, approximately 80 kilometres to the east. Galore Creek Mining (NovaGold/TeckCominco) is building a limited-access road and tunnel to their Galore Creek deposit, which lies about 10 kilometres to the east of the JW property, and which is scheduled for completion in late 2008.

The JW property covers a bowl formed by the north fork (“North Fork Creek”) of Jack Wilson Creek, approximately seven kilometres above its confluence with the Stikine River. Topography is rugged, with elevations ranging from 300 to 1200 metres.

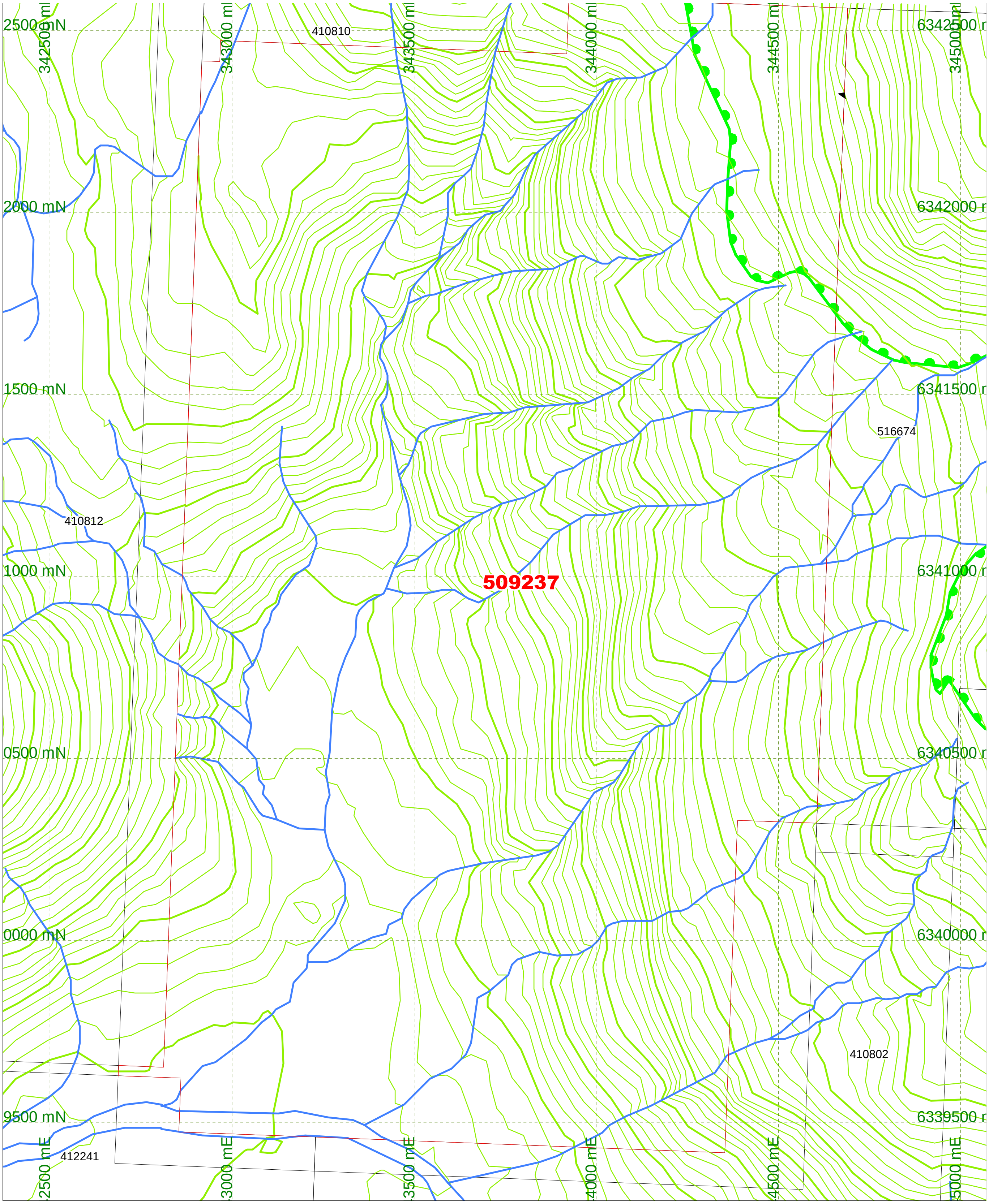
ROMIOS GOLD RESOURCES INC

JW Property

LOCATION MAP

Date: JUL 2007
Scale: 1:8,000,000
Figure 1
U.T.M. Zone UTM 9 - NAD83
Mining District LIARD
N.T.S. 104G/4
State/Province BC

EQUITY



ROMIOS GOLD RESOURCES INC

JW Property

**TENURE
MAP**

Date: JUL 2007	Scale: 1:15,000	Figure	
U.T.M. Zone UTM 9 - NAD83	Mining District LIARD		2
N.T.S. 104G/4	State/Province BC		

The bowl and lower slopes are covered by a dense growth of hemlock and spruce with an undergrowth of devil's club and huckleberry. Steeper open slopes are covered by dense slide alder growth. Open alpine vegetation is present above treeline, which lies near 1200 metres. Both summer and winter temperatures are moderate although annual rainfall may exceed 200 cm and several metres of snow commonly fall at higher elevations. The property can be worked from mid-May until October.

5.0 HISTORY

Table 2 summarizes all known exploration work carried out on the current JW property.

Table 2: JW Exploration Programs

Program	Geochemistry	Geophysics	Drilling/ Trenching	Reference
Kennco (1959)	silts, rocks			Rayner (pers comm)
Kennco (1963-65)	74 soils (Cu, Mo), rocks	magnetics, IP	trenches	BCDM (1965); Hallof (1965); Rayner (1965)
Bellex (1988)	4 silts, 267 soils, 54 rocks			Awmack and Yamamura (1988b)
Sarabat (1988)	4 silts, 125 soils, 97 rocks			Awmack and Yamamura (1988a)
Bellex (1989)	193 soils, 130 rocks		3 trenches	Chung (1990)
Bellex (1990)	201 soils, rocks	11.0 km magnetics, VLF-EM, IP	8 trenches, 5 DDH (1,392m; 4,566')	Blann and Vulimiri (1990)
Sarabat (1990)	151 rocks		trenches	Ostensoe (1990)
Romios (2007)		134.2 km airborne magnetics, EM	3 DDH (481.50m; 1,580')	This report
Totals	>8 silts, 860 soils, >432 rocks	>11.0 km magnetics, IP; 11.0 km VLF-EM	>11 trenches, 8 DDH (1,841m; 6,040')	

The Galore Creek district was extensively explored for its copper potential throughout the 1960's, following the discovery in 1955 of the Galore Creek deposit (928 Mt @ 0.50% Cu and 0.28 g/t Au). This work led to the discovery of the Copper Canyon deposit (165 Mt @ 0.35% Cu and 0.54 g/t Au) and several Cu-Au porphyry prospects including the JW and Trek. A second wave of exploration in the late 1980's focused on gold, following the discovery of the Snip and Eskay Creek mines 50 kilometres to the south and the recognition that similar geology extends north through the Galore Creek area.

In 1959, during the course of a regional stream sediment geochemistry survey, Kennco Explorations Ltd. sampled a narrow quartz-pyrite vein on the current JW property, which assayed 113 g/tonne Au (G. Rayner, pers. comm.). From 1963 to 1965, Kennco conducted geological mapping, soil geochemistry, ground geophysics and hand-trenching on the property. A trench beside North Fork Creek was reported to assay 0.76% Cu across 13.1 metre, but no further work was reported.

A second wave of exploration through the Galore Creek area in the late 1980's focused on gold, following the discovery of the Snip and Eskay Creek mines 70 kilometres to the south and the recognition that similar geology extends north through the Galore Creek area (Figure 3). In particular, several known porphyry prospects were restaked to evaluate their gold potential and to search for porphyry-related veins.

From 1988 to 1990, the current JW claim was explored intensively by Bellex Mining (southern part of property) and Sarabat Gold (northern part), with the claim Boundary near the north end of the North Fork Creek bowl, in the vicinity of the Boundary Zone. As a result, Bellex concentrated on the North Fork Cu-Au porphyry prospect, while Sarabat discovered and explored a number of Au-rich quartz-sulphide veins in the steeper terrain to the north, including the Diorite, Jake's, Fourteen, Cliff and Boundary veins.

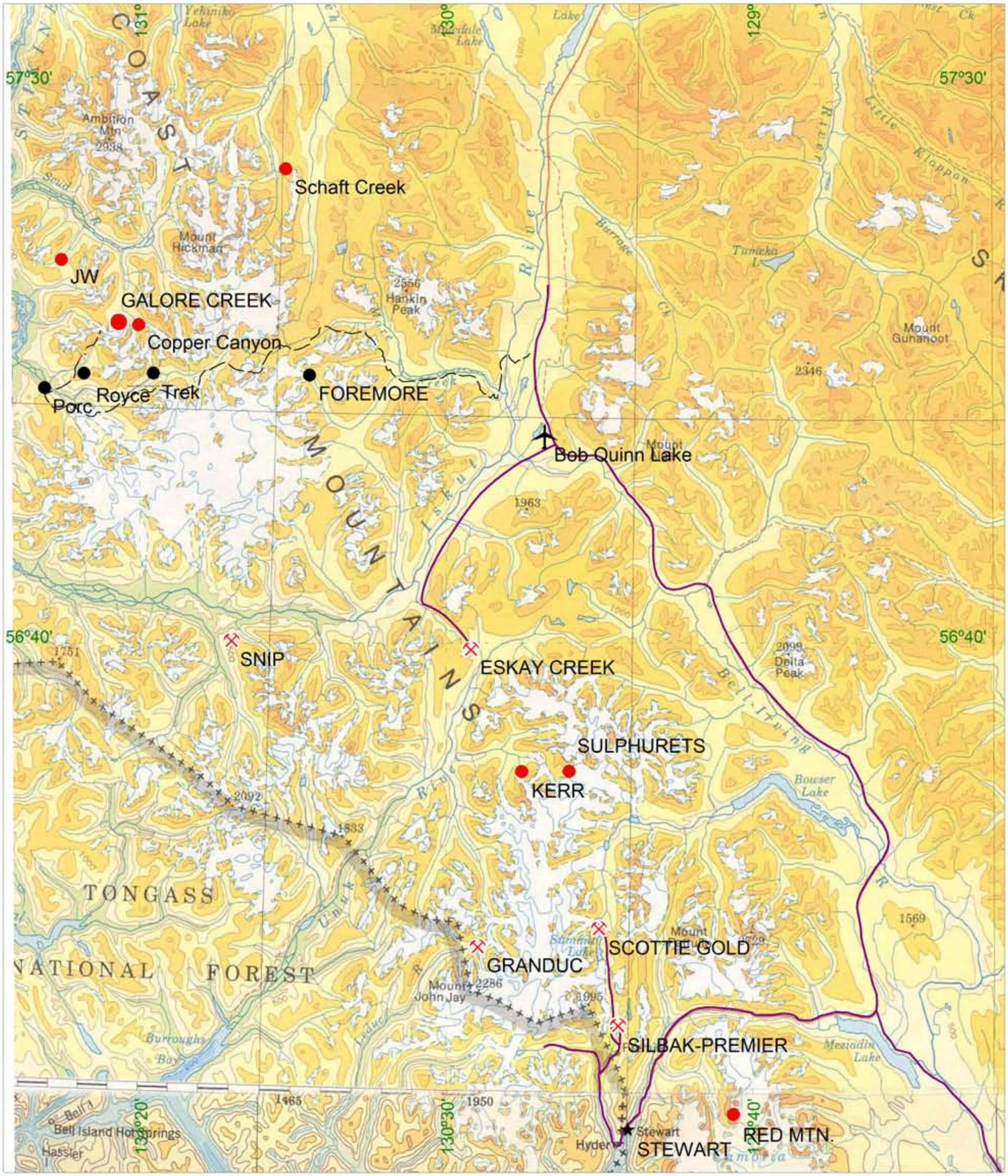
On the North Fork prospect, Bellex carried out mapping, prospecting, hand-trenching and grid-based soil geochemical, IP, magnetic and VLF-EM surveys from 1988 to 1990, outlining a strong, open-ended 500 x 1,400 metre Cu-Au soil geochemical anomaly. In 1990, Bellex drilled 5 holes to test Kennco's mineralized trench and their own geophysical anomalies. Hole 90JW-1 was drilled under Kennco's old trench which had been re-blasted and re-sampled, assaying 0.98% Cu and 0.5 g/t Au across 14 metres. In the drill hole, this projected zone was much narrower and dominantly pyritic, but further down the hole unexpectedly intersected a wide quartz-pyrrhotite-pyrite-chalcopyrite vein from 223.0 to 248.4m. Although this vein locally averaged 25% sulphides, as a whole its copper and gold grades were similar to those of its chlorite-epidote altered diorite wallrock. Holes 90JW-2 and 90JW-3 were 75 and 50 metre step outs to the south and north of 90JW-1, respectively, in an attempt to hit the quartz-sulphide vein along strike. Both holes hit the presumed extension of this vein, but its width had narrowed to tens of centimetres in each case. However, each of these holes hit two mineralized sections. The upper zones in 90JW-2 and 90JW-3 were hosted within strong epidote-chlorite altered diorite (90JW-2) and lapilli tuff (90JW-3). The lower zone in 90JW-2 was distinguished by pervasive silicification and Kspar flooding through the diorite matrix with decreasing epidote and chlorite. The lower zone in 90JW-3 centred on a pervasively silicified and Kspar-altered magnetite breccia. Each of the five holes intersected minor porphyry-style Cu-Au mineralization, including 45m grading 0.24% Cu and 0.36 g/t Au in hole 90JW-3 (Table 3).

Table 3: Pre-2007 Drill Intersections

Hole Number	Depth (m)	From (m)	To (m)	Length (m)	Au (g/t)	Cu (%)
90JW-1	281.9	219.0	270.0	51.0	0.48	0.24%
(incl.)		223.0	248.4	25.4	0.50	0.34%
90JW-2	278.8	48.0	65.8	17.8	0.20	0.09%
(and)		246.0	273.0	27.0	0.15	0.12%
90JW-3	286.5	18.0	63.0	45.0	0.36	0.24%
(and)		132.0	158.0	26.0	0.31	0.14%
90JW-4	258.8	115.8	132.7	16.9	0.04	0.11%
(and)		151.0	181.0	30.0	0.09	0.13%
90JW-5	286.5	121.5	160.0	38.5	0.09	0.25%
(and)		280.7	286.5*	5.8	0.00	0.26%

*Bottomed in mineralization

At the same time, Sarabat carried out mapping, prospecting, chip sampling and limited geophysical and geochemical surveys on the north end of the property in 1988 and 1990. A number of Au-rich veins were discovered and investigated, but none were ever drilled. No work was carried out by either Bellex or Sarabat after 1990 and the claims gradually lapsed.



Mineral Deposits

Kerr -	135 Mt @ 0.78% Cu, 0.34 g/t Au
Sulphurets -	0.75 Mt @ 15.4 g/t Au, 648 g/t Ag
Scottie Gold -	198,000 t @ 16.5 g/t Au
Granduc -	15.6 Mt @ 1.22% Cu
Snip -	1.3 Mt @ 24.5 g/t Au
Eskay Creek -	2.7 Mt @ 47 g/t Au, 2135 g/t Ag
Silbak-Premier -	5.3 Mt @ 10.9 g/t, 233 g/t Ag
red Mountain -	2.5 Mt @ 12..8 g/t Au
Schaft Creek -	971 Mt @ 0.31% Cu, 0.033% MoS2
Galore Creek-	264 Mt @ 0.62% Cu, 0.35 g/t Au, 5.9 g/t Ag
Copper Canyon -	36 Mt @ 0.75% Cu, 1.17 g/t Au, 17 g/t Ag

0 10 20 40
kilometres

N

ROMIOS GOLD RESOURCES INC

JW Property

Regional Mineral Deposits

Date: OCT 2007	Scale: 1:1,000,000
U.T.M. Zone UTM 9 - NAD83	Mining District LIARD
N.T.S. 104A,B,G,H	State/Province BC

Figure 3

5.1 2007 Exploration Program

Exploration on the JW property in 2007 consisted of an airborne geophysical survey and 481.50 metres of diamond drilling in three holes on the North Fork porphyry prospect. The exploration was carried out by Equity Engineering Ltd. under contract to Romios Gold Resources Inc. The author directed the diamond drilling out of a tent camp near the junction of Jack Wilson Creek and the Stikine River. Helicopter support for the drilling was provided by a contract Long Ranger on site operated by Quantum Helicopters of Terrace, BC. A magnetic declination of 22°30'E was used for all compass measurements. All maps and UTM coordinates are referenced to the 1983 North American Datum (NAD-83).

The airborne geophysical survey was subcontracted to Fugro Airborne Surveys of Mississauga Ontario, covering the property with 134.2 line-kilometres of helicopter-borne magnetic/EM survey. East-west flightlines were spaced 100 metres apart. Due to scheduling problems and conflicts with mountain goat kidding season, the geophysical survey was not completed prior to the drilling and will be reported separately.

Diamond drilling was subcontracted to Driftwood Diamond Drilling of Smithers, BC, who proved very capable. Core was logged at the camp area, with geological, geotechnical and magnetic susceptibility data recorded (Appendix B). Core was sawn longitudinally, with one-half used for analyses and the other half retained on site, cross-stacked at the location of JW06-08 with timbers used on this project. Core samples were analyzed by ALS Chemex Labs of North Vancouver for Au (fire assay) and 34-elements by aqua regia acid digestions and ICP-AES. Laboratory certificates are attached in Appendix C. Pulp assays were carried out for high geochemical values of Au, Ag or Cu; the assays were used for plotting and calculations. The procedures, results and conclusions of the sampling QA/QC program are summarized in Appendix E.

6.0 GEOLOGICAL SETTING

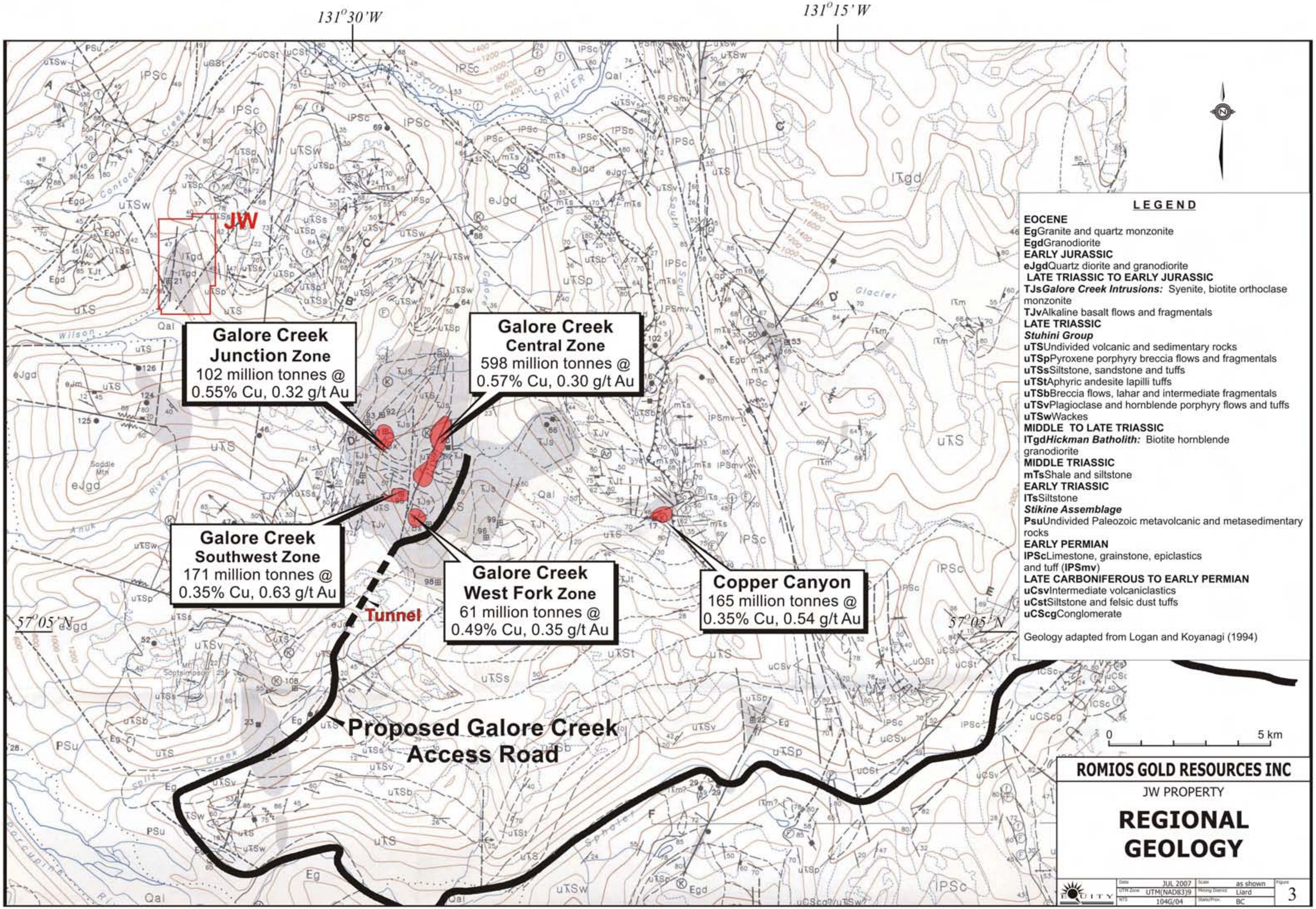
6.1 Regional Geology

The regional geology in the Galore Creek area consists of mid-Paleozoic and Mesozoic island arc successions, intruded by Triassic, Jurassic and Eocene plutons (Figure 4). Regional mapping has been carried out at a scale of 1:50,000 by Logan et al (1989) and Logan and Koyanagi (1989, 1994) of the BCGS.

The Paleozoic Stikine Assemblage comprises four main subdivisions:

- Devonian to Carboniferous(?) variably foliated limestone, phyllite, mafic and felsic flows and tuff;
- overlain apparently conformably by 700m of Lower to Middle Carboniferous limestone;
- overlain conformably to unconformably by >300m of Upper Carboniferous(?) to Permian thick-bedded conglomerate, siliceous siltstone and mafic to intermediate volcaniclastics;
- overlain apparently conformably by >800m of Lower Permian fossiliferous limestone.

A narrow belt of Lower and Middle Triassic sedimentary rocks, comprising silty shales, argillites, limy dolomitic siltstones, cherty siltstones and rare carbonaceous limestones, extends northerly from Copper Canyon. Elsewhere, the Stikine Assemblage is unconformably overlain by island arc volcanic and sedimentary rocks of the Upper Triassic Stuhini Group. Volcanic rocks comprise the bulk of the Stuhini Group stratigraphy in the Galore Creek area, with three different calcalkaline volcanic suites: a lower subalkaline hornblende-bearing basaltic andesite, a subalkaline to alkaline augite-porphyritic basalt and an uppermost alkaline orthoclase and pseudoleucite-bearing shoshonitic basalt. The lower suite is most voluminous and least distinctive, with aphyric and sparse hornblende and plagioclase-phyric flows, breccia and tuff. Rocks are fine to medium-grained, massive and fragmental textures are common. The middle suite consists of augite and feldspar-phyric breccia flows and fragmental rocks. The upper volcanic unit consists of an interbedded sequence of basic, coarse pyroxene feldspar flow breccias, orthoclase-feldspar crystal tuffs and coarse pseudoleucite flows and/or sills.



LEGEND

EOCENE
EgGranite and quartz monzonite
EgdGranodiorite
EARLY JURASSIC
eJgdQuartz diorite and granodiorite
LATE TRIASSIC TO EARLY JURASSIC
TJs*Galore Creek Intrusions:* Syenite, biotite orthoclase monzonite
TJvAlkaline basalt flows and fragmentals
LATE TRIASSIC
Stuhini Group
uTSUndivided volcanic and sedimentary rocks
uTSpPyroxene porphyry breccia flows and fragmentals
uTSsSiltstone, sandstone and tuffs
uTStAphyric andesite lapilli tuffs
uTSbBreccia flows, lahar and intermediate fragmentals
uTSvPlagioclase and hornblende porphyry flows and tuffs
uTSwWackes
MIDDLE TO LATE TRIASSIC
lTgd*Hickman Batholith:* Biotite hornblende granodiorite
MIDDLE TRIASSIC
mTsShale and siltstone
EARLY TRIASSIC
lTsSiltstone
Stikine Assemblage
PsuUndivided Paleozoic metavolcanic and metasedimentary rocks
EARLY PERMIAN
lPScLimestone, grainstone, epiclastics and tuff (**lPSmv**)
LATE CARBONIFEROUS TO EARLY PERMIAN
uCsvIntermediate volcaniclastics
uCstSiltstone and felsic dust tuffs
uCScgConglomerate

Geology adapted from Logan and Koyanagi (1994)

ROMIOS GOLD RESOURCES INC

JW PROPERTY

REGIONAL GEOLOGY

EQUITY

Date	JUL 2007	Scale	as shown	Figure
UTM Zone	UTM(NAD83)9	Mining District	Liard	3
NTS	104G/04	State/Prov.	BC	

Four suites of intrusive rocks have been distinguished in the region. The Hickman batholith (~230-226 Ma) is a composite 1200 km^2 body which shows crude zonation from pyroxene diorite in the core to biotite granodiorite near the margins. The alkalic Galore Creek Intrusions (~210-198 Ma) consist of ten phases of orthoclase-porphyritic syenite intrusions cutting coeval Stuhini Group rocks of the upper volcanic unit (Logan, 2005; Enns et al., 1995; Mortensen et al., 1995). These are spatially and genetically related to the Galore Creek and Copper Canyon Cu-Au porphyry deposits.

Calcalkaline intrusions of the Early Jurassic Texas Creek suite (~205-187 Ma) are common through the Stewart/Unuk/Iskut/Galore area and are associated with a number of porphyry (Kerr) and related vein (Sulphurets, Scottie, Snip, Silbak Premier, Red Mountain) deposits.

Small Eocene (~51-55 Ma) circular stocks and plugs of biotite quartz monzonite are scattered throughout the area. Logan and Koyanagi (1994) believe them to be satellite bodies to the main Coast Plutonic Complex, which lies to the west. They are generally equigranular, medium-grained and unaltered.

The dominant structures in the Galore Creek area are two approximately orthogonal fold trends, an earlier westerly trend and a later one trending northerly. These structures deform earlier synmetamorphic, pre-Permian structures and related northeast striking penetrative foliations. East-dipping reverse faults which imbricate the Stikine Assemblage and offset Early Jurassic plutons are associated with north-trending folding. Northeast sinistral fault zones and younger north-striking extensional faults host Eocene stocks and Miocene dykes, respectively (Logan and Koyanagi, 1994).

6.2 Property Geology and Mineralization

No mapping was done on the JW property in 2007. The geology and mineralization described below and shown on Figures 5 and 6 has been compiled from 1988-90 property-scale mapping.

Structurally and stratigraphically complex flows (uTSp), volcaniclastics (uTSb) and wackes (uTSs) of the Upper Triassic Stuhini Group have been intruded by a dark green diorite (TJd). The diorite, which underlies the bulk of the JW property, is thought to be coeval with, and subvolcanic to, the Stuhini Group volcanic rocks (Figures 4, 5). This would make it roughly the same age as the Galore Creek Intrusions hosting the Galore Creek and Copper Canyon porphyry deposits.

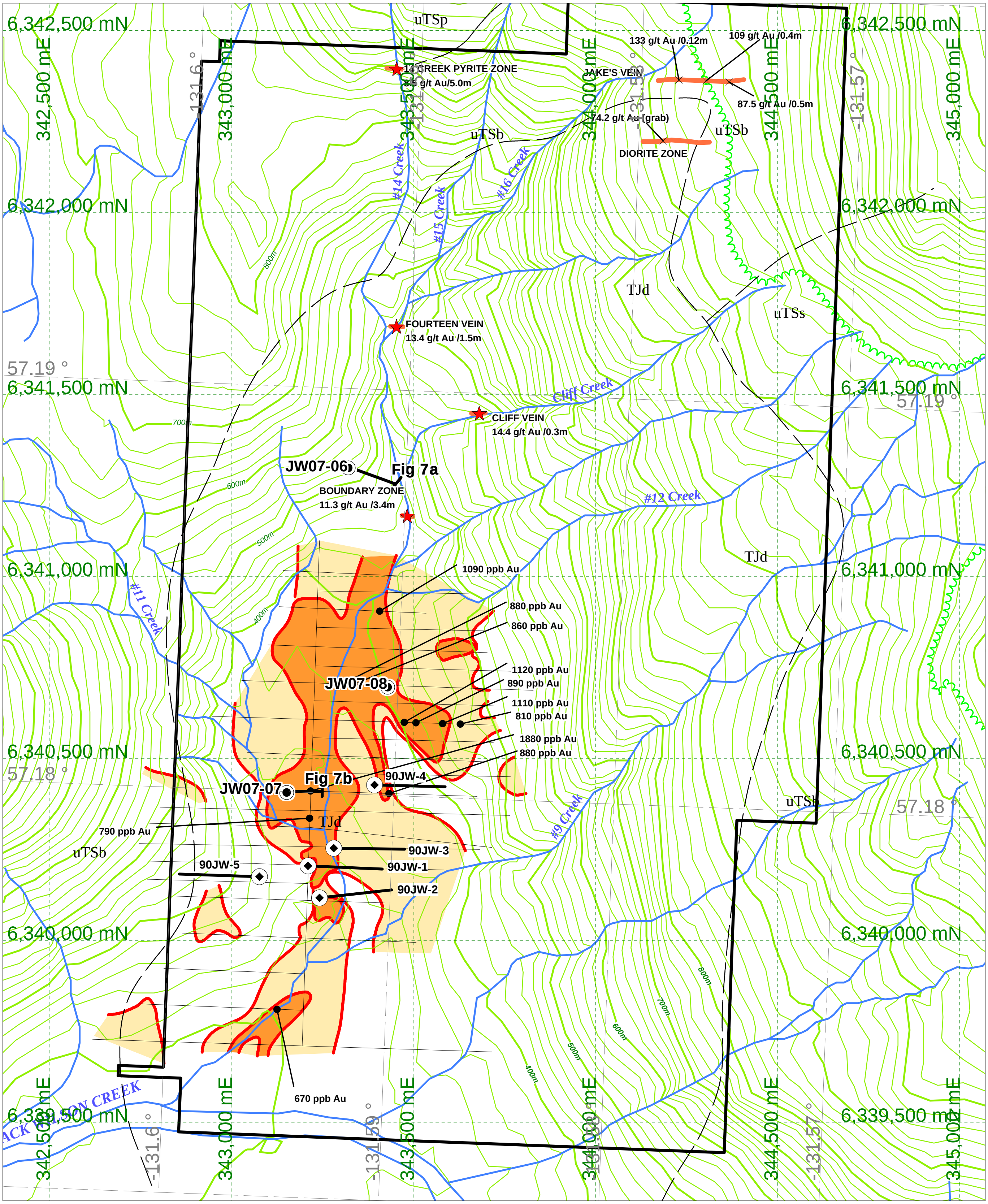
Two distinct, but related, styles of mineralization are present on the JW property:

- porphyry Cu-Au mineralization at lower levels and along the west margin of the microdiorite in the North Fork prospect; and
- Au-rich quartz-sulphide veins, both within the microdiorite (Boundary, Diorite, Fourteen and Cliff veins) and the intruded Stuhini Group volcanics (Jake's vein). All of the significant veining found to date trends ESE and dips moderately to steeply south.

6.2.1 North Fork Cu-Au Porphyry Prospect

The North Fork bowl is underlain by diorite which has been highly fractured and moderately to intensely chlorite-epidote altered. A number of strong shear zones are present and extensively chloritized. Widespread chalcopyrite mineralization is largely fracture-controlled and increases with the intensity of chloritic alteration. The best historic drill intersection (45m @ 0.24% Cu and 0.4 g/t Au in hole 90JW-3) was from a highly chloritic, sheared interval. It appears that this interval represents one of a number of subparallel shear zones which trend 020° and are offset by faults trending 090° and 150°.

Drilling has shown that beneath the central chlorite-epidote (propylitic) alteration zone are weak to strong zones of pervasive silica-anhydrite-Kspar (potassic) alteration. Magnetite occurs as veinlets, clots and breccias in the east-central grid area, commonly associated with Kspar alteration. A magnetite breccia in hole 90JW-3 assayed 0.14% Cu and 0.3 g/t Au across 26 metres.

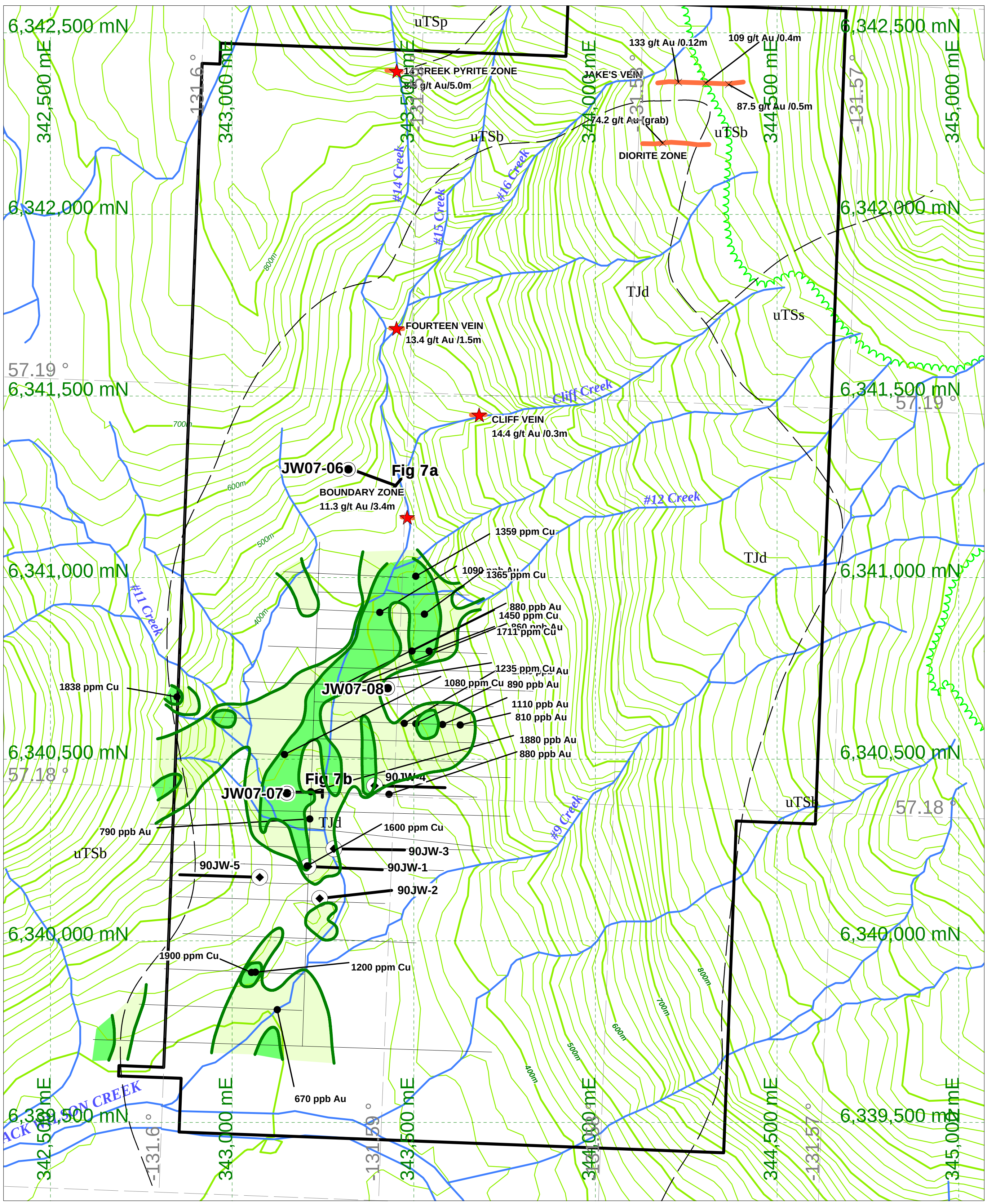


ROMIOS GOLD RESOURCES INC

JW Property

Property Geology
Au Geochem

Date: JUL 2007	Scale: 1:15,000	Figure		
U.T.M. Zone UTM 9 - NAD83	Mining District LIARD	5		
N.T.S. 104G/4	State/Province BC			



ROMIOS GOLD RESOURCES INC

JW Property

Property Geology
Cu Geochem

Date: JUL 2007	*Scale:* 1:15,000	*Figure*
U.T.M. Zone UTM 9 - NAD83	*Mining District* LIARD	6
N.T.S. 104G/4	*State/Province* BC	

EQUITY

The North Fork bowl is flanked to the north, east and west by gossanous cliffs with up to 3% disseminated pyrite and scattered gold-bearing quartz-sulphide veins, representing a pyrite halo around the North Fork porphyry prospect. To the south, it is bounded by the broad valley of Jack Wilson Creek.

6.2.2 Quartz-Sulphide Veins

A number of quartz-sulphide veins have been reported north and west of the North Fork porphyry prospect, within its pyritic halo.

- **Jake's Vein:** Jake's Vein is a quartz-pyrite-chalcopyrite-galena vein following a small creek near the northern boundary of the JW property. It lies near the contact between the microdiorite and volcaniclastic rocks, striking east-west and dipping 66º to the south. Sarabat reported 29 chip samples along 170 metres of strike-length of the vein, with a **weighted average of 25.3 g/t Au across 0.23 metres**. The best chip samples assayed **169 g/t Au across 0.4m** and 133 g/t Au across 0.12m (Ostensoe, 1990).
- **Diorite Vein:** The Diorite Vein is hosted within microdiorite approximately 170 metres south of Jake's Vein. It is a quartz-pyrite vein running along the north edge of a 10-20 metre wide zone of ankerite alteration, at its contact with relatively fresh microdiorite, and has been sampled along 200 metres of strike length. Widths vary from a few centimetres to >1 metre and grades are generally low, but one **grab sample assayed 74.2 g/t Au** (Ostensoe, 1990).
- **Fourteen Vein:** The Fourteen Vein, a quartz vein with pods of coarse pyrite, magnetite and chalcopyrite in chlorite-ribboned quartz, is exposed near the junction of North Fork Creek and #14 Creek. It trends 110º/35ºS, pinching and swelling from 40 centimetres to >2 metres in width. A **1.5 metre grab sample** across the vein **assayed 13.4 g/t Au**; the chloritic, magnetitic and pyritic wallrock graded 830 ppb Au across 7 metres (Awmack and Yamamura, 1988).
- **Cliff Vein:** The Cliff Vein, a further 200 metres south, is a 30-100 cm wide quartz-pyrite-chlorite-magnetite vein which strikes 105°/70°S. A grab sample assayed **14.4 g/t Au** across 30 centimetres (Awmack and Yamamura, 1988).
- **Boundary Zone Vein:** Several intensely silicified shear zones are exposed within microdiorite near the foot of the North Fork canyon. A **3.4 metre chip sample** across one of these shears assayed **11.3 g/t Au** and 0.11% Cu; a 1.0 metre grab sample across another portion of the same shear assayed 42.8 g/t Au and 0.24% Cu (Awmack and Yamamura, 1988). In 1990, a 0.5m wide x 1.0m tall panel sample assayed **39.7 g/t Au** and 0.21% Cu (sample JW-T3-06), presumably from the same silicified shear zone with milky blue quartz veinlets (Chung, 1990). Although the location of the surface showing(s) is uncertain, hole JW07-06 may have intersected the Boundary Zone Vein at depth.
- **14 Creek Pyrite Zone:** Between 800 and 1000 metres elevation, #14 Creek follows a well-defined easterly-dipping fault and shear zone crossing well-bedded tuffaceous and siliceous sediments. Up to 15% pyrite is present in some very fine-grained strata. A **5-metre grab** sample across one of these assayed **8.5 g/t Au** (Awmack and Yamamura, 1988); slabs of massive pyrite nearby returned 61.5 g/t Au (Ostensoe, 1990).

7.0 2007 DIAMOND DRILLING

A total of 3 holes totalling 481.50 metres were cored with BTW tools in the North Fork prospect (Table 4) between June 3 and 11, 2007. These holes were designed to test the Boundary Zone Vein and the heart of the Cu-Au soil geochemical anomaly, as defined by >200 ppb Au and >500 ppm Cu (Figures 5 and 6), to the north and northwest of the 1990 drilling. All holes were directed to the east, in order to cut mineralized shears which had been previously mapped (Blann and Vulimiri, 1990) as trending 000° to 020° and dipping steeply to the west.

Table 4: 2007 Diamond Drilling Survey Data

Drill Hole	Collar (UTM NAD 83)			Azimuth	Inclination	Length (m)
	Northing	Easting	Elev (m)			
JW07-06	6341301	343320	547	110	-60	297.5
JW07-07	6340409	343151	362	50	-50	151.5
JW07-08	634699	343428	384	110	-70	32.5
					TOTAL	481.5

Notable Au-Cu mineralization was encountered in JW07-06 (Table 5); the favourable intercepts in this hole consisted of both a broad low grade Au Cu intersection, as well as a narrow high grade gold zone within the lower part of the hole JW07-07 failed to intersect any significant mineralization while JW07-08 was abandoned in overburden.

Table 5: Significant 2007 Drill Intersections

Drill Hole	From (m)	To (m)	Interval (m)	Au (g/t)	Cu (ppm)	Mo (ppm)
JW07-06	279.80	282.20	2.4	31.87	333	4
JW007-06	53.60	81.00	27.4	0.23	1655	2
Incl.	71.90	81.00	9.1	0.49	2233	1

7.1 Drill Hole JW07-06 (Figure 7a)

This hole was designed to test the porphyry potential north of the Cu and Au soil anomaly as well as the Boundary Zone at depth. The hole intersected massive flow andesites and andesite breccia (ANDS and ANBX) which were dark to light green, fine to very fine grained with moderate silica, and epidote alteration. The unit also is weakly carbonate altered in patches with small veinlets of quartz, carbonate and epidote. The only mineralization visible is two to four percent fine grained pyrite disseminations and veinlets. Beneath the andesites is a granodiorite (GRDR), which is green to dark green, mottle textured, medium to fine grained with moderate to strong silica and epidote alteration, weak chlorite and spotty weak carbonate and potassium alteration. JW07-06 intersected a broad low grade section (27.4m of 0.23 g/t Au and 0.17% Cu) contained within the inter-fingered contact between the andesite volcanics and the granodiorite intrusive. Within the large interval of granodiorite is an interval of breccia which is dark green with a fine grain matrix containing pebble sized clasts of quartz, diorite and granodiorite with moderate chlorite alteration. The clasts are rounded to sub-rounded pebbles from 1 to 5 cm in size. Below the granodiorite there is an interval of lamprophyre (LMPR) dyke which is very narrow and light grey to buff coloured and has a fine grained groundmass with phyric feldspar quartz and magnetite. The bottom of the hole consisted of more massive andesite flows.

Hole JW07-06 was collared north of the northern limit of the soil geochemical grid, above and behind an outcrop mapped by Blann and Vulimiri (1990) as the Boundary Zone (“strong fracture-controlled chalcopyrite-pyrite mineralization”), although it is not clear exactly where the Boundary Zone Vein (3.4m @ 11.3 g/tonne Au and 0.11% Cu) is located relative to this outcrop. Hole JW07-06 intersected 2.4m (1.8m true width) grading 31.87 g/t Au within an interval of andesite which is moderately epidote altered and contains several narrow quartz carbonate veins with low pyrite content. The largest vein is a 20cm quartz calcite and pyrite vein with irregular contacts oriented at 50^o to core axis located at 281.80m. The vein also contains minor magnetite and chlorite along the irregular contacts.

7.2 Drill Hole JW07-07 (Figure 7b)

Hole JW07-06 was designed to test under the main part of the Au Cu soil anomaly, immediately north of L2+00N from the geophysical and soil grid. The objective was to test the southern lobe of the Au/Cu anomaly. This hole intersected predominantly massive flow andesites as described in JW07-06 as well as a small interval of microdiorite (DIOR) which is fine to very fine grained and grey to buff with overall homogenous crystalline texture. This hole failed to intercept any significant alteration or mineralization.

7.3 Drill Hole JW07-08

This hole tested the centre of the broadest portion of the anomalous gold and copper soil geochemistry. It was abandoned at 32.5 metres without reaching bedrock.

8.0 INTERPRETATION AND CONCLUSIONS

Two main conclusions can be drawn from the 2007 drilling on the JW property. The first is that the soil geochemistry from the grid in the valley is largely unrelated to the underlying bedrock. Secondly, there may be significant vein mineralization at depth in the area of the Boundary Zone.

- The surface soils are not an effective tool in the valley.
- Alluvial material in the valley might be derived from the surface vein showings in the northern portion of the property, or as yet discovered mineralization in the cliffs above the bowl.
- Where surface work has been done in the past, there are several gold-rich veins; the veins appeared too narrow to be of economic interest (such as Jake's Vein with 25.3 g/t Au across 0.23m). The vein swarm in JW07-06 (31.87 g/t across 1.8m true width) indicates the potential for better widths for these gold-rich veins on the JW.
- Additional drilling is required in the vicinity of the Boundary Zone to better understand the orientation and significance of this gold rich vein swarm in JW07-06.
- Further work, including diamond drilling, should be undertaken on the known veins, looking for potentially economic widths.
- The differing deposit focuses of Bellex and Sarabat during the 1988-1990 exploration led to the discovery of several gold rich veins on the northern (Sarabat) part of the JW property, but none on the southern (Bellex) part. Mapping and prospecting should search for these in the cliffs around the North Fork bowl on the southern portion of the property, particularly in the broad contact zone between the diorite and volcanics.
- The contacts between the intrusives and the volcanics appear to be most favourable for hosting broader, lower grade porphyry-style Cu-Au mineralization.
- The porphyry potential of the JW property has not been eliminated by the 2007 drilling, but has been limited by the uneconomic results in the two holes that reached bedrock and the realization that the Cu-Au soil geochemical anomaly is at least partially unrelated to underlying bedrock mineralization. The results of the 2007 airborne geophysical survey may give new insight into the JW's porphyry potential.
- The results from the drilling performed by Bellex suggested that the potassic zone could potentially extend to the north, as far as the Boundary Zone, but this program indicates that if the potassic zone does extend to the north it would be at depths greater then drilled during this program. Phyllic alteration in the drilling is not present in any significant intensity, while a propylitic zone is present as a weak and subtle alteration of predominantly the andesite.

The veins on the northern end of the property appear to be porphyry peripheral quartz-(carbonate)-sulphide veins, similar to the Snip Deposit, located about 60 kilometres to the south in a similar geological setting. From 1991 to 1999, the Snip Mine produced 32.093 tonnes of gold, 12.183 tonnes of silver and 249 tonnes of copper from about 1.2 million tonnes of ore. The intercept in JW07-06 of 1.8m (true width) averaging 31.9 g/tonne Au corresponds to a series of narrow carbonate veinlets and a 15cm quartz carbonate vein with irregular contacts and oriented @ 50^o to core axis. Sulphide mineralization in this intersection consisted of 2-4% pyrite, with no other base metal sulphides (this intersection is accompanied by just 332 ppm Cu and 25 ppm Zn. The Boundary Zone vein is approximately 170m vertically above this intercept, where exposed on surface. The Boundary Zone was discovered in 1988 with a grab sample assaying 42.87 g/t Au and 0.24% Cu; a 3.4m chip sample in the same area assayed 11.27 g/t and 0.11% Cu in a siliceous shear oriented approximately 020^{o}. Panel sampling was done in 1990, apparently over the same vein (locations are unconfirmed due to snow cover during this program), giving one assay of 39.7 g/t Au and 0.2% Cu. The location and nature of the mineralization in JW07-06 when compared with the Boundary Zone shear is suggestive of a relation between the two, although the Cu content is considerably lower.

Respectfully submitted,

Rory Kutluoglu, Geologist

Vancouver, British Columbia

October 1, 2007

Appendix A: References

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Appendix B: Statement of Expenditures

STATEMENT OF EXPENDITURES

JW PROPERTY

June 3 –11, 2007

PROFESSIONAL FEES AND WAGES:

Henry Awmack, P.Eng.				
	2.63 days @	$650/day	$ 1,709.50	
Shellie Davies, Cook/First Aid				
	14.00 days @	$500/day	7,000.00	
Rory Kutluoglu, Project Geologist				
	49.70 days @	$650/day	32,305.00	
John Quock, Senior Sampler				
	24.00 days @	$325/day	7,800.00	
Scott Parker, GIS/Logistics				
	51.50 hours @	$75/hour	3,862.50	
Neil Visser, Logistics				
	151.00 hours @	$75/hour	11,325.00	
Clerical				
	71.50 hours @	$35/hour	2,502.50	$ 66,504.50

EQUIPMENT RENTALS

Generator (6.5kVA)				
	5 days @	$35/day	$ 175.00	
Generator (1kVA)				
	3 days @	$20/day	60.00	
Rental Truck Insurance				
	13 days @	$10/day	130.00	
Field Camp				
	154 days @	$35/manday	5,390.00	
Satellite Phones (Iridium)				
	6 weeks @	$62.50/week	375.00	
	951 minutes @	$1.69/min	1,607.19	
Chainsaw				
	38 days @	$30/day	1,140.00	
Field Computers				
	37 days @	$40/day	1,480.00	
First Aid Equipment (Level III)				
	25 days @	$30/day	750.00	
PDAs				
	15 days @	$20/day	300.00	
Core Saw				
	13 days @	$25/day	325.00	11,732.19

EXPENSES:

Chemical Analyses	$ 5,446.46
Field Consumables	49.56
Materials and Supplies	15,467.16
Plot Charges	6.95
Printing and Reproductions	7.50
Camp Food	1,019.09

STATEMENT OF EXPENDITURES (con't)

Meals	1,477.93	
Accommodation	2,615.76	
Taxis and Airporters	98.96	
Truck Rental (non-Equity)	2,110.93	
Automotive Fuel	661.22	
Helicopter Charters	252,509.18	
Airfare	4,432.41	
Telephone Distance Charges	69.53	
Courier	46.02	
Freight	11,076.47	
Bulk Fuel	18,337.36	
Drum Deposits	5,555.08	
Geophysical Equipment Rental	112.50	
Padbuilding	20,475.00	
Radio Rental (non-Equity)	769.43	
Other Equipment Rental (non-Equity)	3,149.80	
Drilling: Mob/Demob	13,260.00	
Driling: Footage	47,910.00	
Drilling: Materials	6,187.91	
Drilling: Standy/Moves/Travel	1,360.00	
Avalanche Consulting	1,150.00	
Expediting	1,262.00	
Contract Labour	7,050.00	423,674.21
SUB-TOTAL:		$ 501,910.90
PROJECT SUPERVISION CHARGES:		
12% on portion <$200,000: ($200,000.00)	$ 24,000.00	
10% on balance: ($301,910.90)	30,191.09	54,191.09
SUB-TOTAL:		$ 556,101.99
GST: 6% on sub-total		33,366.12
TOTAL:		**$ 589,468.11**

Appendix C: Diamond Drill Logs

Note: Quality control and quality assurance samples have been included in the geochemical section of these logs. Geochemical standards inserted into the sample stream show up as 0.00 metre width on the logs. Duplicate samples can be identified by repetition of the sample interval on the log.

MINERALS AND ALTERATION TYPES

AL	alunite	EN	enargite	MT	marcasite
AS	arsenopyrite	EP	epidote	NE	neotocite
AZ	azurite	GE	goethite	PA	pyrargyrite
BA	barite	GL	galena	PL	pyrolusite
BI	biotite	GR	graphite	PO	pyrrhotite
BO	bornite	HE	hematite	PY	pyrite
BT	pyrobitumen	HS	specularite	QZ	quartz veining
CA	calcite	HZ	hydrozincite	RE	realgar
CB	Fe-carbonate	JA	jarosite	RN	rhodonite
CC	chalcocite	KF	potassium feldspar	SB	stibnite
CD	chalcedony	MC	malachite	SI	silicification
CL	chlorite	MG	magnetite	SM	smithsonite
CP	chalcopyrite	MN	Mn-oxides	SP	sphalerite
CV	covellite	MR	mariposite/fuchsite	SR	scorodite
CY	clay	MS	sericite	TT	tetrahedrite

ALTERATION INTENSITY

s	strong	m	moderate	w	weak

DRILL LOG

Project: JW	**Collar Elevation (m):** 547.0
Hole JW07-06	**Azimuth (º):** 110
Location: 6341301 m North 343320 m East	**Dip (º):** -60.0
Logged by: R. Kutluoglu	**Length (m):** 297.50
Drilled by: Driftwood Drilling	**Horizontal Projection:**
Assayed by: ALS Chemex	**Vertical Projection:**
Core Size: BTW	
Date Started: 2007/06/05 **Date Completed:** 2007/06/05	
Dip Tests By: Acid Test	
Objective This hole will test the northern extents of the soil grid where the Au and Cu values are still anomalous as well as test The Boundary Zone (11.3 g/t Au over 3.4m) to check its expression at depth.	

Summary Log:

0.0 - 3.05 Casing
3.05 - 35.10 Andesite Flow
35.10 - 39.80 Andesite Breccia
39.80 - 56.50 Andesite Flow
69.90 - 81.00 Andesite
56.50 - 69.90 Granodiorite
81.00 - 183.05 Granodiorite
183.05 - 188.85 Breccia - Intrusive (?)
188.85 - 230.60 Granodiorite
230.60 - 234.15 Lamprophyric Dyke
234.15 - 297.50 Andesite Flow
297.50 - EOH

Downhole surveys:

Depth	Dip	Azimuth
0.00	-60.00	110.00
159.60	-62.00	110.00
297.50	-65.00	110.00

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype** **& Description**

0.00 3.05 CASN

Minimal recovery, broken fragments of green fine-grained grained andesite with minor narrow weakly rusty fractures.

3.05 35.10 ANDS

Dark green to light green, fine-grained grain to very fine-grained grain. Highly fractured and rubbled core in upper portion of interval, fracture density >30/m. Moderate to strong semi-pervassive silica alteration. Diffuse wispy veins of silica flow within the unit.

« 3.05- 5.40 Hematite 3.00%» Rusty fractures.

« Pyrite 0.50%» Trace fine-grained grained disseminated pyrite.

« 3.05- 23.70 Silicification 3.00-4.00*» Pervassive to semi-pervassive silica alteration, with wispy variable shallow angle to core banding of more intense silica alteration.

« Chlorite 1.00*» Pervassive but weak chlorite alteration to groundmass.

« 11.30- 14.40 Magnetite 3.00-5.00% 0.20-0.50cm» Patchy blebs and stringers of magnetite with minor trace epidote veinlets.

< @ 15.30 Vein 35.00º 0.50cm > Set of banded quartz veins, late fractures creating minor offset within bands.

« 15.60- 17.00 Epidote 2.00* 0.20cm» Variable angled veinlets and stringers of fracture fill epidote.

« 16.40- 16.70 Albite 6.00%» 0.5cm parallel to core axis, veins of massive albite with late disjointing fractures, very hard, buff to tan in colour.

« 18.00- 21.00 Pyrite 1.00% 0.05cm» Very fine-grained grained disseminated pyrite mostly associated with patchy wispy epidote alteration.

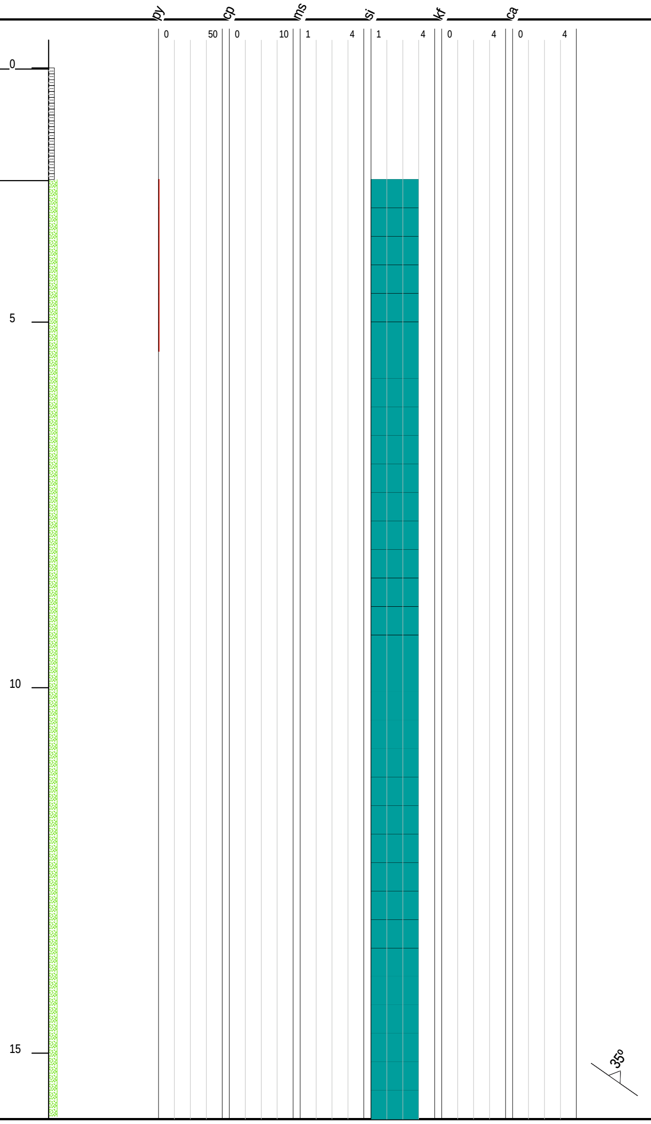
« Epidote 1.00-2.00*» Weak to moderate wispy patches and veinlets of epidote alteration.

« 21.30- 23.70 Calcite 1.00-2.00* 0.10-1.00cm» Blebs of carbonate and fracture fill veinlets. Appears to cut other textures and structures, and appears to be cross cutting veinlets of carbonate.

< @ 21.80 Shear Plane 20.00º 5.00 > Brittle rusty fractures with strong foliation and strong chlorite alteration along fracture with waxy textured fracture planes.

« 22.45- 22.55 Magnetite 35.00%» Massive fine-grained grained magnetite as wispy large clot. Minor fine-grained grained pyrite and minor hematite fracture fill.

« 23.70- 29.36 Fragmental, possible flow brecia »



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
0.00	4.88	4.88	200701	0.08	-0.20	20
4.88	7.92	3.04	200702	0.01	-0.20	23
7.92	10.97	3.05	200703	0.01	0.20	45
10.97	14.02	3.05	200704	-0.01	-0.20	79
14.02	17.06	3.04	200705	0.01	0.20	55
17.06	20.12	3.06	200706	0.01	-0.20	48
20.12	22.00	1.88	200707	0.02	0.30	216
22.00	23.70	1.70	200708	0.02	0.30	450
23.70	25.20	1.50	200709	0.04	0.40	359
23.70	25.20	1.50	200710	0.04	0.50	326
25.20	27.20	2.00	200711	0.05	0.60	608
27.20	29.26	2.06	200712	0.02	0.30	277
29.26	31.25	1.99	200713	0.03	0.40	459
31.25	33.25	2.00	200714	0.02	0.40	357
33.25	35.10	1.85	200715	0.01	0.30	278

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype & Description**

« 23.70- 25.20 Fault 40.00°» Rusty fragmented core. Brittle and very rusty with moderate chlorite alteration.

< @ 25.50 Breccia 60.00° 2.00cm > Chlorite matrix and fine-grained grained disseminated pyrite, with clast consisting of silica altered andesite.

« 25.50- 26.90 Pyrite 2.00%» Very fine-grained grained disseminated pyrite, more concentrated within epidote altered patches.

« 26.25- 26.60 Pyrite 4.00%» Coarse blebs of pyrite and greater amount of fine-grained grained disseminated pyrite.

< @ 25.70 Fault 45.00° 5.00cm > Rusty fault gouge, fine-grained grained rubbled core. Brick red and dark green.

« 25.20- 28.60 Calcite 1.00-2.00* 0.10-0.50cm» Fracture fill and blebby carbonate.

« Chlorite 1.00*»

« Silicification 2.00*»

« Epidote 1.00-2.00*» patchy fine-grained grained wispy alteration.

< @ 29.65 Vein 50.00° 1.00cm > White calcite vein with sharp contact and minor iron red oxidation.

< @ 29.75 Vein 40.00° 1.00cm > Vuggy calcite vein, with moderate chlorite and hematite along vuggy surfaces.

« 29.10- 28.60 Pyrite 3.00%» Fine grained disseminated pyrite in ground mass as well as associated with epidote alteration patches and veinlets.

« Epidote » Epidote alteration as stringers and veinlets of epidote alteration with patches of blebbly alteration.

« 29.20- 30.40 Epidote 2.00* 0.50cm» Veinlets of epidote alteration at a density of 10/m and variable thickness from mm to 2cm. Variable orientation between 25 and 60 degrees.

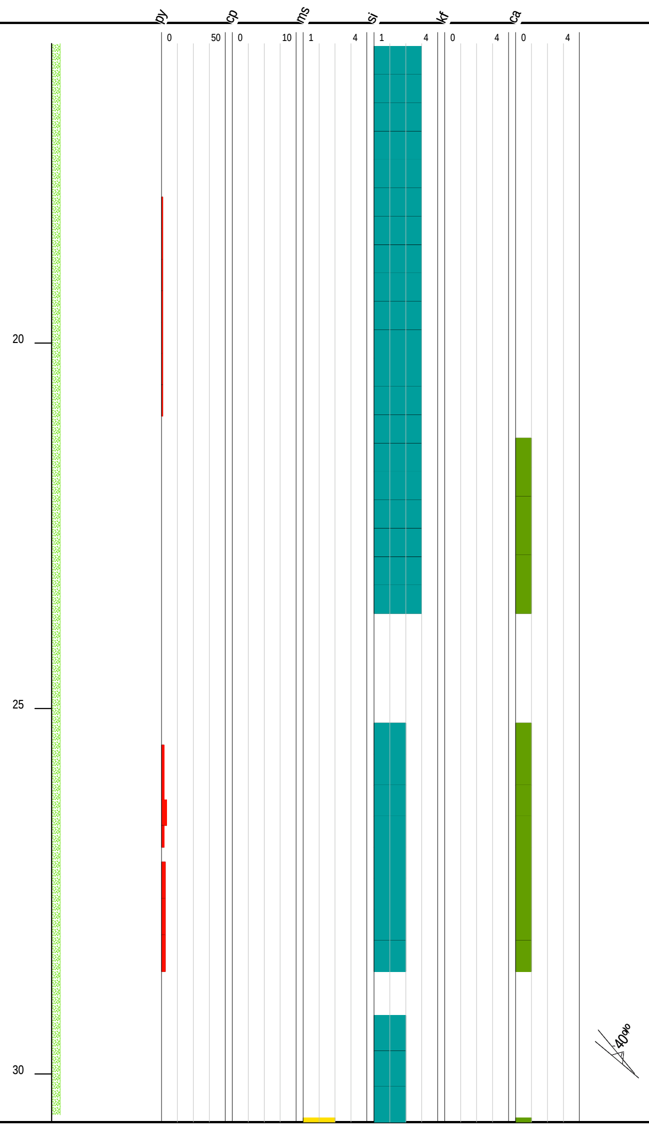
« 29.20- 37.50 Silicification 2.00*» Overprinted by epidote alteration and carbonate stringers the rock is moderately to strongly altered silica with weak to moderate chlorite alteration.

« Chlorite 1.00*»

« 30.60- 37.50 Sericite 2.00*» Veins of variable orientation between 20 and 90 degrees with moderate sericite alteration, moderate to weak epidote alteration and moderate to strong silica alteration.

« Calcite 1.00*» Very fine-grained veinlets of carbonate cutting the epidote silica sericite veins.

« 32.50- 32.70 dyke LMPR 40°» Fine grained white to buff with sharp contacts. Weakly chlorite alteration and trace fine-grained grained disseminated pyrite. Minor phyric pyroxenes (?).



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm

*graphic log not to scale

From	To	Rocktype	& Description

< @ 33.10 Vein 25.00° 0.30cm > Iron carbonate vein, pink to white.
« 32.70- 35.10 Silicification 3.00*» Banded silica alteration oriented at ~40 TCA.
< @ 35.00 Vein 60.00° 0.70cm > Carbonate and chlorite vein cutting epidote clot. Epidote containing small blebs of mafic mineral, possible actinolite or pyroxene?
« Sericite 2.00*»
« Epidote 1.00*»
« Silicification 2.00*» As veinlets and vein sets cutting core at variable angles between 0 to 90 degrees TCA.

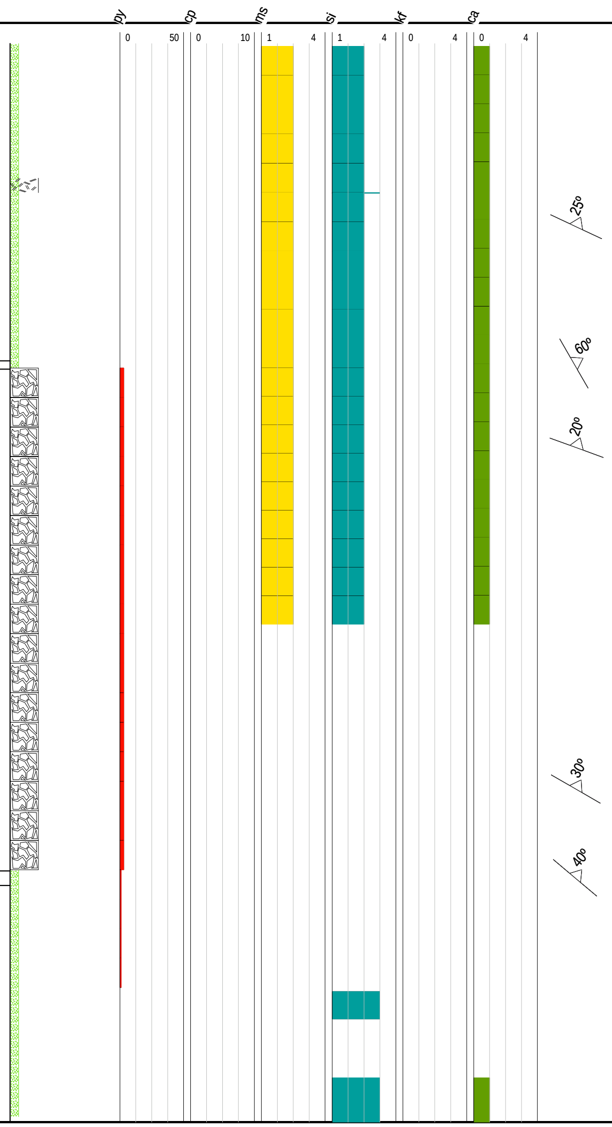
35.10 39.80 ANBX

Andesite breccia:
Variably matrix to clast supported, with monzanite porphyrytic and pyroxene pyrite mafic clasts and chloritised andesite with a fine-grained to very fine-grained grained sericite epidote silica matrix. Epidote pyrite and sericite appear to overprint the breccia as it is present crossing into clasts from the matrix.
« Pyrite 3.00%» fine-grained grain disseminated pyrite and minor medium-grained sized blebs of pyrite.
« 35.10- 35.40 Albite 50.00%» Hand specimen taken for petrology work.
< @ 35.80 Vein 20.00° 0.30cm > Very fine-grained grained hard vein cutting clasts and matrix, appears to be sericite or albite with minor chlorite (white to greenish white).
< @ 39.00 Vein 30.00° 0.40cm > Chlorite vein.
« 35.10- 39.80 Epidote 2.00-3.00*» Wispy patches of alteration and as fine-grained blebs in matrix.

39.80 56.50 ANDS

Dark green to light green, fine-grained grain to very fine-grained grain.
Moderate to strong semi-pervassive silica alteration. Diffuse wispy veins of silica flow within the unit.
Moderate to strong patchy wispy epidote alteration. Weak chlorite alteration.

< @ 39.80 Contact 60.00° 0.01cm > Hazy contact denoted by slightly darker shade of light green and abscence of fragments.
< @ 39.85 Vein 40.00° 0.30cm > Carbonate vein with minor chlorite patches, sharp contacts.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
35.10	37.10	2.00	200716	0.02	0.50	560
37.10	38.40	1.30	200717	0.02	0.40	460
38.40	39.80	1.40	200718	0.04	0.60	730
39.80	41.80	2.00	200719	0.03	0.50	580
41.80	41.80	0.00	200720	0.03	0.50	389
41.80	43.75	1.95	200721	0.02	0.30	277
43.75	45.70	1.95	200722	0.02	0.30	273
45.70	47.50	1.80	200723	0.04	0.60	639
47.50	49.50	2.00	200724	0.03	0.60	573
49.50	51.50	2.00	200725	0.05	0.60	635
51.50	53.60	2.10	200726	0.06	0.60	653
53.60	55.00	1.40	200727	0.15	0.90	1000
55.00	56.50	1.50	200728	0.05	0.70	795

*graphic log not to scale

From **To** **Rocktype & Description**

« 39.80- 41.30 Epidote 1.00*» Small blebby alteration and minor wispy veinlets.

« Pyrite 1.00%» Trace fine-grained grained disseminated pyrite.

« 41.35- 41.70 Silicification 3*» Band of flood silica, green and very competent, upper contact 50 TCA, lower 75 TCA.

« 42.45- 43.90 Silicification 3.00*» Flood silica alteration, weak chlorite and weak carbonate alteration.

« Calcite 1.00*»

« Epidote 1.00*»

« Chlorite 1.00*»

« 43.70- 45.00 Calcite 3.00* » Wavy veinlets of carbonate and blebs.

« Pyrite 3.00%» Very fine-grained grained disseminated pyrite.

« 45.00- 46.30 Epidote » Moderate wispy patches of epidote alteration with minor trace fine-grained grainded disseminated pyrite.

« Pyrite 1.00%»

« Silicification 3.00*»

« 47.00- 47.50 Pyrite 6.00%» Fine vein set at variable angles between 0 and 60 degrees TCA. The veins are narrow and also contain chlorite.

< @ 48.90 Vein 10.00º 1.00cm > Carbonate vein with minor chlorite and sharp contacts.

< @ 49.90 Vein 50.00º 0.50cm > Carbonate vein.

< @ 50.50 Shear Plane 50.00º 4.00 > Strongly foliated with moderate silica alteration and a veinlet of pyrite at the centre of the shear.

« 50.60- 52.80 Pyrite 4.00%» Very fine-grained grain disseminated pyrite and wispy veinlets of mm scale.

« 51.10- 53.30 Silicification 3.00*» Flood silica.

« Calcite 1.00*» Wispy veinlets cutting flood silica.

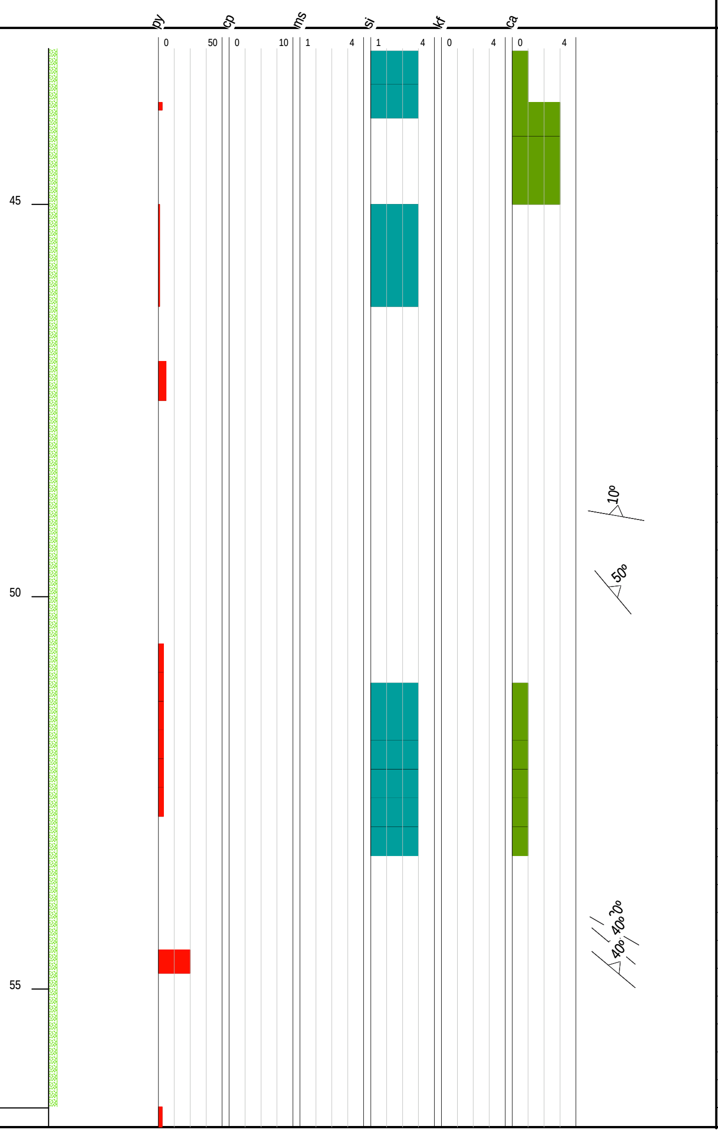
< @ 54.20 Vein 30.00º 2.00cm > Fine to medium-grained fine-grained grain vein.

< @ 54.40 Vein 40.00º 0.30cm > Carbonate vein with moderate chlorite content.

< @ 54.70 Vein 40.00º 1.00cm > Chlorite carbonate vein cutting blebby semi-massive pyrite.

« 54.50- 54.80 Pyrite 25.00%» As blebs of fine-grained to medium-grained grain pyrite.

56.50 69.90 GRDR



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
56.50	58.50	2.00	200729	0.01	-0.20	128

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype & Description**

Dark green to very dark green. Fine medium-grained grained, with feldspars quartz and horneblende. Weak foliation noted in some portions of the interval @ 40 TCA. minor fine-grained grained trace pyrite.

< @ 56.50 Contact 50.00° > Sharp contact with minor clay fracture fill. Upper portion of unit appears almost phyric with plagiclase lathes and minor blebby pyrite.

< @ 59.70 Vein 30.00° 2.00cm > Carbonate quartz vein.

« 56.50- 65.80 Pyrite 3.00%» Wispy fine-grained grain disseminated and veinlets of pyrite. Additional small blebs of medium-grained grain pyrite . Variable angle to the core between 10 and 75 TCA.

« 59.60- 63.30 Silicification 3.00*» Strong silica alteration overprinting of primary texture. Cut by late carbonate and fine-grained veinlets of pyrite.

« 65.90- 67.20 Pyrite 7.00%» As medium-grained coarse-grained blebs of pyrite, weakly aligned to 40 TCA.

« 67.20- 69.90 Pyrite 2.00%» Mainly fine-grained grained disseminated pyrite with minor rare blebs of medium-grained grained pyrite.

69.90 81.00 ANDS

Light to dark green, fine-grained to medium-grained grained. Higly fractured with density >/= to 30/m. Millimetre scale veinlets of epidote, density 10/m, variable angle TCA. Moderate to weak chlorite alteration. Minor patchy carbonate veinlets and blebs.

< @ 70.80 Vein 30.00° 0.30cm > Quartz carbonate vein with minor chlorite along contacts.

< @ 72.40 Vein 55.00° 2.00cm > Pyrite quartz vein. Fine grained pyrite with quartz core.

< @ 72.50 Vein 15.00° 3.00cm > Broad alteration vein of epidote sericite chlorite and silica. Wispy, undulatory contacts. Chloritic phase segregated (later?) bands, with whole vein silica altered. Trace disseminated pyrite.

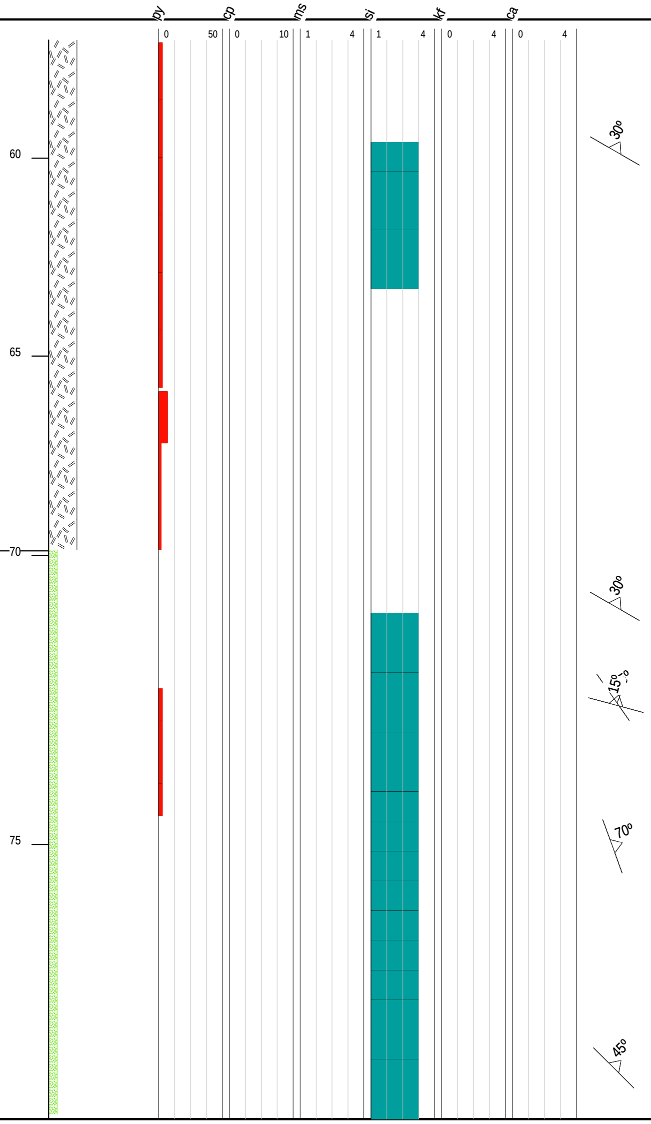
« 72.30- 74.50 Pyrite 3.00%» As very fine-grained grained disseminataed pyrite and as wispy stringers within epidote veins.

< @ 75.00 Vein 70.00° 0.10cm > Narrow pyrite vein of fine-grained grained pyrite and epidote.

< @ 76.00 Breccia 40.00° 35.00cm > Magnetite matrix supported breccia with minor trace pyrite and flow and crackle texture.

« 71.00- 79.95 Silicification 3.00*» Very compitent core, slightly milky bands of alteration.

< @ 78.80 Vein 45.00° 0.50cm > Small carbonate vein pair, with minor



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
56.50	58.50	2.00	200730	0.01	0.20	98
58.50	60.50	2.00	200731	0.11	1.10	1590
60.50	62.50	2.00	200732	0.22	2.10	3720
62.50	64.00	1.50	200733	0.03	-0.20	249
64.00	65.80	1.80	200734	0.21	2.20	4130
65.80	67.20	1.40	200735	0.04	-0.20	219
67.20	69.90	2.70	200736	0.04	-0.20	229
69.90	71.90	2.00	200737	0.08	0.60	1415
71.90	73.90	2.00	200738	0.39	0.80	1820
73.90	76.00	2.10	200739	1.04	1.20	2730
73.90	76.00	2.10	200740	0.53	1.10	2610
76.00	78.00	2.00	200741	0.54	1.00	2660
78.00	79.50	1.50	200742	0.21	0.70	2230
79.50	81.00	1.50	200743	0.10	0.50	1520

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype** **& Description**

chlorite and darkened alteration halo.

81.00 183.05 GRDR

Green to dark green, mottled textured. Fine to medium-grained grain with minor phyric Mafic mineral. Moderately magnetic with patchy sweats and veinlets of epidote alteration. Minor pyrite sweats.

« Pyrite 2.00%» Fine grained disseminated pyrite and minor clots and sweats of pyrite.

« Magnetite 5.00-10.00%»

« Epidote 1.00-2.00*»

« Calcite 1.00*»

< @ 81.75 Vein 65.00º 10.00cm > Strong silica alteration.

< @ 82.00 Vein 30.00º 5.00cm > Fine grained epidote and pyrite patch with sharp contacts and minor hematite staining.

« 82.00- 82.20 Hematite 15.00%» Hematization as fracture fill along multiple fractures, minor clay component, possible small shear zone.

< @ 82.25 Vein 40.00º 0.01cm > Vein set of iron carbonate.

< @ 83.50 Vein 35.00º 1.00cm > Wispy irregular carbonate vein with minor trace chalcopyrite. « Chalcopyrite 0.01%»

< @ 83.65 Vein 30.00º 0.50cm > Carbonate vein.

« 87.40- 88.50 Epidote 3.00*»

« Sericite 1.00*»

« Chlorite 1.00*»

« Silicification 2.00*»

« Calcite 1.00*»

Altered patch with minor and narrow quartz veins and carbonate veins and blebs cutting at 15 to 45 TCA. More pervasive silica and epidote alteration with minor sericite within the altered area.

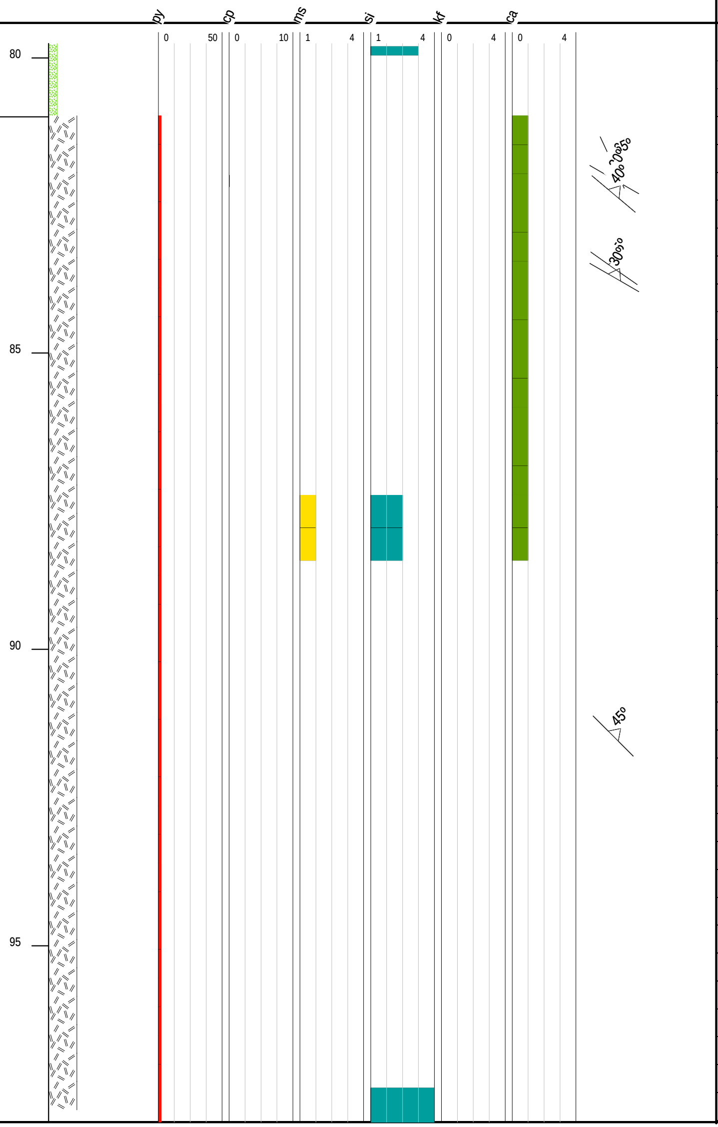
« 89.20- 89.90 Epidote 2.00*» Narrow bands of epidote alteration oriented 25 TCA, possible flow texture or foliation.

< @ 91.40 Vein 45.00º 8.00-10.00cm > Carbonate chlorite quartz vein with vuggy section with possible actinolite coated with carbonate. as small fiberous rectangular lathes.

« 91.50- 97.40 Magnetite 5.00%» Weak to moderate magnetite content as fine-grained grains in ground mass.

« 97.40- 98.20 Silicification 4.00*» Flood silica segment , grey, moderate to heavily fractured rock. Disseminated pyrite and rusty fracture planes. Minor carbonate. No primary textures.

< @ 99.00 Vein 60.00º 1.00cm > Quartz vein with possible k-spar alteration



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
81.00	83.00	2.00	200744	0.02	-0.20	244
83.00	85.00	2.00	200745	-0.01	-0.20	52
85.00	87.20	2.20	200746	-0.01	-0.20	48
87.20	89.20	2.00	200747	-0.01	-0.20	54
89.20	91.20	2.00	200748	-0.01	-0.20	49
91.20	93.30	2.10	200749	0.01	-0.20	50
93.30	93.30	0.00	200750	0.01	-0.20	98
93.30	95.40	2.10	200751	-0.01	-0.20	44
95.40	97.40	2.00	200752	0.02	-0.20	137
97.40	98.20	0.80	200753	0.16	1.10	570
98.20	100.20	2.00	200754	0.02	0.40	427
100.20	102.40	2.20	200755	0.24	1.80	2870
102.40	104.40	2.00	200756	0.08	0.70	1255
104.40	106.40	2.00	200757	0.13	0.60	1625
106.40	108.50	2.10	200758	0.10	0.20	714
108.50	110.50	2.00	200759	0.11	0.30	896
108.50	110.50	2.00	200760	0.08	0.20	731
110.50	112.50	2.00	200761	0.04	0.30	818
112.50	114.60	2.10	200762	0.07	0.40	995
114.60	116.60	2.00	200763	0.02	0.20	694
116.60	118.40	1.80	200764	0.03	0.20	591
118.40	121.00	2.60	200765	0.03	0.30	584
121.00	123.00	2.00	200766	0.04	0.20	808
123.00	124.90	1.90	200767	0.04	0.80	1350
124.90	126.80	1.90	200768	0.04	0.60	1155
126.80	129.30	2.50	200769	0.03	0.20	463
129.30	129.80	0.50	200771	0.02	-0.20	485
129.30	129.80	0.50	200770	0.03	-0.20	596
129.80	132.30	2.50	200772	0.03	0.30	579
132.30	135.75	3.45	200773	0.08	1.10	1840
135.75	136.90	1.15	200774	0.05	0.50	612
136.90	139.00	2.10	200775	0.05	0.40	441
139.00	142.00	3.00	200776	0.06	1.20	2100
142.00	143.50	1.50	200777	0.05	0.40	728
143.50	145.10	1.60	200778	0.07	0.20	248
145.10	146.60	1.50	200779	0.05	0.30	460
146.60	146.60	0.00	200780	-0.01	-0.20	3

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype** **& Description**

along contact. Trace fine-grained to medium-grained grain disseminated pyrite along contacts and somewhat brecciated within the quartz vein. Carbonate and hemtite vein cutting quartz and pyrite.

< @ 100.85 Vein 25.00º 0.50cm > Carbonate vein with minor chlorite on contacts.

< @ 102.00 Vein 70.00º 0.70cm > Pyrite vein with k-spar alteration and silica sericite alteration.

« 101.80- 102.10 Silicification 2.00*»

« Sericite 1.00*»

« K-feldspar 2.00*»

< @ 102.35 Vein 40.00-90.00º 0.20cm > Carbonate vein.

< @ 102.45 Vein 65.00º 5.00cm > Quartz vein with pyrite intermingled and along contacts as fine-grained to medium-grained grain blebs. Alteration halo of k-spar and sericite.

< @ 103.50 Breccia 70.00-30.00º 20.00cm > Magnetite breccia with 3% fine-grained grained disseminated pyrite

« 103.50- 103.80 Pyrite 3.00%»

« 105.20- 105.70 Pyrite 5.00%» Fine grained wispy disseminated pyrite.

« 105.70- 108.50 Pyrite 1.00%» Fine grained wispy veinlets and small blebs.

« Chlorite 1.00*»

« Silicification 2.00*» Patchy silica alteration.

« K-feldspar 1.00*» Small clots of k-spar alteration.

« Epidote 2.00*»

< @ 108.20 Vein 90.00º > Carbonate chlorite vein and blebs.

« 108.50- 109.70 Pyrite 3.00%» Associated with magnetite an as fine-grained grain disseminated pyrite in ground mass.

< @ 109.60 Vein 30.00º 0.60cm > Magnetite vein with minor blebs of pyrite.

« 110.70- 110.90 Silicification 3.00*» Very fine-grained and pervasive silica alteration.

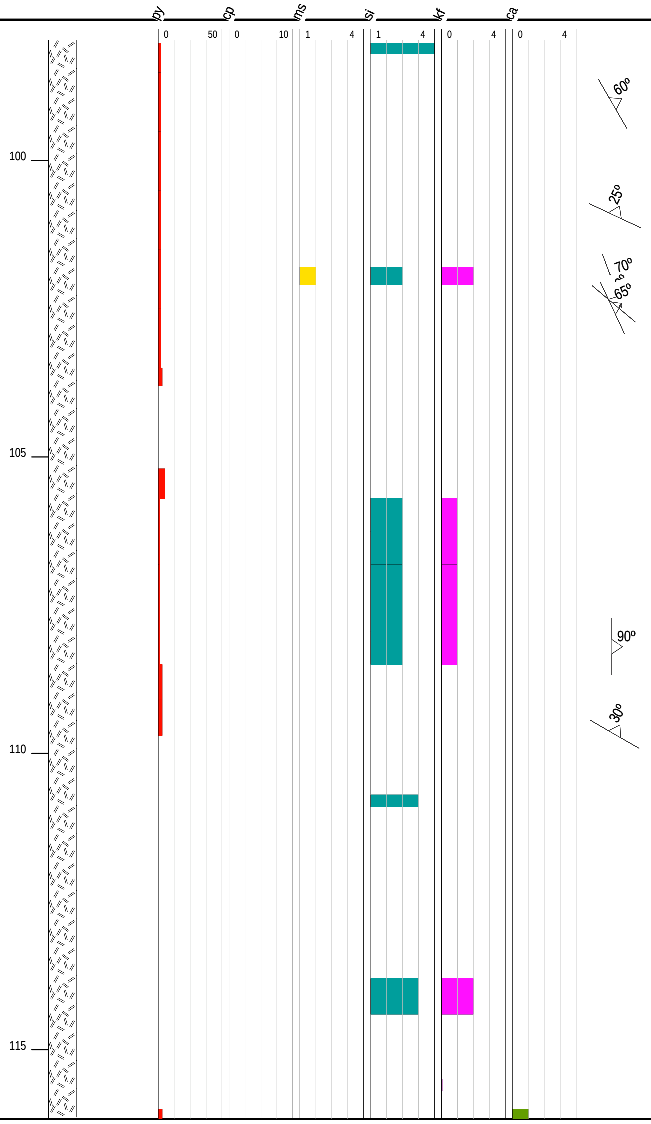
« 112.30- 113.20 Magnetite 5.00%» Patchy moderate magnetic responce.

« 113.80- 114.40 Silicification 3.00*» « K-feldspar 2.00*» Patches of potassic and silica alteration.

« 115.50- 115.70 K-feldspar » Orange pink patch of alteration with epidote vein intermingled.

« 116.00- 117.00 Pyrite 3.00%» Fine grained disseminated, small veinlets and as small blebs.

« Calcite 1.00*» Small veinlets variable angle to the core.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
146.60	148.10	1.50	200781	0.03	-0.20	120
148.10	151.20	3.10	200782	0.04	-0.20	209
151.20	154.00	2.80	200783	0.03	0.20	335
154.00	155.30	1.30	200784	0.03	-0.20	150
155.30	157.30	2.00	200785	0.03	0.20	429
157.30	157.80	0.50	200786	0.02	0.50	2290
157.80	160.00	2.20	200787	0.01	-0.20	231
160.00	162.00	2.00	200788	0.03	0.20	321
162.00	164.00	2.00	200789	0.01	-0.20	252
164.00	164.00	0.00	200790	0.01	-0.20	248
164.00	166.40	2.40	200791	0.01	-0.20	107
166.40	168.40	2.00	200792	0.03	-0.20	105
168.40	169.50	1.10	200793	0.02	-0.20	54
169.50	171.50	2.00	200794	0.03	0.20	311
171.50	173.50	2.00	200795	0.03	0.40	316
173.50	175.60	2.10	200796	0.01	-0.20	176
175.60	177.10	1.50	200797	0.03	0.30	474
177.10	178.60	1.50	200798	0.06	1.00	476
178.60	180.20	1.60	200799	0.01	0.20	206
178.60	180.20	1.60	200800	0.01	-0.20	201
180.20	181.70	1.50	200801	0.01	-0.20	235
181.70	183.05	1.35	200802	-0.01	-0.20	37

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype & Description**

« 116.50- 116.80 Chalcopyrite 0.01%» Ultra rare fine-grained grains of disseminated chalcopryite.

< @ 117.75 Vein 75.00º 0.50cm > Carbonate chlorite vein.

« 117.25- 117.80 Pyrite 3.00%»

« 117.80- 121.00 Silicification 3.00-4.00*» Gradational upper contact and sharper lower contact @ 50 TCA.

< @ 119.70 Vein 90.00º 1.00cm > Wispy fine-grained grained pyrite.

< @ 120.10 Vein 35.00º 0.70cm > Quartz vein, translucent white.

< @ 120.60 Vein 80.00º 0.10cm > Narrow veinlets in stockwork fashion consisting of pyrite and disseminated pyrite around the veins.

« 121.00- 121.60 Epidote 2.00-3.00*» Large patch of epidote alteration and vein sets @ 45 TCA.

« 122.10- 122.30 K-feldspar 2.00*» Patchy potassic alteration denoted by pinkish orange and hardness.

« 122.30- 122.90 Pyrite 2.00%» Patches of blebby pyrite and minor mm scale stringers.

< @ 124.00 Vein 60.00º 0.10cm > Pyrite and chalcopyrite with epidote on contacts.

« 124.90- 125.70 ANBX » Variable clast size, matrix supported, possible flow breccia.

« 126.50- 126.80 Magnetite 5.00%»

« 126.90- 128.90 Silicification 3.00*» Highly fractured core, variable fracture angles, but strongly silica altered, moderate to weak chlorite alteration.

« Chlorite 2.00*»

« 129.30- 129.78 Sericite 2.00*» Sericite alteration centred around a strongly bleached and vuggy segment @ 129.60m which contains pyrite and minor trace chalcopyrite with blue oxidation on the surface of some chalcopyrite grains.

« Silicification 2.00-3.00*»

« Pyrite 1.00%»

« Chalcopyrite 0.10%»

« Epidote 1.00*»

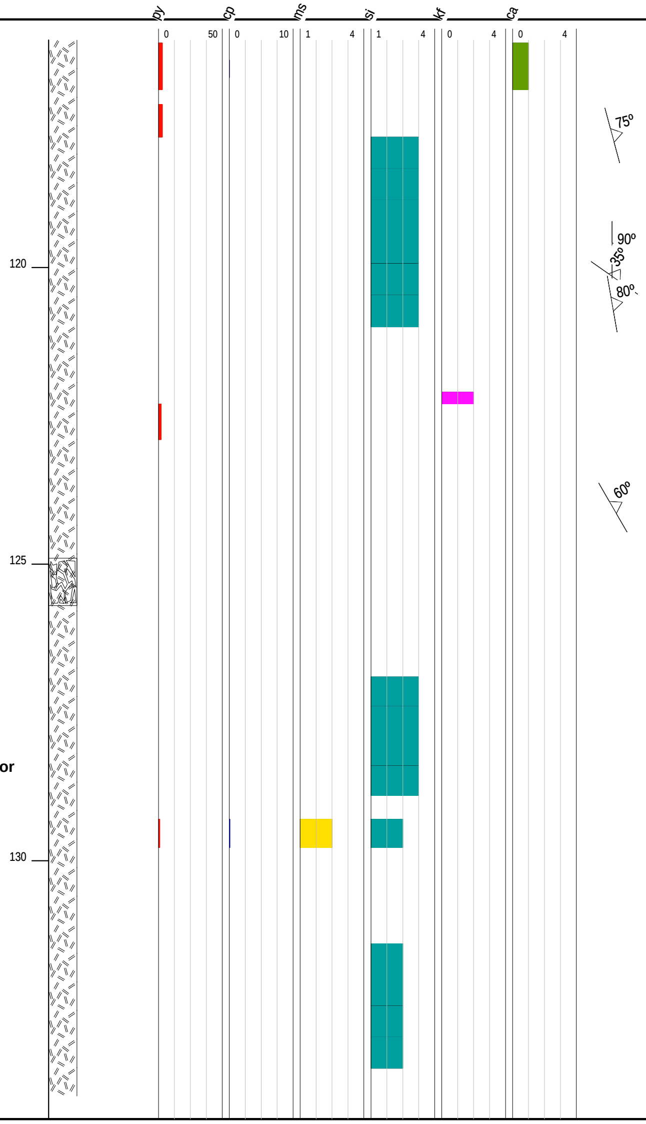
« 130.40- 130.70 Epidote 3.00-4.00*» Rubbled core with pervasive epidote alteration.

« 131.40- 133.50 Silicification 2.00-3.00*»

« Chlorite 2.00*»

« Epidote 1.00*»

« 135.70- 136.15 Silicification 3.00*» Lower contact oriented @ 60 TCA.

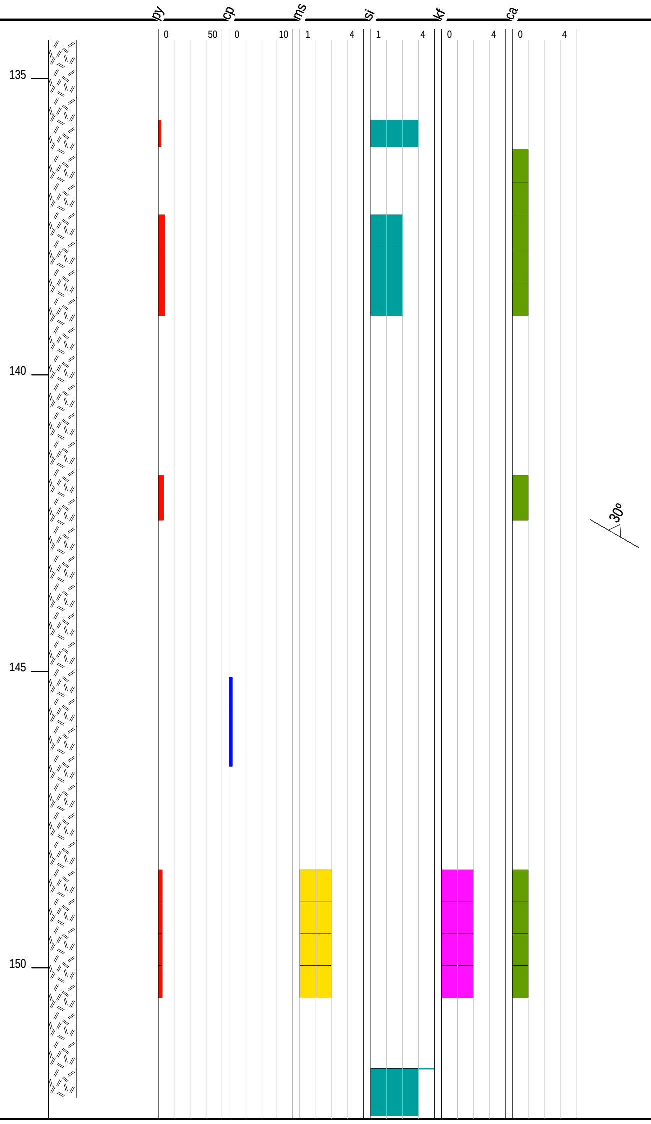


From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm

*graphic log not to scale

From **To** **Rocktype** **& Description**

« Pyrite 2.00%» « Magnetite 6.00%» Spotty magnetite mineralization.
« 136.20- 139.00 Calcite 1.00*» Small veinlets and stringers.
« 137.30- 139.00 Pyrite 5.00-7.00%» Pyrite as fine-grained disseminated grains, or as wispy veinlets. Disseminated in greater amounts in the epidote altered portion.
« Silicification 2.00-3.00*»
« 137.60- 138.80 Epidote 2.00-3.00*»
« 139.00- 141.70 Epidote 2.00*»
« 141.70- 142.45 Pyrite 4.00%» Wispy blebs and fine-grained disseminated grains with small mm scale wispy veinlets.
« Calcite 1.00*» Minor iron carbonate.
« Magnetite 10.00%»
< @ 142.60 Vein 30.00° 0.50cm > Carbonate vein cutting magnetite bleb.
« 145.10- 146.60 Chalcopyrite 0.50%» Small blebs and trace disseminated chalcopyrite in a more dioritic looking segment, as well as within carbonate chlorite veins running at variable angles TCA between 0 and 40.
« 148.35- 150.50 K-feldspar 2.00-3.00*»
« Sericite 2.00*»
« Epidote 1.00-2.00*»
« 148.35- 150.50 Pyrite 3.00%» As disseminated blebs and part of epidote patches.
« Calcite 1.00*» As small veinlets predominantly oriented around 30 to 40 TCA.
« 151.70- 152.50 Silicification 4.00*» Flood silica, weakly foliated 40 TCA.
« 151.70- 152.50 Silicification 3.00-4.00*» Strong silica alteration, glassy and dark grey.
« 154.00- 155.30 Silicification 3.00-4.00*» Strong silica alteration glassy dark grey with large cm scale blebs of wispy elongated pyrite, appears to have a weak foliation or flow banding oriented @ 20 TCA.
« Pyrite 5.00%»
« 155.30- 156.20 Pyrite 3.00%» As medium-grained sized blobs of pyrite and fine-grained wispy diseminated grains.
< @ 157.60 Vein 50.00° 1.00cm > Carbonate vein with medium-grained bleb of chalcopyrite. Medium to coarse-grained grains and slightly vuggy, open space filing vein.
« 157.50- 157.80 Chalcopyrite 2.00%»
< @ 158.60 Breccia 80.00° 30.00cm > Small pocket of matrix dominated breccia, very fine-grained grain quartz and mafic mineral with epidote and



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype** **& Description**

silica altered clasts. Clast size 1 to 2 cm.

< @ 158.90 Vein 25.00° 0.70cm > Glassy grey quartz vein with trace fine-grained grain euhedral cubic pyrite.

< @ 158.95 Vein 70.00° 1.00cm > Glassy grey quartz vein with trace fine-grained grain euhedral cubic pyrite. Joining with previous vein.

« 160.50- 160.70 K-feldspar 1.00-2.00*» Weakly pink to orange pink, very competent but fractured core. Epidote veins cutting at variable steep angles.

« 160.80- 161.30 Pyrite 8.00%» Wispy elongate large blebs of pyrite oriented @ 60 TCA within epidote veins.

< @ 161.80 Breccia 20.00cm > Fine grained dark green matrix with diorite to granodirote clasts. Matrix supported, cut by epidote veins @ ~ 20 to 50 TCA with fine-grained disseminated pyrite in the epidote veinlets as well as in the matrix of the breccia. Both are then cut by small sub mm veinlets of carbonate @ 50 TCA and also contains pyrite.

« 163.10- 163.70 Silicification 4.00*» Strong silica alteration with coarse-grained blebs of pyrite at upper gradational contact.

< @ 164.95 Contact 40.00-15.00° > Sharp alteration boundary with variable angle lower portion more epidote rich alteratoion, upper is moderate to strong silica alteration.

« 165.20- 165.40 Magnetite 5.00-7.00%» Weak to moderately magnetic slightly purple hue with weak carbonate alteration and fine-grained grain disseminated pyrite.

« Calcite 1.00*» « Pyrite 3.00%»

< @ 165.65 Vein 40.00° > Medium grain vuggy carbonate vein.

< @ 166.90 Vein 45.00° 1.00cm > Fine grain carbonate vein with minor chlorite

< @ 167.55 Vein 60.00° 1.00cm > Carbonate quartz vein with no visible sulphides.

< @ 167.60 Vein 55.00° 0.50cm > Carbonate quartz vein.

< @ 167.70 Vein 60.00° 0.20cm > Carbonate chlorite vein.

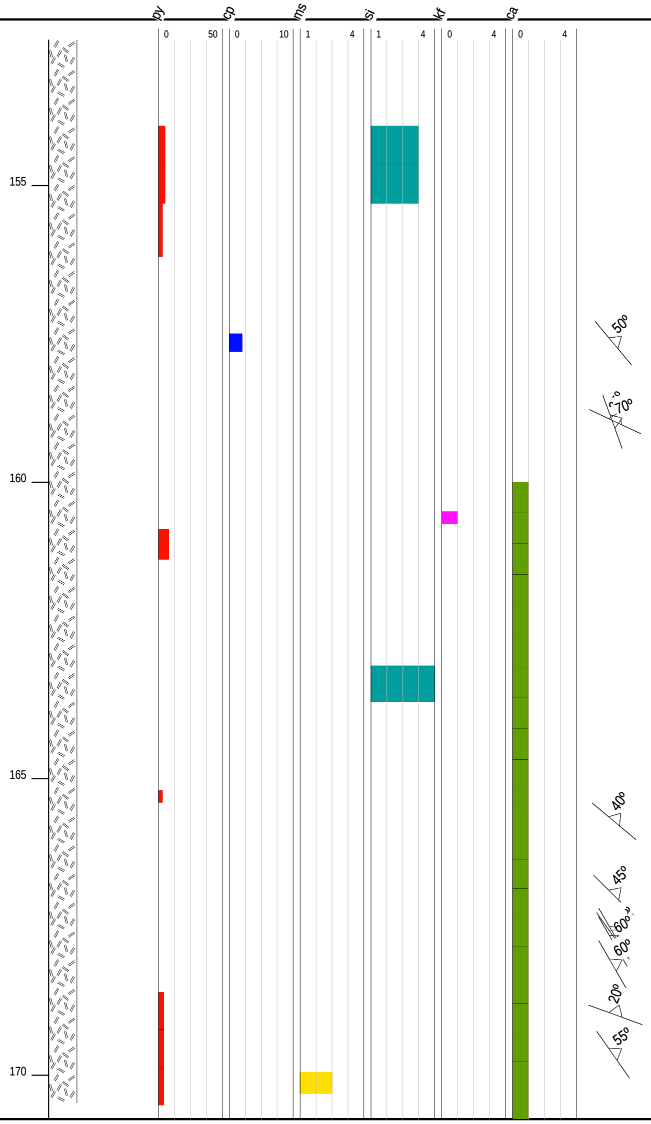
< @ 168.10 Vein 60.00° 1.00cm > Carbonate quartz vein.

< @ 168.90 Vein 20.00° 1.00cm > Glassy white quartz vein with trace fine-grained disseminated pyrite.

< @ 169.60 Vein 55.00° 1.00cm > Medium grain euhedral calcite vein.

« 168.60- 170.50 Pyrite 4.00%» Fine wispy diseseminated pyrite.

« 169.95- 170.30 Albite 3.00%» « Sericite 2.00*» Strongly bleached core with minor hematite veins.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype** **& Description**

< @ 170.30 Fault 70.00° 3.00cm > Clay fault gouge. sharp contacts.

< @ 171.90 Vein 5.00° 0.50cm > Glassy grey white quartz vein with trace medium-grained grain euhedral cubic pyrite.

< @ 172.50 Vein 5.00° 1.00cm > Vein oriented parallel to core for 20cm , glassy grey quartz with trace to 2% interstitial euhedral to subhedral cubic pyrite grains.

« 160.00- 175.40 Calcite 1.00*» As small sub mm scale veinlets of variable angle, high density of fracture fill type carbonate.

< @ 173.30 Vein 85.00° 0.50cm > Vein set of three quartz veins, located in small 15 cm zone of potassium feldspar alteration.

« 173.20- 173.35 K-feldspar 2.00*»

« 176.20- 177.05 K-feldspar 2.00*» « Silicification 2.00*» « Sericite 1.00*» « Epidote 1.00-2.00*»

Alteration with gradational contacts and fine-grained grain disseminated pyrite. « Pyrite 3.00%»

< @ 176.80 Fault 60.00° 0.20cm > Clay rich weakly carbonate altered segment of core with crackle breccia texture along contacts and hematite matrix in narrow 5cm interval.

« 177.00- 183.50 Epidote 2.00*»

« Sericite 1.00*»

« K-feldspar 1.00*» Small pods and patchy alteration.

« Calcite 1.00*» Small spaced fracture fill veinlets.

« Pyrite 2.00%» Very fine-grained grain disseminated cubes .

< @ 180.21 Vein 70.00° 1.00cm > Quartz with fine-grained interanal pyrite and fragmented contact.

183.05 188.85 BRXV

Dark green fine-grained grain with pebble sized clasts of quartz diorite and granodiorite. Moderately chloritised. Rounded to subrounded clasts of variable size between 1 and 5 cm.

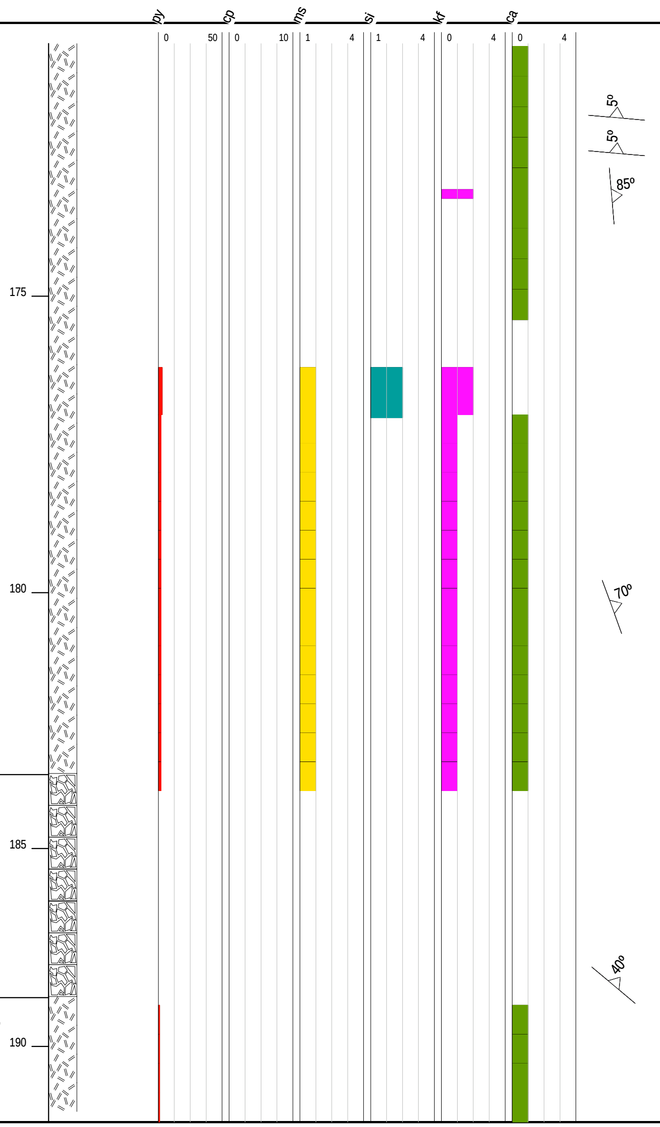
Fine grain disseminated pyrite, late stringers of carbonate and epidote lateration.

< @ 188.45 Vein 40.00° 1.50cm > Quartz carbonate pyrite vein.

188.85 230.60 GRDR

Green and dark green, mottled textured, medium-grained to fine-grained grained. moderate to strong silica epidote alteration, weak chlorite carbonate and spotty patches of potassium alteration.

« 188.85- 189.35 Magnetite 4.00%» Spotty weak magnetite with slightly



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
183.05	184.70	1.65	200803	0.02	-0.20	65
184.70	186.70	2.00	200804	0.12	0.20	334
186.70	188.85	2.15	200805	0.04	-0.20	48
188.85	189.35	0.50	200806	0.23	0.20	210
189.35	191.20	1.85	200807	0.62	0.30	434
191.20	193.05	1.85	200808	0.01	0.20	172
193.05	194.85	1.80	200809	0.02	0.30	265
193.05	194.85	1.80	200810	0.04	0.30	335

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype** **& Description**

purple tint to interval.

« 189.05- 193.05 Epidote 2.00-3.00*» Patchy vein stockwork and pervasive epidote alteration, trace fine-grained grain disseminated pyrite. « Pyrite 1.00%»

« Hematite 2.00%» Fracture fill.

« Calcite 1.00*» Weak late microfractures of carbonate.

< @ 193.05 Breccia 75.00º 55.00cm > Matrix dominated with granodiorite and quartz fragments, weak chlorite alteration, weak late chlorite alteration, vuggy calcite vein with specular hematite pair cutting breccia.

< @ 193.25 Vein 40.00º 0.30cm > Medium grain subhedral to euhedral calcite and fine-grained grain specular hematite.

« 193.80- 194.85 Epidote 2.00-3.00*» Moderate to strong patchy epidote alteration as pervasive alteration and as fine-grained veinlets cutting at variable angles to the core.

« Silicification 1.00*»

« Pyrite 1.00%» Very fine-grained grain disseminated cubes of pyrite.

< @ 196.80 Vein 60.00º 0.70cm > Carbonate albite magnetite chalcopyrite vein, two stages, carbonate second cutting interfingering with albite and sulphide.

« 198.40- 202.20 Silicification 3.00*»

« Pyrite 3.00%» Wispy blebs of fine-grained grain pyrite to medium-grained grain anhedral pyrite.

« 202.40- 208.80 Epidote 2.00-3.00*» « Silicification 2.00*» « Calcite -1.00*» « Chlorite 1.00*» « Pyrite 3.00%» As fine-grained to medium-grained eudhedral to subhedral cubic pyrite.

< @ 204.80 Vein 20.00º 2.00cm > Magnetite pyrite carbonate vein.

< @ 204.95 Vein 40.00º 0.50cm > Iron carbontate.

« 209.60- 210.70 K-feldspar 2.00*» Patchy pink orange alteration .

« Epidote 2.00*» As small veinlets and pervasive alteration.

< @ 210.00 Vein 35.00º 0.20cm > Carbonate magnetite vein.

< @ 210.90 Vein 50.00º 1.00cm > Carbonate chlorite albite, kfeld?

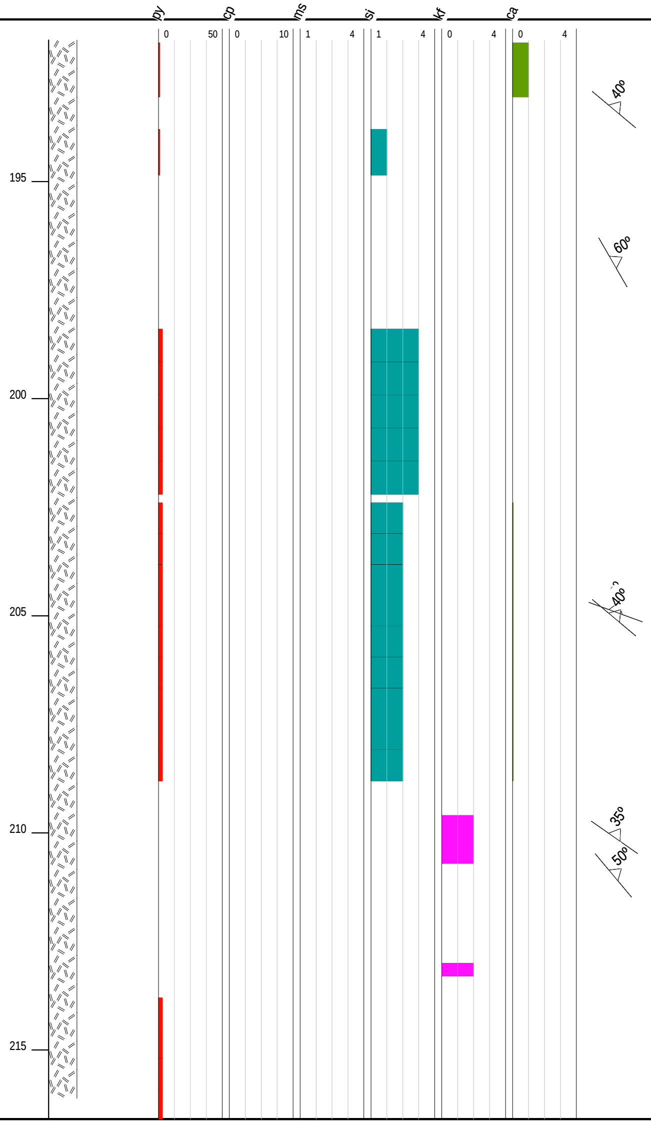
« 212.00- 213.75 Specularite 1.00%» Along fractures fine-grained grain.

« Biotite 1.00*» Along fractures fine-grained to very fine-grained grain.

« 213.00- 213.30 K-feldspar 2.00*»

« 213.80- 221.00 Pyrite 3.00%» Patchy blebs and very fine-grained disseminated grains.

« 213.00- 222.00 Epidote 2.00-3.00*» pistacio green alteration pervasive patches and whispy veinlets and veinsets, variable angle TCA.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
194.85	196.90	2.05	200811	0.04	0.30	304
196.90	198.40	1.50	200812	0.01	0.20	114
198.40	200.40	2.00	200813	0.13	0.20	117
200.40	202.40	2.00	200814	0.01	0.20	92
202.40	204.00	1.60	200815	0.02	0.20	93
204.00	206.00	2.00	200816	0.03	0.30	88
206.00	208.80	2.80	200817	0.02	-0.20	92
208.80	209.60	0.80	200818	0.01	-0.20	54
209.60	211.00	1.40	200819	0.02	0.30	221
211.00	211.00	0.00	200820	0.03	0.30	128
211.00	213.00	2.00	200821	0.02	0.20	108
213.00	215.00	2.00	200822	0.01	-0.20	66
215.00	217.00	2.00	200823	0.01	0.70	240
217.00	219.00	2.00	200824	0.01	0.30	243
219.00	221.00	2.00	200825	0.01	0.20	103
221.00	223.00	2.00	200826	-0.01	-0.20	82
223.00	225.00	2.00	200827	0.01	0.20	66
225.00	227.40	2.40	200828	0.01	0.20	85
227.40	229.15	1.75	200829	0.10	0.80	495
227.40	229.15	1.75	200830	0.08	0.60	405
229.15	229.60	0.45	200831	0.75	4.20	792

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From	To	Rocktype	& Description

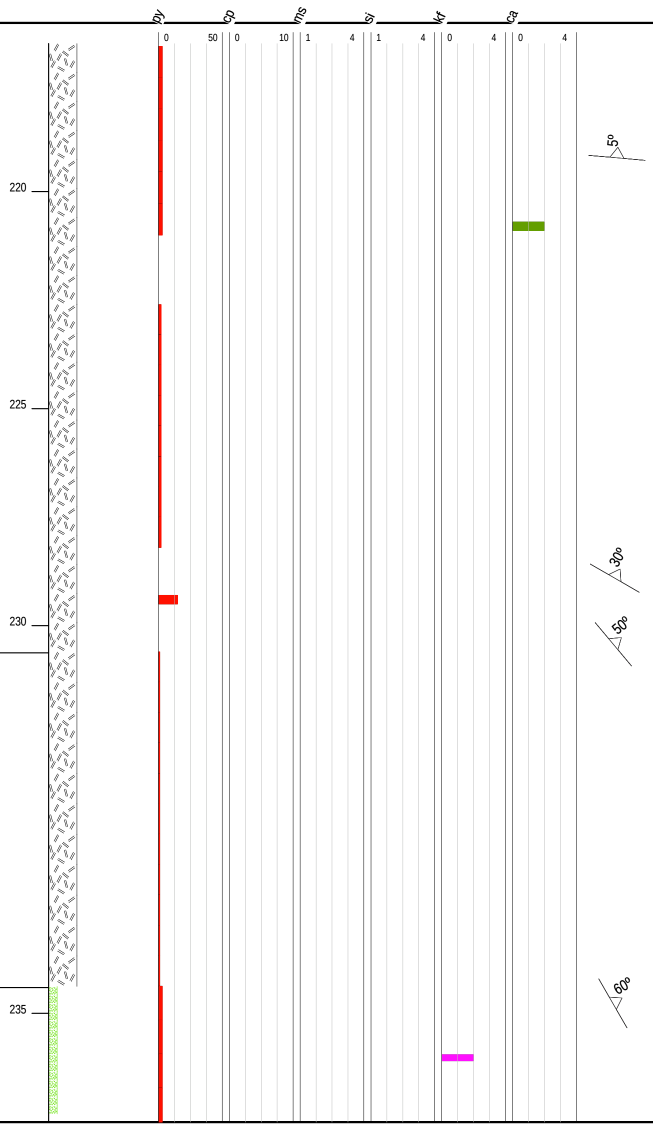
< @ 219.10 Vein 5.00º 10.00cm > Quartz vein with 5% blebby pyrite and pinkish hue to vein, possible k-feld or albite?
« 220.70- 220.90 Calcite 2.00*» « Epidote 3.00*»
< @ 222.00 Breccia 90.00º 60.00cm > Magnetite matrix granodiorite fragments, well rounded clasts. Matrix dominated .
« 222.60- 228.20 Magnetite 5.00-8.00%» Wispy patchy fine-grained grain magnetite.
« Epidote 2.00*» « Pyrite 2.00%» Mainly fine-grained disseminated grains with rare patches of medium-grained blebs.
< @ 228.80 Vein 30.00º 0.30cm > Quartz pyrite magnetite vein.
« 220.00- 229.00 Hematite 2.00%» As rusty fracture coatings and specular hematite along fractured core.
< @ 230.35 Vein 50.00º 25.00cm > Glassy white quartz vein with some coarse-grained grains and overall massive texture with moderate fracture content. Upper contact undulatory and inturdes wallrock. Disseminated and blebby pyrite in upp contact and into core of vein, gradational drop off of sulphide at basal contact.
« 229.30- 229.50 Pyrite 15.00%»
sharp lower contact.

230.60 234.15 LMPR

Light grey to buff coloured. Fine grain ground mass with phyric feldspar quartz and magnetite. Magnetite exhibiting alteration coating of chlorite. Overall homogenious minor carbonate alteration along fractures.
Trace very fine-grained grain pyrite.
« Pyrite 1.00%»
< @ 230.60 Contact 50.00° > With a 20 cm chill margin denoted by a lightening of colour over the margin.
< @ 234.15 Contact 50.00° > Chill margin and contact in rubbled core, but comparable to upper contact and contact still clear.
Moderately magnetic

234.15 297.50 ANDS

Dark green to green, fine-grained grained. Overall massive with weak patchy foliation @ 45 TCA, this is also the dominant vein orientation over the interval. Patchy epidote alteration, weak carbonate alteration as very fine-grained veinlets and microfractures.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
229.60	232.30	2.70	200832	-0.01	-0.20	19
232.30	234.15	1.85	200833	0.01	-0.20	24
234.15	235.90	1.75	200834	0.03	0.30	163
235.90	237.80	1.90	200835	0.12	0.30	328
237.80	238.80	1.00	200836	0.05	0.40	347
238.80	239.80	1.00	200837	-0.01	-0.20	12
239.80	240.80	1.00	200838	0.01	0.20	37

*graphic log not to scale

Project: JW **Hole Number:** JW07-06

From **To** **Rocktype** **& Description**

« Epidote 2.00*»

« Pyrite 3.00%» As blebs and fine-grained disseminated grains.

< @ 234.60 Vein 60.00º 3.00cm > Quartz pyrite vein.

« 236.30- 236.50 K-feldspar 2.00*»

< @ 238.80 Vein 50.00º 370.00cm > 3.4m White translucent to opaque massive quartz vein with minor amounts of chlorite wisps and trace pyrite. Fragment of carbonate altered wallrock from 241.2 to 241.55m.

« 244.00- 248.40 Epidote 2.00-3.00*» « Magnetite 10.00-15.00%» Mottled banded magnetite with moderately strong epidote alteration in patches of variable size. Wispy with no clear banding orientation.

« Pyrite 1.00%» Trace very fine-grained grain pyrite, medium-grained grain along fracture planes.

« 247.90- 250.30 Chalcopyrite 1.00%» very fine-grained grain wispy disseminated chalcopyrite assoiated with stronger silica alteration and in carbonate veinlets.

« Silicification 2.00*»

« 254.60- 255.45 Magnetite 25.00%» « Chalcopyrite 1.00%» « Pyrite 1.00%» Magnetite breccia cut by carbonate veins of variable angles, but dominant angle 20 TCA. Epidote altered fragments of andesite in magnetite matrix, matrix supported brecia. Chalcopyrite and pyrite in magnetite matrix and carbonate veins.

« 259.70- 261.10 Silicification 3.00*» Strong grey glassy silica alteration with possible potassium feldspar alteration, and fine-grained calicite veins cutting parallel to 80 TCA.

Trace fine-grained grain disseminated pyrite and chalcopyrite both in the silica alteration and within the calcite veins, minor wispy magnetite with dominant amount of chalcopyrite, but still trace levels.

« Chalcopyrite 0.50%» « Pyrite 0.50%» « Calcite 1.00*» « Magnetite 5.00%»

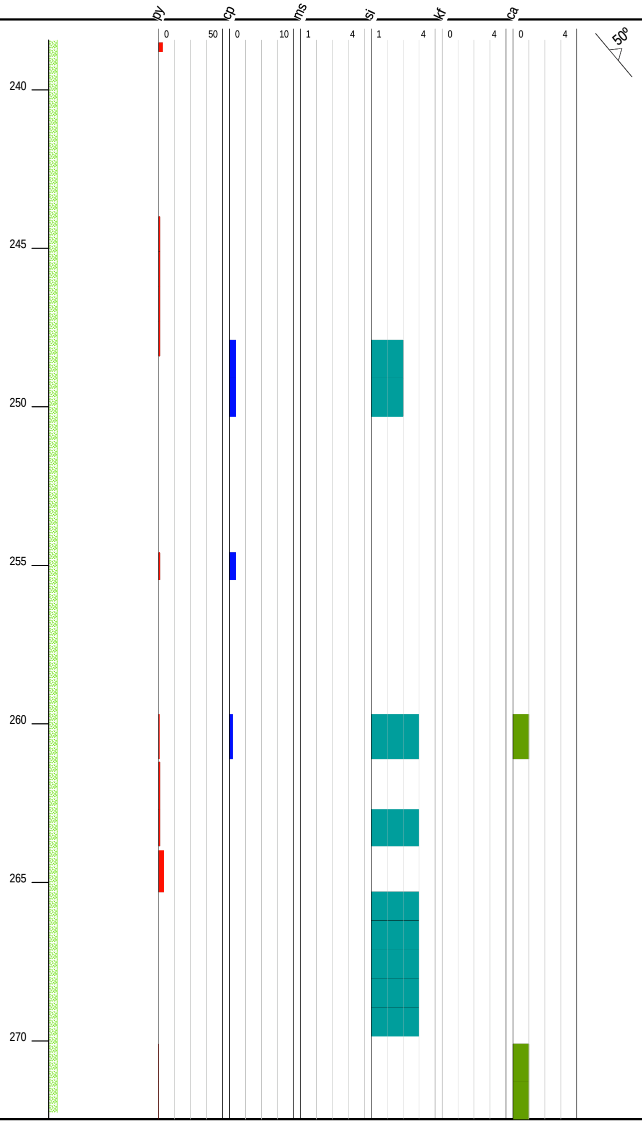
« 261.20- 262.70 Epidote 2.00*» « Magnetite 5.00%»

« Pyrite 1.00%»

« 262.70- 263.85 Silicification 3.00*» Weakly brecciated (vein sets) of variable shallow angle magneite with trace pyrite. « Pyrite 1.00%» « Magnetite 3.00%»

« 264.00- 265.30 Pyrite 4.00%» Very fine-grained to fine-grained grain disseminated pyrite.

« 265.30- 269.85 Silicification 3.00*» Grey to buff flood silica, spotty weak foliation @ 35 TCA. Trace fine-grained grain wisps of pyrite. Minor magnetite, weak carbonate alteration as microfracture fill and small mm scale



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
240.80	241.80	1.00	200839	-0.01	-0.20	12
240.80	241.80	1.00	200840	0.01	-0.20	18
241.80	243.00	1.20	200841	0.04	3.20	163
243.00	245.05	2.05	200842	0.03	0.30	150
245.05	246.55	1.50	200843	0.04	0.20	162
246.55	248.40	1.85	200844	0.03	-0.20	185
248.40	250.90	2.50	200845	0.04	0.80	1100
250.90	252.70	1.80	200846	0.02	0.70	416
252.70	254.80	2.10	200847	0.04	0.80	1260
254.80	255.45	0.65	200848	0.03	1.00	967
255.45	257.90	2.45	200849	0.03	0.30	735
257.90	257.90	0.00	200850	0.03	0.40	999
257.90	259.70	1.80	200851	0.03	0.30	619
259.70	261.20	1.50	200852	0.01	-0.20	243
261.20	263.85	2.65	200853	0.02	-0.20	336
263.85	265.30	1.45	200854	0.02	-0.20	322
265.30	267.30	2.00	200855	0.03	-0.20	461
267.30	269.85	2.55	200856	0.06	0.50	840
269.85	271.85	2.00	200857	0.02	-0.20	166
271.85	273.85	2.00	200858	0.02	-0.20	9
273.85	275.80	1.95	200859	0.01	-0.20	45
273.85	275.80	1.95	200860	0.01	-0.20	46
275.80	277.82	2.02	200861	0.02	-0.20	23
277.82	279.80	1.98	200862	0.03	0.40	537
279.80	281.80	2.00	200863	28.90	6.40	315
281.80	282.20	0.40	200864	46.70	6.50	421
282.20	284.40	2.20	200865	0.18	-0.20	24
284.40	286.40	2.00	200866	0.77	0.20	71
286.40	288.30	1.90	200867	0.05	0.30	560
288.30	290.10	1.80	200868	0.04	-0.20	58
290.10	292.10	2.00	200869	0.02	0.20	59
290.10	292.10	2.00	200870	0.03	-0.20	45
292.10	292.60	0.50	200871	0.02	-0.20	59
292.60	293.50	0.90	200872	0.02	-0.20	31
293.50	295.60	2.10	200873	0.01	-0.20	98
295.60	297.50	1.90	200874	0.07	0.30	78

*graphic log not to scale

Project: JW

Hole Number: JW07-06

From	To	Rocktype	& Description

veins.

« 270.10- 278.20 Calcite 1.00*» Minor veins from 1mm to 1.5cm of carbonate with minor chlorite cutting 50 to 70 TCA. Sub-mm scale microfracture fill at variable angles to core.

« Epidote 2.00*» Patches and veins of alteration, variable angle to the core.

« Pyrite 0.01%» Rare trace fine-grained grain disseminated pyrite.

« 280.20- 281.00 Epidote 2.00*»

« Pyrite 0.10%» Trace very fine-grained rare disseminated grains of pyrite.

< @ 281.80 Vein 50.00º 20.00cm > Irregular contacts quartz calcite and pyrite (5%) vein intermingled into wallrock.
Minor magnetite. Chloritised contacts.

« 280.20- 297.50 Chlorite 1.00*»

« 284.20- 289.90 Epidote 2.00*» Patches of alteration and vein sets variable angles TCA.

« Magnetite 5.00%» Mostly patchy veinlets and along quartz and carbonate veins at variable angles from 5 to 80 TCA.

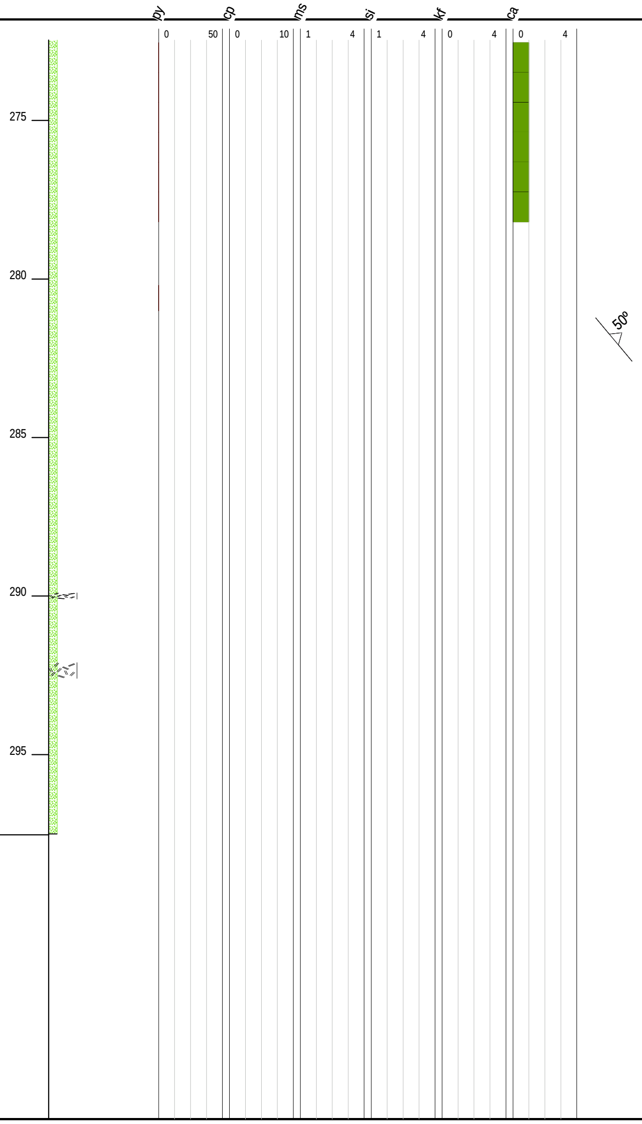
« 289.90- 290.10 MNZT » Small dyke with sharp irregular contacts. Trace pyrite and cut by calcite hematite veins, cross contacts, later emplaced.

« 292.10- 292.60 MNZT » Small dyke with sharp irregular contacts. Trace pyrite and cut by calcite hematite veins, cross contacts, later emplaced.

< @ 80.00 Contact 0° > Sharp contact with clalcite alteration to upper portion of dyke.

< @ 292.60 Contact 50.00° > Sharp lower contact with chlorite seam along contact.

297.50 297.50 EOH



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm

*graphic log not to scale

Drill Log Legend

▨ **ANBX**

▨ **ANBX**

▨ **ANDS**

▨ **BRXV**

CASING **CASN**

■ **EOH**

▨ **GRDR**

▨ **LMPR**

▨ **LMPR**

▨ **MNZT**

╲ **Vein**

EQUITY ENGINEERING LTD.

DRILL LOG

Project: JW	**Collar Elevation (m):** 362.0
Hole JW07-07	**Azimuth (º):** 90.0
Location: 6340418 m North 343140 m East	**Dip (º):** -50.0
Logged by: R.Kutluoglu	**Length (m):** 151.50
Drilled by: Driftwood Drilling	**Horizontal Projection:**
Assayed by: ALS Chemex	**Vertical Projection:**
Core Size: BTW	
Date Started: 2007/06/09 **Date Completed:** 2007/06/09	
Dip Tests By: -	
Objective This hole is an augmented combination of a 2 hole fence across the broadest extent of a large Au and Cu geochemical anomaly across the valley. This hole is also designed to intercept the NW trending fault which divides	

Summary Log:

146.40 - 151.50 Diorite
32.00 - 146.40 Andesite Flow
0.0 - 32.00 Casing
151.50 - EOH
Hole terminated due to required bit change and inability to re-enter hole. Casing only penetrated to half the depth of the overburden due to ground conditions and potentially the strength of the drill.

EQUITY ENGINEERING LTD.

DRILL LOG

Project: JW

Hole ID: JW07-07

Downhole surveys:

Depth	Dip	Azimuth
0.00	-50.00	90.00

Project: JW

Hole Number: JW07-07

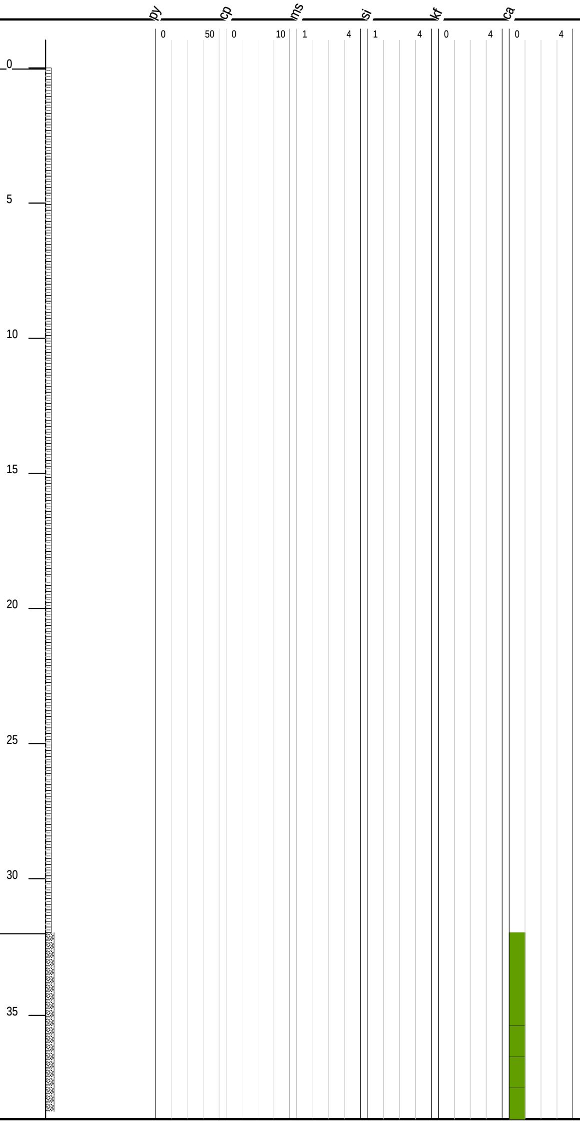
From	To	Rocktype	& Description
0.00	32.00	CASN	

Dark black mud, with minor rounded fragments

From	To	Rocktype	& Description
32.00	146.40	ANDS	

Dark green fine-grained grain. Horneblende biotite chlorite, feldspar. Massive with weakly to moderately banded and patchy epidote and carbonate +/- pyrite. Carbonate present as micro fracture fill with a moderate fracture density of 17 / m.

« 42.90- 44.85 Epidote 2.00-3.00*» Roughly flat lying banded epidote carbonate alteration, dominantly epidote with minor calcite seams of mm scale.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
37.80	42.10	4.30	200875	0.01	0.30	313
42.10	45.10	3.00	200876	0.02	0.30	404
45.10	48.16	3.06	200877	0.02	0.30	353
48.16	51.20	3.04	200878	0.04	-0.20	156
51.20	53.15	1.95	200879	0.03	-0.20	161
51.20	53.15	1.95	200880	0.03	0.20	150
53.15	55.15	2.00	200881	0.03	0.30	319

*graphic log not to scale

Project: JW **Hole Number:** JW07-07

From **To** **Rocktype & Description**

Trace fine-grained grain and blebby pyrite.

< @ 49.80 Vein 45.00º 1.00cm > Vuggy calcite quartz vein with minor trace very fine-grained grain pyrite. Minor chlorite alteration.

< @ 52.00 Vein 25.00º 0.50cm > Carbonate vein with wispy contacts.

« 51.00- 57.00 Pyrite 5.00%» Very fine-grained disseminated grains with higher concentration visible in the patchy epidote alteration.

« 32.00- 68.70 Epidote 2.00*»

« Chlorite 2.00*»

« Calcite 1.00*»

« 62.00- 68.70 Pyrite 5.00%» As both very fine-grained disseminated grains and wispy blebs.

« 60.40- 61.70 small Dyke GRNT » Fine grain quartz feldspar and chlorite with very fine-grained grain disseminated pyrite.

« Pyrite 5.00%»

« 68.70- 103.60 Epidote 2.00*» As mottled patches of alteration with wispy elongation @ 40 to 90 TCA.

« Calcite 1.00*» As fracture fill and contact alteration around some of the patches of epidote alteration and minor pervasive alteration.

« Pyrite 5.00%» Very fine-grained grain disseminated pyrite.

« 87.70- 89.50 Minor subinterval, gradational contacts VCLC »

Volcaniclastic subinterval? More defined alteration potentially clasts, groundmass is more carbonate altered then above intervals, alteration halos around some clasts, more biotite rich interval.

« 87.70- 103.60 Biotite 2.00*»

« 99.50- 99.75 small dyke GRNT 50.00-70.00°» Fine grain feldspar quartz chlorite (after horneblende?) with flat lying epidote veining.

« 103.60- 121.00 Epidote 2.00*» Mottled texture and patchy alteration.

« Pyrite 3.00%» Fine disseminated grains and wispy weak carbonate micro fracture fill and as fine-grained veinlets variable angle to the core. as well as present as fine-grained pervasive alteration around epidote altered patches.

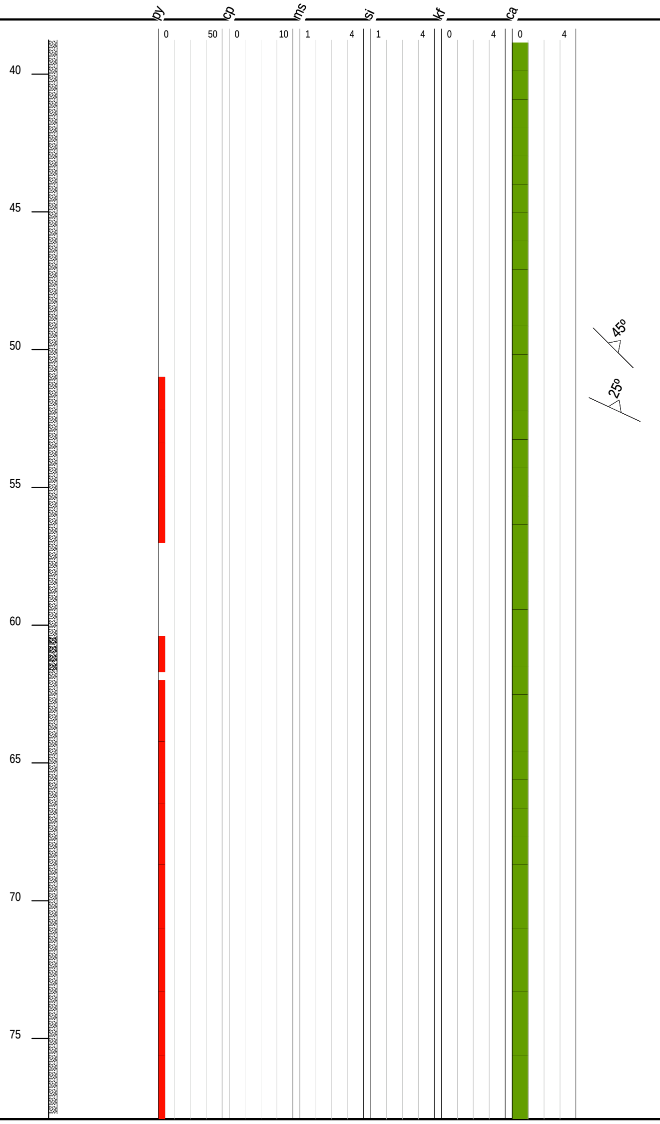
« Calcite 1.00*»

« 109.00- 113.55 Magnetite 5.00-10.00%» Pervasive alteration, core is moderately magnetitc.

« 103.00- 121.00 Chlorite 1.00*»

« Biotite 1.00-2.00*» Pervasive fine-grained grain biotite also as halo around epidote clasts. slight purple hue in larger clots.

< @ 115.10 Vein 40.00º 0.50cm > Carbonate vein rimmed by pyrite with chlorite alteration halo.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
55.15	57.30	2.15	200882	0.07	-0.20	222
57.30	59.35	2.05	200883	0.02	0.30	239
59.35	60.40	1.05	200884	0.03	0.30	192
60.40	61.70	1.30	200885	0.02	0.20	229
61.70	63.40	1.70	200886	0.03	0.60	627
63.40	65.45	2.05	200887	0.02	0.30	392
65.45	67.40	1.95	200888	0.02	0.60	557
67.40	69.50	2.10	200889	0.02	0.30	400
69.50	69.50	0.00	200890	0.01	0.60	415
69.50	71.50	2.00	200891	0.02	0.30	284
71.50	73.50	2.00	200892	0.01	0.20	306
73.50	75.60	2.10	200893	0.02	0.50	616
75.60	77.60	2.00	200894	0.01	0.30	380
77.60	79.60	2.00	200895	0.01	0.30	297
79.60	81.70	2.10	200896	-0.01	-0.20	151
81.70	83.70	2.00	200897	0.01	-0.20	114
83.70	85.75	2.05	200898	0.02	0.20	232
85.75	87.80	2.05	200899	0.01	0.20	142
85.75	87.80	2.05	200900	0.01	-0.20	149
87.80	89.80	2.00	200901	0.05	0.30	483
89.80	91.90	2.10	200902	0.04	0.40	549
91.90	93.90	2.00	200903	0.05	0.40	383
93.90	95.90	2.00	200904	0.04	0.40	518
95.90	97.90	2.00	200905	0.04	0.40	414
97.90	99.50	1.60	200906	0.03	0.20	269
99.50	100.00	0.50	200907	0.01	-0.20	6
100.00	102.00	2.00	200908	0.01	0.20	173
102.00	104.00	2.00	200909	0.01	-0.20	177
102.00	104.00	2.00	200910	0.02	-0.20	179
104.00	106.10	2.10	200911	0.02	0.20	343
106.10	108.10	2.00	200912	0.02	0.20	369
108.10	110.10	2.00	200913	0.02	0.50	603
110.10	112.20	2.10	200914	0.03	0.40	629
112.20	114.20	2.00	200915	0.06	0.30	645
114.20	116.20	2.00	200916	0.05	0.30	621
116.20	118.30	2.10	200917	0.04	-0.20	88
118.30	119.90	1.60	200918	0.03	-0.20	535
119.90	121.05	1.15	200919	0.11	0.20	202
121.05	121.05	0.00	200920	-0.01	-0.20	1

*graphic log not to scale

Project: JW **Hole Number:** JW07-07

From **To** **Rocktype** **& Description**

< @ 117.30 Vein 50.00º 0.20cm > Carbonate pyrite chlorite biotite vein set (pair of veins).

< @ 120.05 Vein 45.00º 1.00cm > Pyrite carbonate vein.

< @ 120.25 Vein 40.00º 1.00cm > Pyrite carbonate vein.

< @ 121.40 Vein 90.00º 8.00cm > Pyrite vein with a Quartz carbonate core 0.5cm wide.

< @ 121.52 Vein 55.00º 1.00cm > Pyrite carbonate calcite vein.

< @ 122.30 Vein 5.00cm > Pyrite calcite quartz vein.

« 122.60- 123.30 Magnetite 3.00%» Banded magnetite alteration.

« 122.60- 146.30 Epidote 2.00*» Patches and wispy veinlets of variable angles between 10 and 60 of epidote alteration.

« Chlorite 1.00*»

« 130.50- 134.30 Calcite 2.00*» Vein set of carbonate, half centimetre in size dominantly oriented @ 60 to 90 TCA.

< @ 133.10 Fault 24.00º 0.01cm > Annealed fault structure with sharp slickenside, rusty fracture plain with glassy texture and hardness >6. Most apparent of a set over the metre interval oriented between 24 and 60 TCA. some with calcite vein along same angle.

« 133.50- 137.40 Specularite 3.00%» Purple lusterous coating on fracture planes, reddish streak.

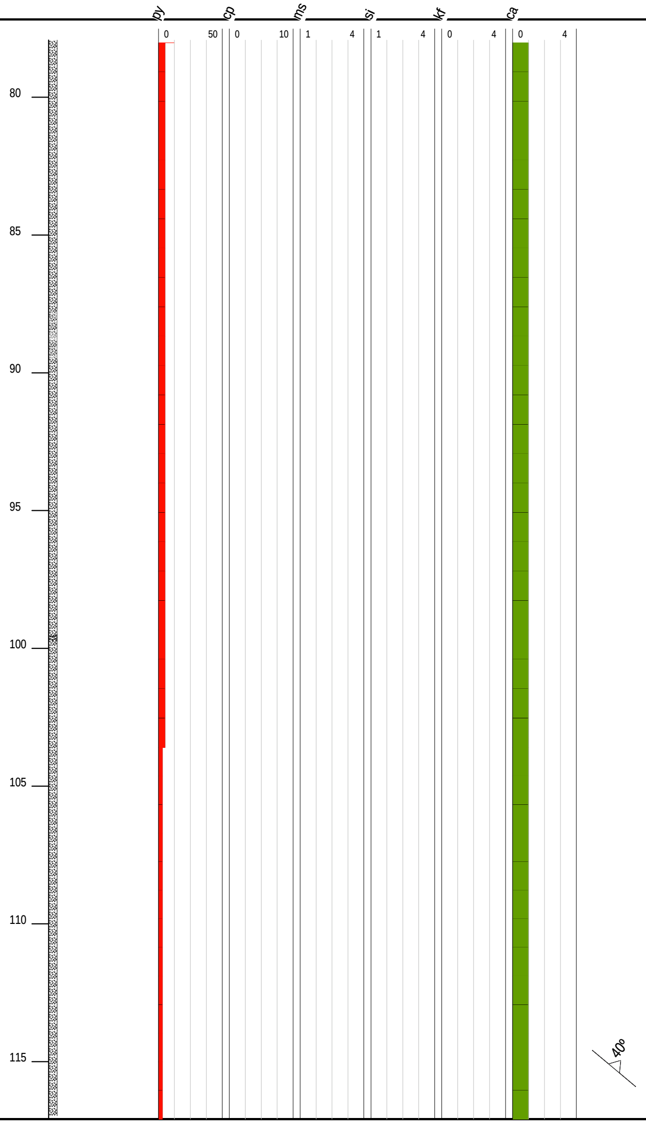
« Pyrite 2.00%» As very fine-grained grain pyrite.

« 133.50- 146.40 Biotite 1.00*» Patchy biotite and chlorite alteration.

< @ 137.25 Vein 45.00º 0.10cm > Vein cset of carbonate and hematite 10cm of core.

« 139.00- 142.60 Pyrite 2.00%»

« 142.60- 146.40 Pyrite 3.00%» Fracture coating and fine-grained grain disseminated and larger blebs of euhedral to subhedral fine-grained pyrite.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
121.05	122.35	1.30	200921	0.12	0.50	796
122.35	124.40	2.05	200922	0.03	0.30	695
124.40	126.30	1.90	200923	0.01	-0.20	95
126.30	128.30	2.00	200924	0.01	-0.20	112
128.30	130.50	2.20	200925	0.02	-0.20	89
130.50	132.50	2.00	200926	0.01	-0.20	90
132.50	134.50	2.00	200927	-0.01	-0.20	91
134.50	136.60	2.10	200928	0.02	-0.20	88
136.60	138.60	2.00	200929	0.02	0.20	105
136.60	138.60	2.00	200930	0.01	-0.20	138
138.60	141.70	3.10	200931	0.01	-0.20	207
141.70	142.60	0.90	200932	0.01	-0.20	145
142.60	144.60	2.00	200933	0.02	-0.20	259
144.60	146.40	1.80	200934	0.02	-0.20	293

*graphic log not to scale

Project: JW **Hole Number:** JW07-07

From	To	Rocktype & Description
146.40	151.50	DIOR

Grey to light green, fine-grained grain relatively massive and equigranular. Weak to moderate epidote alteration weak chlorite alteration, fine-grained grain disseminated pyrite.

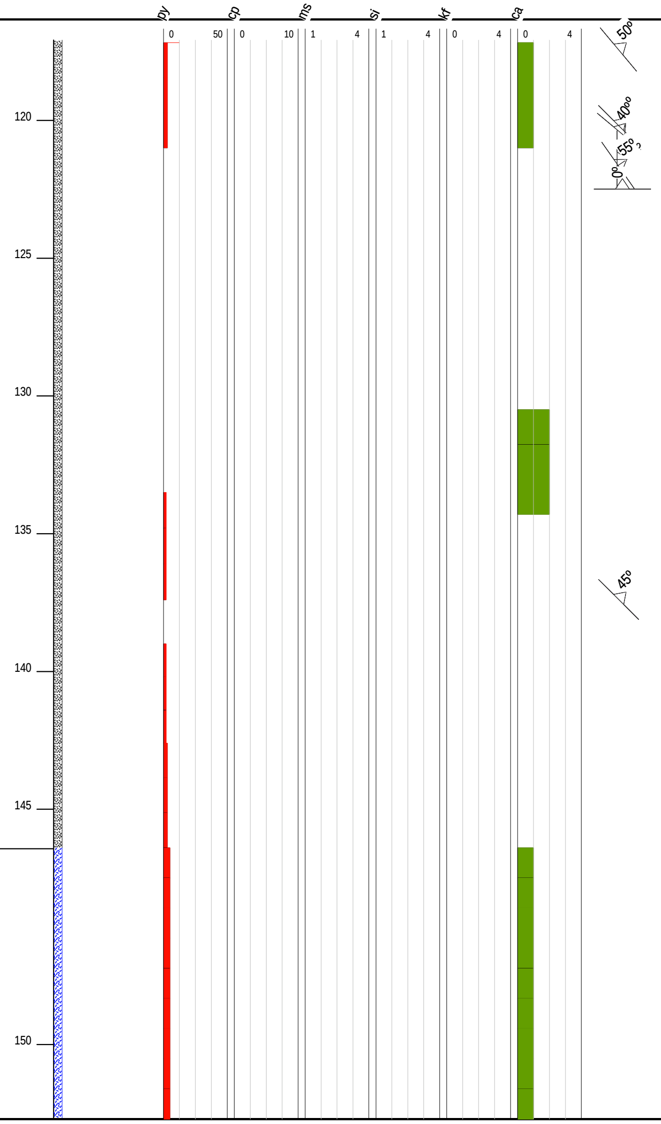
« Pyrite 5.00%»

« Chlorite 1.00*»

« Calcite 1.00*»

« Epidote 1.00-2.00*»

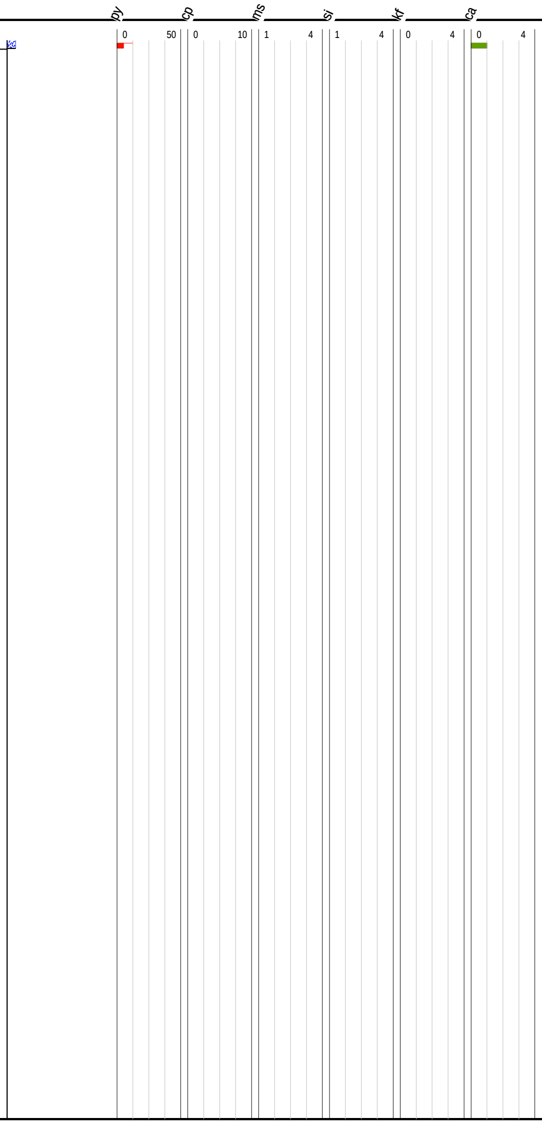
Gradational and fragmented contact in heavily rubbled core.



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm
146.40	148.10	1.70	200935	0.04	0.30	544
148.10	151.50	3.40	200936	0.06	0.50	721

Project: JW **Hole Number:** JW07-07

From	To	Rocktype & Description
151.50	151.50	EOH



From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm

*graphic log not to scale

Drill Log Legend

▨	ANDS
CASING	CASN
▨	DIOR
■	EOH
▨	GRNT
▨	VCLC
╲	Vein

DRILL LOG

Project: JW	**Collar Elevation (m):** 384.0
Hole JW07-08	**Azimuth (º):** 110
Location: 6340641 m North 343164 m East	**Dip (º):** -70.0
Logged by: R.Kutluoglu	**Length (m):** 32.50
Drilled by: Driftwood Drilling	**Horizontal Projection:**
Assayed by: ALS Chemex	**Vertical Projection:**
Core Size: BTW	
Date Started: 2007/06/11 **Date Completed:** 2007/06/11	
Dip Tests By: -	
Objective This hole will test under the main exposed mineralization and trenching, this hole runs parallel and immediately north of L2+00N. The objective is to test the southern lobe of the Au/Cu anomaly. It is sited to the west of the 90JW-4	

Summary Log:

0.0 - 32.80 Casing

32.80 - EOH
Failed to find rock, drillers drilled ahead 10m without touching anything solid. Hole terminated, surface geochemistry not likely to be derived from bedrock source, hole succeeded in testing that.

Downhole surveys:

Depth	Dip	Azimuth
0.00	-70.00	110.00

Project: JW

Hole Number: JW07-08

From	To	Rocktype & Description
0.00	32.50	CASN Failed to reach bedrock, drillers drove casing to 32.50m and drilled ahead several metres and still failed to find bedrock. Overburden material is sand and gravels transitioning to a rich black mud at depth.
32.50	32.50	EOH

py	cp	ms	si	kf	ca
0 50	0 10	1 4	1 4	0 4	0 4

From	To	Width	Sample	Au ppm	Ag ppm	Cu ppm

2007/09/25 **Equity Engineering Ltd.**

*graphic log not to scale

Drill Log Legend

CASING	**CASN**	■	**EOH**	╲	**Vein**

Appendix D: Certificates of Analysis

ALS Chemex
EXCELLENCE IN ANALYTICAL CHEMISTRY
ALS Canada Ltd.
212 Brooksbank Avenue
North Vancouver BC V7J 2C1
Phone: 604 984 0221 Fax: 604 984 0218 www.alschemex.com

To: **EQUITY ENGINEERING LTD.**
700 - 700 W. PENDER STREET
VANCOUVER BC V6C 1G8

Finalized Date: 29-JUN-2007
Account: EIARG

CERTIFICATE VA07061776

Project: JW

P.O. No.: RG07-03

This report is for 123 Drill Core samples submitted to our lab in Vancouver, BC, Canada on 15-JUN-2007.

The following have access to data associated with this certificate:

QUITY ENGINEERING GENERA	RORY KUTLUOGLU	MICHAEL SMITH

SAMPLE PREPARATION

ALS CODE	DESCRIPTION
WEI-21	Received Sample Weight
LOG-22	Sample login - Rcd w/o BarCode
CRU-QC	Crushing QC Test
PUL-QC	Pulverizing QC Test
CRU-31	Fine crushing - 70% <2mm
SPL-21	Split sample - riffle splitter
PUL-31	Pulverize split to 85% <75 um
LOG-24	Pulp Login - Rcd w/o Barcode

ANALYTICAL PROCEDURES

ALS CODE	DESCRIPTION	INSTRUMENT
Au-AA23	Au 30g FA-AA finish	AAS
Au-GRA21	Au 30g FA-GRAV finish	WST-SIM
ME-ICP41	35 Element Aqua Regia ICP-AES	ICP-AES

To: **EQUITY ENGINEERING LTD.**
ATTN: EQUITY ENGINEERING GENERAL
700 - 700 W. PENDER STREET
VANCOUVER BC V6C 1G8

This is the Final Report and supersedes any preliminary report with this certificate number. Results apply to samples as submitted. All pages of this report have been checked and approved for release.

Signature:

Lawrence Ng, Laboratory Manager - Vancouver

Project: JW

CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method	WEI-21	Au-AA23	Au-GRA21	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Recvd Wt.	Au	Au	Ag	Al	As	B	Ba	Be	Bi	Ca	Cd	Co	Cr	Cu
	Units	kg	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
	LOR	0.02	0.005	0.05	0.2	0.01	2	10	10	0.5	2	0.01	0.5	1	1	1
200733		1.50	0.032		<0.2	2.94	<2	<10	140	<0.5	<2	1.80	<0.5	24	11	249
200734		2.80	0.206		2.2	1.14	4	<10	90	<0.5	<2	2.10	0.6	29	12	4130
200735		3.86	0.044		<0.2	2.42	3	<10	100	<0.5	2	1.86	<0.5	25	13	219
200736		4.84	0.043		<0.2	2.73	3	<10	120	<0.5	<2	1.75	<0.5	25	19	229
200737		3.48	0.084		0.6	1.16	<2	<10	70	<0.5	<2	2.04	<0.5	19	12	1415
200738		3.42	0.387		0.8	1.04	8	<10	70	<0.5	<2	3.28	<0.5	15	12	1820
200739		1.68	1.035	1.02	1.2	0.88	4	<10	100	<0.5	<2	1.63	<0.5	19	12	2730
200740		1.68	0.530		1.1	0.99	6	<10	120	<0.5	2	1.63	<0.5	20	14	2610
200741		3.76	0.541		1.0	0.93	8	<10	60	<0.5	<2	1.77	<0.5	15	10	2660
200742		2.80	0.209		0.7	1.03	4	<10	70	0.6	<2	2.23	<0.5	13	9	2230
200743		2.36	0.104		0.5	1.02	<2	<10	70	<0.5	<2	2.33	<0.5	12	15	1520
200744		3.72	0.019		<0.2	2.56	6	<10	200	0.5	<2	2.11	<0.5	26	21	244
200745		3.00	<0.005		<0.2	2.86	8	<10	270	0.5	<2	2.20	<0.5	24	14	52
200746		3.32	<0.005		<0.2	2.92	2	<10	270	<0.5	<2	1.74	<0.5	24	8	48
200747		3.62	<0.005		<0.2	2.68	3	<10	770	<0.5	<2	2.18	<0.5	21	7	54
200748		3.62	<0.005		<0.2	2.88	3	<10	230	0.5	<2	2.15	<0.5	23	7	49
200749		2.90	0.005		<0.2	2.72	6	<10	250	<0.5	<2	2.42	<0.5	22	8	50
200750		2.58	0.005		<0.2	2.68	3	<10	240	<0.5	<2	2.41	<0.5	22	8	98
200751		1.28	<0.005		<0.2	2.67	3	<10	130	<0.5	<2	2.60	<0.5	22	7	44
200752		3.38	0.020		<0.2	2.63	<2	<10	120	<0.5	<2	2.08	<0.5	20	7	137
200753		0.98	0.159		1.1	1.20	9	<10	30	<0.5	2	4.22	<0.5	42	2	570
200754		3.64	0.016		0.4	2.28	<2	<10	200	<0.5	<2	2.71	1.0	19	8	427
200755		3.84	0.236		1.8	1.39	2	<10	90	<0.5	<2	2.61	0.5	13	12	2870
200756		3.90	0.075		0.7	1.43	10	<10	90	<0.5	2	2.13	<0.5	12	9	1255
200757		3.52	0.131		0.6	1.33	7	<10	90	0.5	<2	1.73	<0.5	32	10	1625
200758		3.26	0.095		0.2	1.54	6	<10	90	0.5	<2	2.73	<0.5	9	13	714
200759		1.74	0.105		0.3	1.72	8	<10	90	0.6	<2	2.15	<0.5	13	9	896
200760		1.72	0.076		0.2	1.90	7	<10	100	0.6	<2	2.24	<0.5	11	9	731
200761		3.14	0.042		0.3	1.41	6	<10	80	0.5	<2	2.00	<0.5	4	9	818
200762		3.68	0.067		0.4	1.34	6	<10	110	0.5	<2	1.59	<0.5	8	12	995
200763		3.94	0.017		0.2	1.17	3	<10	70	<0.5	<2	1.49	<0.5	4	8	694
200764		3.34	0.033		0.2	1.28	3	<10	80	0.5	<2	2.62	<0.5	11	11	591
200765		4.16	0.032		0.3	0.65	8	<10	100	<0.5	<2	1.86	<0.5	34	16	584
200766		3.48	0.044		0.2	0.86	6	<10	110	<0.5	<2	2.41	<0.5	6	18	808
200767		3.36	0.040		0.8	0.52	2	<10	90	<0.5	<2	2.28	<0.5	6	17	1350
200768		3.28	0.039		0.6	0.65	4	<10	90	<0.5	<2	1.76	<0.5	8	20	1155
200769		4.10	0.025		0.2	0.52	3	<10	100	<0.5	<2	1.97	<0.5	2	15	463
200770		0.64	0.032		<0.2	1.03	4	<10	40	<0.5	<2	1.88	<0.5	3	4	596
200771		0.54	0.024		<0.2	0.90	4	<10	30	<0.5	<2	1.69	<0.5	3	4	485
200772		5.22	0.033		0.3	0.78	3	<10	100	<0.5	<2	1.11	<0.5	4	6	579

ALS Chemex
EXCELLENCE IN ANALYTICAL CHEMISTRY
ALS Canada Ltd.

212 Brooksbank Avenue
North Vancouver BC V7J 2C1
Phone: 604 984 0221 Fax: 604 984 0218 www.alschemex.com

To: **EQUITY ENGINEERING LTD.**
700 - 700 W. PENDER STREET
VANCOUVER BC V6C 1G8

Finalized Date: 29-JUN-2007
Account: EIARG

Project: JW

CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method Analyte Units LOR	ME-ICP41 Fe % 0.01	ME-ICP41 Ga ppm 10	ME-ICP41 Hg ppm 1	ME-ICP41 K % 0.01	ME-ICP41 La ppm 10	ME-ICP41 Mg % 0.01	ME-ICP41 Mn ppm 5	ME-ICP41 Mo ppm 1	ME-ICP41 Na % 0.01	ME-ICP41 Ni ppm 1	ME-ICP41 P ppm 10	ME-ICP41 Pb ppm 2	ME-ICP41 S % 0.01	ME-ICP41 Sb ppm 2	ME-ICP41 Sc ppm 1
200733		5.27	10	1	1.13	<10	2.30	949	2	0.08	15	1440	<2	0.65	2	7
200734		4.45	10	<1	0.57	10	0.96	199	2	0.13	21	2530	<2	3.88	<2	7
200735		4.96	10	1	0.80	10	1.92	793	2	0.09	22	1810	<2	1.37	<2	7
200736		4.76	10	<1	0.80	<10	2.26	896	3	0.08	18	1430	16	0.30	3	7
200737		2.62	10	<1	0.54	10	0.99	288	2	0.08	17	2450	5	1.42	<2	5
200738		4.32	10	1	0.54	10	0.97	352	2	0.08	13	2360	5	3.37	3	4
200739		4.11	<10	<1	0.52	10	0.70	171	3	0.05	16	1550	2	3.02	3	3
200740		3.89	<10	<1	0.60	10	0.74	178	2	0.07	16	1620	2	2.91	<2	4
200741		4.16	<10	1	0.41	10	0.77	155	2	0.05	19	2000	2	1.62	<2	3
200742		2.53	<10	<1	0.37	10	1.00	168	2	0.07	12	2030	2	1.69	<2	4
200743		2.18	10	<1	0.47	10	0.86	205	2	0.08	12	1960	3	1.31	2	3
200744		5.35	10	1	1.39	<10	2.33	883	1	0.06	15	1390	3	0.96	2	6
200745		5.81	10	<1	0.97	<10	2.47	934	1	0.05	10	1620	5	0.08	3	6
200746		4.96	10	1	0.69	<10	2.36	903	<1	0.05	7	1660	6	0.08	3	7
200747		4.44	10	<1	0.27	<10	2.13	850	<1	0.04	6	1610	6	0.14	2	6
200748		5.20	10	2	0.69	<10	2.33	913	<1	0.05	9	1730	5	0.08	2	7
200749		5.26	<10	<1	0.61	<10	2.33	922	<1	0.05	7	1660	5	0.11	2	6
200750		5.22	10	<1	0.65	<10	2.34	925	<1	0.05	7	1650	4	0.10	2	6
200751		5.61	10	1	0.24	10	2.30	1115	1	0.04	7	1630	3	0.07	3	8
200752		5.31	10	1	0.28	10	2.19	942	<1	0.05	7	1660	6	0.14	2	7
200753		5.51	<10	<1	0.36	10	0.83	470	7	0.05	11	1850	11	4.87	3	7
200754		5.00	10	<1	0.44	<10	1.97	827	1	0.04	7	1710	55	0.83	2	8
200755		4.39	<10	1	0.61	10	1.18	299	3	0.06	13	2170	13	2.36	<2	4
200756		5.57	<10	<1	0.78	<10	1.40	266	2	0.09	14	1750	10	3.93	4	4
200757		4.38	<10	1	0.79	<10	1.30	223	3	0.07	14	2010	6	3.15	2	4
200758		3.92	10	1	0.49	10	1.47	320	2	0.07	14	1610	6	1.66	3	7
200759		4.37	10	<1	0.69	<10	1.57	261	2	0.06	15	1810	7	0.95	3	6
200760		4.83	10	<1	0.70	<10	1.67	283	3	0.08	14	1950	2	1.02	2	7
200761		2.88	10	<1	0.57	10	1.30	222	2	0.07	10	1880	4	0.92	3	4
200762		3.08	<10	1	0.78	10	1.18	210	2	0.07	11	2040	5	0.77	2	4
200763		1.80	10	<1	0.66	<10	1.24	214	2	0.07	8	1400	8	0.55	<2	3
200764		3.52	10	<1	0.54	10	1.25	291	2	0.07	12	1970	6	1.69	3	5
200765		3.44	<10	<1	0.24	<10	0.58	195	3	0.03	12	1560	3	2.87	2	2
200766		2.22	<10	<1	0.23	10	0.61	248	2	0.07	11	1320	2	0.64	2	2
200767		1.46	<10	<1	0.21	<10	0.55	231	2	0.03	7	870	2	0.48	2	2
200768		3.34	<10	<1	0.25	10	0.39	159	2	0.07	11	1730	3	1.03	<2	1
200769		1.32	<10	1	0.21	<10	0.41	186	2	0.03	4	1060	<2	0.28	<2	2
200770		1.20	<10	<1	0.10	10	0.66	197	1	0.10	6	2560	5	0.09	2	2
200771		1.09	<10	<1	0.07	10	0.58	174	2	0.08	6	2570	<2	0.10	2	2
200772		1.62	<10	<1	0.21	<10	0.39	141	2	0.07	4	1490	2	0.50	<2	1

To: **EQUITY ENGINEERING LTD.**
700 - 700 W. PENDER STREET
VANCOUVER BC V6C 1G8

Project: JW

CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method Analyte Units LOR	ME-ICP41 Sr ppm 1	ME-ICP41 Th ppm 20	ME-ICP41 Ti % 0.01	ME-ICP41 Tl ppm 10	ME-ICP41 U ppm 10	ME-ICP41 V ppm 1	ME-ICP41 W ppm 10	ME-ICP41 Zn ppm 2
200733		143	<20	0.37	<10	<10	132	<10	91
200734		155	<20	0.22	<10	<10	127	<10	20
200735		155	<20	0.36	<10	<10	135	<10	73
200736		180	<20	0.34	<10	<10	126	<10	85
200737		185	<20	0.20	<10	<10	120	<10	26
200738		182	<20	0.19	<10	<10	126	<10	24
200739		148	<20	0.15	<10	<10	101	<10	13
200740		164	<20	0.17	<10	<10	104	<10	15
200741		185	<20	0.15	<10	<10	120	<10	13
200742		159	<20	0.17	<10	<10	105	<10	11
200743		178	<20	0.21	<10	<10	101	<10	15
200744		166	<20	0.32	<10	<10	137	<10	71
200745		132	<20	0.29	<10	<10	152	<10	82
200746		164	<20	0.30	<10	<10	140	<10	95
200747		279	<20	0.21	<10	<10	112	<10	76
200748		194	<20	0.26	<10	<10	135	<10	89
200749		173	<20	0.26	<10	<10	140	<10	92
200750		155	<20	0.26	<10	<10	140	<10	96
200751		116	<20	0.20	<10	<10	148	<10	101
200752		135	<20	0.20	<10	<10	127	<10	87
200753		165	<20	<0.01	<10	<10	64	<10	18
200754		146	<20	0.20	<10	<10	141	<10	102
200755		237	<20	0.18	<10	<10	134	<10	32
200756		195	<20	0.18	<10	<10	158	<10	27
200757		156	<20	0.16	<10	<10	123	<10	20
200758		192	<20	0.15	<10	<10	150	10	28
200759		232	<20	0.17	<10	<10	179	<10	23
200760		262	<20	0.20	<10	<10	200	<10	25
200761		238	<20	0.19	<10	<10	134	<10	22
200762		196	<20	0.20	<10	<10	148	<10	20
200763		115	<20	0.16	<10	<10	84	<10	40
200764		188	<20	0.17	<10	<10	144	<10	23
200765		96	<20	0.12	<10	<10	79	<10	12
200766		219	<20	0.16	<10	<10	103	<10	11
200767		96	<20	0.12	<10	<10	79	10	9
200768		170	<20	0.16	<10	<10	126	<10	10
200769		116	<20	0.11	<10	<10	64	<10	8
200770		313	<20	0.16	<10	<10	70	10	15
200771		275	<20	0.13	<10	<10	62	10	14
200772		244	<20	0.12	<10	<10	81	<10	13

CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method	WEI-21	Au-AA23	Au-GRA21	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Recvd Wt.	Au	Au	Ag	Al	As	B	Ba	Be	Bi	Ca	Cd	Co	Cr	Cu
	Units	kg	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
	LOR	0.02	0.005	0.05	0.2	0.01	2	10	10	0.5	2	0.01	0.5	1	1	1
200773		5.20	0.079		1.1	0.59	2	<10	70	<0.5	<2	1.36	<0.5	2	9	1840
200774		1.92	0.049		0.5	0.49	4	<10	140	<0.5	<2	1.52	<0.5	16	9	612
200775		3.54	0.048		0.4	0.77	3	<10	110	<0.5	<2	2.18	<0.5	40	4	441
200776		4.76	0.063		1.2	0.64	4	<10	100	<0.5	<2	1.34	<0.5	7	2	2100
200777		2.90	0.051		0.4	0.56	3	<10	60	<0.5	<2	2.13	<0.5	9	3	728
200778		2.38	0.068		0.2	0.62	5	<10	70	<0.5	<2	1.70	<0.5	18	3	248
200779		3.00	0.048		0.3	1.00	4	<10	70	<0.5	<2	2.63	<0.5	15	6	460
200780		0.10	<0.005		<0.2	0.01	2	<10	10	<0.5	<2	0.01	<0.5	<1	<1	3
200781		2.06	0.026		<0.2	0.67	3	<10	100	<0.5	<2	1.64	<0.5	8	3	120
200782		5.22	0.041		<0.2	0.68	3	<10	60	<0.5	<2	2.13	<0.5	13	4	209
200783		5.22	0.034		0.2	0.72	4	<10	190	<0.5	2	2.17	<0.5	10	6	335
200784		1.88	0.027		<0.2	0.56	10	<10	70	<0.5	<2	1.72	<0.5	24	6	150
200785		3.74	0.028		0.2	0.90	2	<10	110	<0.5	<2	2.16	<0.5	15	6	429
200786		1.40	0.017		0.5	0.94	3	<10	40	<0.5	<2	2.41	0.5	4	3	2290
200787		2.90	0.014		<0.2	1.14	4	<10	80	<0.5	<2	2.96	<0.5	6	8	231
200788		3.74	0.030		0.2	0.54	<2	<10	50	<0.5	<2	1.87	<0.5	10	9	321
200789		1.66	0.014		<0.2	0.46	5	<10	60	<0.5	<2	1.79	<0.5	7	5	252
200790		1.68	0.013		<0.2	0.45	3	<10	60	<0.5	<2	1.78	<0.5	7	5	248
200791		4.08	0.006		<0.2	0.51	<2	<10	110	<0.5	<2	1.86	<0.5	4	9	107
200792		3.64	0.030		<0.2	0.37	3	<10	80	<0.5	<2	2.26	<0.5	21	6	105
200793		1.88	0.018		<0.2	0.37	7	<10	120	<0.5	<2	2.24	<0.5	15	8	54
200794		3.36	0.030		0.2	0.66	4	<10	240	0.5	<2	3.63	<0.5	15	18	311
200795		3.48	0.026		0.4	0.66	2	<10	300	<0.5	<2	2.88	0.6	10	23	316
200796		3.56	0.007		<0.2	0.47	<2	<10	120	<0.5	<2	2.18	<0.5	4	8	176
200797		2.64	0.034		0.3	0.68	5	<10	220	<0.5	<2	4.42	<0.5	7	25	474
200798		2.86	0.061		1.0	0.48	2	<10	60	<0.5	5	3.19	<0.5	10	4	476
200799		1.50	0.011		0.2	0.45	7	<10	120	<0.5	<2	2.80	<0.5	8	6	206
200800		1.40	0.013		<0.2	0.45	4	<10	130	<0.5	<2	2.87	<0.5	7	7	201
200801		2.96	0.006		<0.2	0.73	3	<10	230	<0.5	<2	1.88	<0.5	4	4	235
200802		2.56	<0.005		<0.2	0.51	5	<10	110	<0.5	<2	1.23	<0.5	2	4	37
200803		2.94	0.020		<0.2	0.46	5	<10	120	<0.5	<2	2.91	<0.5	8	7	65
200804		3.98	0.116		0.2	0.29	<2	<10	90	<0.5	<2	2.32	<0.5	24	4	334
200805		3.78	0.041		<0.2	0.40	7	<10	120	<0.5	<2	3.07	<0.5	31	6	48
200806		0.94	0.233		0.2	1.21	4	<10	350	0.6	<2	4.05	<0.5	16	12	210
200807		3.40	0.624		0.3	0.95	2	<10	110	<0.5	<2	1.86	<0.5	5	5	434
200808		2.96	0.008		0.2	0.86	6	<10	120	<0.5	<2	1.54	<0.5	6	7	172
200809		1.76	0.022		0.3	0.53	4	<10	80	<0.5	2	1.91	<0.5	5	5	265
200810		1.54	0.041		0.3	0.60	3	<10	80	<0.5	2	1.80	<0.5	4	5	335
200811		3.82	0.042		0.3	0.68	4	<10	90	<0.5	<2	1.83	<0.5	4	4	304
200812		3.68	0.007		0.2	0.62	<2	<10	100	<0.5	2	1.47	<0.5	5	4	114

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CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method Analyte Units LOR	ME-ICP41 Fe % 0.01	ME-ICP41 Ga ppm 10	ME-ICP41 Hg ppm 1	ME-ICP41 K % 0.01	ME-ICP41 La ppm 10	ME-ICP41 Mg % 0.01	ME-ICP41 Mn ppm 5	ME-ICP41 Mo ppm 1	ME-ICP41 Na % 0.01	ME-ICP41 Ni ppm 1	ME-ICP41 P ppm 10	ME-ICP41 Pb ppm 2	ME-ICP41 S % 0.01	ME-ICP41 Sb ppm 2	ME-ICP41 Sc ppm 1
200773		1.56	<10	<1	0.16	<10	0.29	136	1	0.04	2	1690	2	0.38	<2	1
200774		3.71	<10	<1	0.25	<10	0.21	169	2	0.04	5	1820	4	1.35	<2	1
200775		4.51	<10	<1	0.28	<10	0.54	286	2	0.06	8	1560	4	2.36	<2	1
200776		2.55	<10	<1	0.18	10	0.35	170	5	0.06	5	1710	2	0.75	<2	1
200777		3.62	<10	<1	0.12	<10	0.30	218	2	0.04	5	1900	2	0.86	2	1
200778		2.95	<10	<1	0.14	<10	0.42	217	2	0.08	6	1580	4	1.70	<2	2
200779		3.42	10	<1	0.14	<10	0.83	415	2	0.07	6	1660	2	1.27	<2	3
200780		0.02	<10	<1	<0.01	<10	<0.01	<5	<1	<0.01	<1	20	<2	0.01	<2	<1
200781		2.07	<10	<1	0.16	<10	0.43	223	1	0.07	4	1290	<2	0.66	<2	1
200782		2.53	<10	<1	0.13	<10	0.42	261	1	0.06	5	1800	3	1.14	<2	2
200783		3.30	<10	<1	0.16	10	0.47	254	1	0.08	5	2040	3	1.31	2	3
200784		3.46	<10	<1	0.11	<10	0.38	179	3	0.06	7	2200	5	2.29	<2	2
200785		3.10	<10	<1	0.17	10	0.67	297	1	0.10	6	2110	3	1.45	<2	3
200786		2.11	<10	<1	0.06	10	0.79	360	<1	0.07	4	2420	3	0.48	<2	5
200787		2.29	10	1	0.12	10	0.91	401	2	0.08	4	2130	2	0.43	<2	6
200788		2.01	<10	<1	0.08	<10	0.33	188	2	0.07	6	1860	5	1.54	<2	2
200789		1.09	<10	<1	0.10	<10	0.22	163	1	0.06	2	1620	4	0.68	<2	2
200790		1.08	<10	<1	0.09	10	0.20	153	1	0.05	3	1770	4	0.66	<2	2
200791		1.58	<10	<1	0.20	<10	0.26	213	1	0.05	1	1440	2	0.39	<2	1
200792		2.27	<10	<1	0.12	<10	0.22	220	2	0.04	2	1350	8	1.32	<2	1
200793		2.81	<10	<1	0.19	<10	0.29	239	5	0.03	6	1560	24	1.60	<2	2
200794		3.07	<10	<1	0.19	10	0.77	507	1	0.03	7	1240	5	0.88	<2	6
200795		1.87	<10	<1	0.24	10	0.61	443	1	0.04	7	1430	23	0.96	<2	5
200796		1.52	<10	<1	0.13	<10	0.23	248	1	0.04	2	1280	3	0.58	<2	1
200797		2.56	10	<1	0.16	20	0.62	637	1	0.03	5	2100	13	1.21	<2	5
200798		2.94	<10	<1	0.21	<10	0.37	386	8	0.04	3	1100	74	2.11	<2	3
200799		2.11	<10	<1	0.14	<10	0.25	305	2	0.04	3	1380	6	1.42	2	2
200800		2.17	<10	<1	0.16	<10	0.26	317	2	0.04	3	1360	4	1.32	<2	2
200801		2.52	<10	<1	0.14	10	0.30	207	3	0.06	2	1710	<2	0.22	<2	2
200802		2.01	<10	<1	0.14	<10	0.17	121	3	0.04	2	1390	3	0.07	<2	1
200803		1.84	<10	<1	0.21	10	0.28	328	2	0.03	3	1740	3	0.43	<2	2
200804		1.39	<10	<1	0.17	10	0.11	187	3	0.02	4	1680	2	1.11	<2	1
200805		1.60	<10	<1	0.23	<10	0.31	266	5	0.02	4	1460	3	1.26	<2	2
200806		4.45	10	1	0.18	10	1.02	541	3	0.04	8	1430	5	0.75	<2	8
200807		3.30	<10	<1	0.16	<10	0.34	204	6	0.04	3	1810	4	0.26	<2	2
200808		1.78	<10	<1	0.21	10	0.33	211	3	0.05	5	1520	11	0.10	3	2
200809		1.87	<10	<1	0.13	10	0.20	189	1	0.02	4	1840	4	0.37	<2	1
200810		1.96	<10	<1	0.13	10	0.21	191	1	0.03	2	1860	3	0.36	<2	1
200811		1.79	<10	<1	0.14	10	0.21	218	<1	0.04	1	1850	2	0.38	2	1
200812		1.38	<10	<1	0.15	10	0.25	188	2	0.03	2	1320	<2	0.39	<2	1

To: **EQUITY ENGINEERING LTD.**
700 - 700 W. PENDER STREET
VANCOUVER BC V6C 1G8

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CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Sr	Th	Ti	Tl	U	V	W	Zn
	Units	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm
	LOR	1	20	0.01	10	10	1	10	2
200773		183	<20	0.12	<10	<10	70	<10	10
200774		139	<20	0.13	<10	<10	142	<10	9
200775		178	<20	0.14	<10	<10	141	10	18
200776		156	<20	0.12	<10	<10	96	10	13
200777		202	<20	0.11	<10	<10	164	10	11
200778		119	<20	0.12	<10	<10	102	10	15
200779		177	<20	0.12	<10	<10	156	<10	29
200780		1	<20	<0.01	<10	<10	<1	<10	2
200781		140	<20	0.14	<10	<10	99	10	15
200782		222	<20	0.12	<10	<10	105	<10	15
200783		195	<20	0.14	<10	<10	125	<10	16
200784		150	<20	0.13	<10	<10	112	10	13
200785		174	<20	0.16	<10	<10	144	<10	25
200786		194	<20	0.15	<10	<10	114	<10	25
200787		269	<20	0.14	<10	<10	140	<10	27
200788		144	<20	0.12	<10	10	64	10	12
200789		162	<20	0.14	<10	<10	51	10	7
200790		162	<20	0.14	<10	<10	51	<10	7
200791		163	<20	0.15	<10	<10	82	<10	11
200792		151	<20	0.11	<10	<10	80	<10	9
200793		103	<20	0.14	<10	<10	110	<10	8
200794		273	<20	0.05	<10	<10	132	<10	25
200795		758	<20	0.05	<10	<10	92	<10	43
200796		232	<20	0.12	<10	<10	67	<10	10
200797		838	<20	0.04	<10	<10	152	<10	29
200798		509	<20	0.08	<10	<10	70	<10	17
200799		193	<20	0.12	<10	<10	76	<10	8
200800		195	<20	0.13	<10	<10	82	<10	9
200801		309	<20	0.15	<10	<10	100	<10	12
200802		198	<20	0.13	<10	<10	78	<10	5
200803		143	<20	0.16	<10	<10	85	<10	7
200804		72	<20	0.14	<10	<10	52	20	2
200805		93	<20	0.15	<10	<10	60	20	5
200806		214	<20	0.09	<10	<10	180	<10	24
200807		378	<20	0.15	<10	<10	130	<10	10
200808		249	<20	0.16	<10	<10	90	<10	16
200809		219	<20	0.14	<10	<10	87	<10	8
200810		249	<20	0.14	<10	<10	91	<10	8
200811		248	<20	0.14	<10	<10	89	<10	11
200812		190	<20	0.12	<10	<10	71	<10	11

CERTIFICATE OF ANALYSIS VA07061776

	Method	WEI-21	Au-AA23	Au-GRA21	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Recvd Wt.	Au	Au	Ag	Al	As	B	Ba	Be	Bi	Ca	Cd	Co	Cr	Cu
	Units	kg	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
Sample Description	LOR	0.02	0.005	0.05	0.2	0.01	2	10	10	0.5	2	0.01	0.5	1	1	1
200813		2.38	0.126		0.2	0.65	4	<10	110	<0.5	2	1.75	<0.5	40	3	117
200814		3.28	0.010		0.2	0.61	<2	<10	120	<0.5	<2	1.25	<0.5	26	3	92
200815		3.42	0.024		0.2	0.76	5	<10	100	<0.5	<2	2.09	<0.5	4	4	93
200816		3.76	0.027		0.3	0.77	5	<10	60	<0.5	3	2.46	<0.5	8	3	88
200817		5.12	0.015		<0.2	0.76	4	<10	90	<0.5	2	1.63	<0.5	5	4	92
200818		1.34	0.009		<0.2	0.48	2	<10	70	<0.5	<2	1.53	<0.5	5	4	54
200819		1.14	0.019		0.3	0.98	<2	<10	70	<0.5	3	2.31	<0.5	5	4	221
200820		1.42	0.026		0.3	0.84	<2	<10	60	<0.5	2	2.52	<0.5	4	4	128
200821		3.86	0.015		0.2	0.74	6	<10	110	<0.5	2	1.80	<0.5	6	4	108
200822		3.10	0.005		<0.2	0.66	3	<10	90	<0.5	2	1.30	<0.5	5	4	66
200823		3.78	0.011		0.7	0.77	4	<10	100	<0.5	<2	1.54	<0.5	8	3	240
200824		3.22	0.007		0.3	0.90	4	<10	100	<0.5	2	1.58	<0.5	9	5	243
200825		3.28	0.011		0.2	0.65	2	<10	120	<0.5	<2	2.10	<0.5	6	4	103
200826		3.42	<0.005		<0.2	0.71	6	<10	70	<0.5	2	1.69	<0.5	5	4	82
200827		3.18	0.006		0.2	0.71	5	<10	70	<0.5	<2	2.07	<0.5	3	3	66
200828		3.78	0.009		0.2	0.70	6	<10	80	<0.5	3	1.92	<0.5	4	5	85
200829		1.60	0.104		0.8	1.19	7	<10	100	<0.5	5	2.10	<0.5	6	4	495
200830		1.42	0.082		0.6	1.04	4	<10	70	<0.5	4	2.00	<0.5	7	7	405
200831		0.84	0.747		4.2	0.93	7	<10	60	<0.5	119	0.79	<0.5	13	4	792
200832		4.34	<0.005		<0.2	1.69	4	<10	120	1.0	2	1.99	<0.5	11	23	19
200833		2.94	0.006		<0.2	1.66	2	<10	120	0.9	3	1.96	<0.5	10	21	24
200834		3.08	0.030		0.3	1.21	3	<10	80	<0.5	2	2.27	<0.5	6	4	163
200835		3.72	0.124		0.3	1.51	4	<10	80	<0.5	<2	2.76	<0.5	7	4	328
200836		1.64	0.048		0.4	1.59	6	<10	100	<0.5	2	2.34	<0.5	7	3	347
200837		1.36	<0.005		<0.2	0.05	<2	<10	10	<0.5	2	0.19	<0.5	<1	13	12
200838		1.44	0.006		0.2	0.04	4	<10	20	<0.5	<2	0.10	<0.5	<1	10	37
200839		0.84	<0.005		<0.2	0.07	<2	<10	20	<0.5	2	0.43	<0.5	<1	10	12
200840		0.82	0.008		<0.2	0.12	<2	<10	30	<0.5	2	0.47	<0.5	1	9	18
200841		1.84	0.044		3.2	0.56	<2	<10	150	<0.5	2	0.70	<0.5	3	9	163
200842		3.92	0.027		0.3	1.36	3	<10	100	0.5	<2	3.15	<0.5	8	3	150
200843		2.86	0.044		0.2	1.37	2	<10	90	0.5	2	3.06	<0.5	8	3	162
200844		3.26	0.032		<0.2	1.08	<2	<10	80	0.6	<2	2.03	<0.5	7	6	185
200845		3.86	0.035		0.8	1.23	<2	<10	160	0.7	<2	2.12	<0.5	8	4	1100
200846		2.24	0.016		0.7	1.80	<2	<10	60	0.7	3	2.21	<0.5	13	4	416
200847		2.98	0.044		0.8	1.78	<2	<10	160	0.8	2	2.44	<0.5	15	5	1260
200848		1.00	0.034		1.0	1.89	<2	<10	100	1.1	5	2.40	<0.5	25	10	967
200849		2.10	0.031		0.3	1.66	3	<10	40	0.7	2	2.54	<0.5	13	9	735
200850		2.24	0.033		0.4	1.66	<2	<10	50	0.7	2	2.58	<0.5	14	9	999
200851		3.00	0.031		0.3	1.77	<2	<10	30	0.7	4	2.92	<0.5	15	4	619
200852		1.94	0.013		<0.2	0.70	2	<10	210	<0.5	2	3.08	<0.5	4	3	243

To: **EQUITY ENGINEERING LTD.**
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CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Fe	Ga	Hg	K	La	Mg	Mn	Mo	Na	Ni	P	Pb	S	Sb	Sc
	Units	%	ppm	ppm	%	ppm	%	ppm	ppm	%	ppm	ppm	ppm	%	ppm	ppm
	LOR	0.01	10	1	0.01	10	0.01	5	1	0.01	1	10	2	0.01	2	1
200813		2.88	<10	<1	0.23	<10	0.45	266	3	0.03	4	760	3	1.86	<2	1
200814		2.06	<10	<1	0.25	<10	0.35	187	3	0.02	3	630	<2	0.92	<2	1
200815		2.01	<10	<1	0.15	10	0.39	247	1	0.04	2	1880	2	0.39	2	1
200816		4.34	<10	<1	0.10	10	0.39	306	4	0.02	5	2430	3	2.08	2	1
200817		1.92	<10	<1	0.15	10	0.33	189	1	0.04	3	2590	2	0.40	2	1
200818		2.06	<10	<1	0.14	<10	0.29	180	1	0.03	4	1720	2	0.50	<2	1
200819		2.31	10	<1	0.16	10	0.69	345	2	0.04	4	2090	<2	0.72	2	2
200820		2.01	10	<1	0.12	<10	0.58	317	2	0.02	4	1970	<2	0.67	2	2
200821		1.96	<10	<1	0.17	10	0.40	233	<1	0.04	3	1600	3	0.72	<2	1
200822		1.77	<10	<1	0.13	10	0.37	188	1	0.03	4	1750	2	0.51	<2	1
200823		2.81	<10	<1	0.17	<10	0.48	231	7	0.03	4	1740	2	1.02	3	1
200824		2.69	<10	<1	0.20	<10	0.68	310	<1	0.03	4	1720	11	1.04	<2	2
200825		2.63	<10	<1	0.19	<10	0.38	297	3	0.03	4	1330	5	1.83	<2	1
200826		3.16	<10	<1	0.15	<10	0.40	226	2	0.01	4	1840	2	0.12	<2	1
200827		4.18	<10	<1	0.15	10	0.14	236	1	0.02	3	2310	<2	0.04	3	1
200828		2.81	<10	<1	0.13	10	0.43	233	2	0.01	5	1510	3	0.16	2	1
200829		3.55	<10	<1	0.19	<10	0.89	358	5	0.02	5	1980	<2	1.05	<2	1
200830		3.35	10	<1	0.15	<10	0.81	328	4	0.01	5	1770	2	1.00	<2	1
200831		5.53	<10	<1	0.22	<10	0.60	283	2	<0.01	6	650	2	4.88	<2	1
200832		3.85	10	<1	0.11	30	1.22	608	2	0.08	10	2990	11	0.15	<2	6
200833		3.89	10	<1	0.11	40	1.19	589	2	0.09	9	3050	10	0.14	<2	6
200834		3.01	<10	<1	0.15	10	0.86	304	3	0.02	5	1760	3	0.76	3	2
200835		5.23	10	<1	0.16	<10	1.15	429	6	0.03	9	1580	3	0.91	<2	3
200836		3.92	10	<1	0.24	<10	1.13	443	2	0.01	7	1660	3	1.38	2	2
200837		0.23	<10	<1	0.02	<10	0.03	30	<1	<0.01	1	70	<2	0.02	2	<1
200838		0.23	<10	<1	0.02	<10	0.02	20	<1	<0.01	1	20	<2	0.09	<2	<1
200839		0.24	<10	<1	0.04	<10	0.03	44	<1	<0.01	1	40	<2	0.05	<2	<1
200840		0.39	<10	<1	0.06	<10	0.06	51	1	<0.01	1	100	<2	0.16	<2	<1
200841		1.27	<10	<1	0.16	<10	0.39	154	4	0.01	4	430	<2	0.35	<2	1
200842		4.04	10	<1	0.19	10	1.03	456	3	0.01	5	1670	3	0.64	<2	3
200843		4.02	10	<1	0.19	10	1.04	455	3	0.01	5	1630	<2	0.67	<2	3
200844		5.91	10	<1	0.09	10	1.02	348	2	0.03	9	1520	<2	0.05	<2	3
200845		4.10	10	<1	0.17	10	1.16	351	6	0.02	8	1340	<2	0.17	<2	4
200846		3.06	10	<1	0.08	10	1.87	449	2	0.03	8	1820	2	0.09	2	7
200847		6.89	10	<1	0.08	10	1.76	486	5	0.03	14	1650	<2	0.17	<2	7
200848		21.5	20	<1	0.13	<10	1.99	572	6	0.03	25	1130	2	0.11	<2	8
200849		4.00	10	<1	0.12	10	1.65	439	4	0.03	8	1740	<2	0.07	3	6
200850		5.00	10	<1	0.13	10	1.65	444	6	0.03	12	1750	<2	0.09	2	6
200851		3.82	10	<1	0.12	10	1.78	469	4	0.04	10	1780	<2	0.08	<2	5
200852		1.56	10	<1	0.12	10	0.61	272	7	0.03	5	1180	<2	0.22	2	3

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CERTIFICATE OF ANALYSIS VA07061776

	Method	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Sr	Th	Ti	Tl	U	V	W	Zn
	Units	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm
Sample Description	LOR	1	20	0.01	10	10	1	10	2
200813		108	<20	0.09	<10	<10	90	<10	18
200814		125	<20	0.09	<10	<10	79	<10	14
200815		244	<20	0.13	<10	<10	112	<10	13
200816		289	<20	0.13	<10	<10	198	<10	15
200817		228	<20	0.14	<10	<10	109	<10	14
200818		133	<20	0.13	<10	<10	107	<10	9
200819		216	<20	0.13	<10	<10	140	10	24
200820		214	<20	0.12	<10	<10	123	<10	19
200821		197	<20	0.14	<10	<10	96	<10	13
200822		178	<20	0.13	<10	<10	84	<10	13
200823		198	<20	0.12	<10	<10	107	<10	16
200824		200	<20	0.15	<10	<10	106	<10	23
200825		221	<20	0.13	<10	<10	90	<10	15
200826		243	<20	0.12	<10	<10	144	<10	11
200827		333	<20	0.14	<10	<10	295	<10	4
200828		224	<20	0.13	<10	<10	147	<10	10
200829		185	<20	0.12	<10	<10	120	10	27
200830		172	<20	0.11	<10	<10	112	10	17
200831		46	<20	0.05	<10	<10	38	10	17
200832		190	<20	0.37	<10	<10	95	<10	85
200833		184	<20	0.38	<10	<10	97	<10	73
200834		252	<20	0.13	<10	<10	118	10	19
200835		206	<20	0.15	<10	<10	201	<10	31
200836		142	<20	0.08	<10	<10	105	<10	27
200837		12	<20	0.01	<10	<10	5	<10	2
200838		8	<20	<0.01	<10	<10	3	<10	<2
200839		13	<20	<0.01	<10	<10	3	<10	<2
200840		16	<20	<0.01	<10	<10	5	<10	<2
200841		41	<20	0.02	<10	<10	33	10	10
200842		239	<20	0.10	<10	<10	146	<10	29
200843		231	<20	0.10	<10	<10	142	<10	25
200844		137	<20	0.09	<10	<10	197	<10	21
200845		173	<20	0.08	<10	<10	127	70	25
200846		169	<20	0.14	<10	<10	111	<10	31
200847		251	<20	0.14	<10	<10	212	<10	35
200848		231	<20	0.15	<10	<10	675	<10	40
200849		243	<20	0.17	<10	<10	143	<10	26
200850		242	<20	0.17	<10	<10	165	<10	25
200851		191	<20	0.15	<10	10	133	<10	27
200852		296	<20	0.05	<10	<10	48	<10	13

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CERTIFICATE OF ANALYSIS VA07061776

Method	WEI-21	Au-AA23	Au-GRA21	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
Analyte	Recvd Wt.	Au	Au	Ag	Al	As	B	Ba	Be	Bi	Ca	Cd	Co	Cr	Cu
Units	kg	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
Sample Description / LOR	0.02	0.005	0.05	0.2	0.01	2	10	10	0.5	2	0.01	0.5	1	1	1
200853	4.18	0.018		<0.2	1.16	<2	<10	220	0.5	<2	3.90	<0.5	10	4	336
200854	2.34	0.023		<0.2	1.78	<2	<10	110	<0.5	2	3.17	<0.5	16	6	322
200855	3.44	0.030		<0.2	0.76	<2	<10	300	<0.5	<2	2.68	<0.5	7	3	461

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CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method Analyte Units LOR	ME-ICP41 Fe % 0.01	ME-ICP41 Ga ppm 10	ME-ICP41 Hg ppm 1	ME-ICP41 K % 0.01	ME-ICP41 La ppm 10	ME-ICP41 Mg % 0.01	ME-ICP41 Mn ppm 5	ME-ICP41 Mo ppm 1	ME-ICP41 Na % 0.01	ME-ICP41 Ni ppm 1	ME-ICP41 P ppm 10	ME-ICP41 Pb ppm 2	ME-ICP41 S % 0.01	ME-ICP41 Sb ppm 2	ME-ICP41 Sc ppm 1
200853		4.59	10	<1	0.19	10	1.35	449	4	0.03	7	1420	<2	0.51	2	6
200854		4.05	10	<1	0.38	10	1.49	613	1	0.04	6	1690	<2	0.96	<2	6
200855		2.14	10	<1	0.25	10	0.86	333	6	0.03	1	1580	<2	0.84	<2	4

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CERTIFICATE OF ANALYSIS VA07061776

Sample Description	Method Analyte Units LOR	ME-ICP41 Sr ppm 1	ME-ICP41 Th ppm 20	ME-ICP41 Ti % 0.01	ME-ICP41 Tl ppm 10	ME-ICP41 U ppm 10	ME-ICP41 V ppm 1	ME-ICP41 W ppm 10	ME-ICP41 Zn ppm 2
200853		404	<20	0.03	<10	<10	116	<10	32
200854		291	<20	0.13	<10	<10	115	<10	49
200855		616	<20	0.03	<10	<10	49	<10	20

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Page: 1
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CERTIFICATE   VA07061791

Project: JW

P.O. No.: RG07-03

This report is for 81 Drill Core samples submitted to our lab in Vancouver, BC, Canada on 15-JUN-2007.

The following have access to data associated with this certificate:

QUITY ENGINEERING GENERA	RORY KUTLUOGLU	MICHAEL SMITH

SAMPLE PREPARATION

ALS CODE	DESCRIPTION
WEI-21	Received Sample Weight
LOG-22	Sample login - Rcd w/o BarCode
CRU-QC	Crushing QC Test
PUL-QC	Pulverizing QC Test
CRU-31	Fine crushing - 70% <2mm
SPL-21	Split sample - riffle splitter
PUL-31	Pulverize split to 85% <75 um
LOG-24	Pulp Login - Rcd w/o Barcode

ANALYTICAL PROCEDURES

ALS CODE	DESCRIPTION	INSTRUMENT
Au-AA23	Au 30g FA-AA finish	AAS
Au-GRA21	Au 30g FA-GRAV finish	WST-SIM
ME-ICP41	35 Element Aqua Regia ICP-AES	ICP-AES

To: **EQUITY ENGINEERING LTD.**
ATTN: EQUITY ENGINEERING GENERAL
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This is the Final Report and supersedes any preliminary report with this certificate number. Results apply to samples as submitted. All pages of this report have been checked and approved for release.

Signature: Lawrence Ng

Lawrence Ng, Laboratory Manager - Vancouver

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CERTIFICATE OF ANALYSIS VA07061791

Sample Description	Method Analyte Units LOR	WEI-21 Recvd Wt. kg 0.02	Au-AA23 Au ppm 0.005	Au-GRA21 Au ppm 0.05	ME-ICP41 Ag ppm 0.2	ME-ICP41 Al % 0.01	ME-ICP41 As ppm 2	ME-ICP41 B ppm 10	ME-ICP41 Ba ppm 10	ME-ICP41 Be ppm 0.5	ME-ICP41 Bi ppm 2	ME-ICP41 Ca % 0.01	ME-ICP41 Cd ppm 0.5	ME-ICP41 Co ppm 1	ME-ICP41 Cr ppm 1	ME-ICP41 Cu ppm 1
200856		4.56	0.057		0.5	0.52	<2	<10	240	<0.5	<2	2.91	<0.5	7	7	840
200857		3.64	0.021		<0.2	1.57	3	<10	110	0.6	<2	2.65	<0.5	5	7	166
200858		3.66	0.017		<0.2	1.52	3	<10	40	0.5	<2	2.58	<0.5	6	9	9
200859		1.88	0.008		<0.2	1.58	<2	<10	30	0.6	<2	2.19	<0.5	10	2	45
200860		1.58	0.007		<0.2	1.72	5	<10	30	0.6	<2	2.15	<0.5	10	1	46
200861		3.50	0.018		<0.2	1.28	4	<10	130	0.5	<2	4.49	<0.5	9	3	23
200862		3.48	0.031		0.4	1.70	4	<10	90	0.6	<2	3.42	<0.5	9	3	537
200863		3.92	>10.0	28.9	6.4	1.79	<2	<10	90	0.6	<2	4.30	<0.5	9	8	315
200864		0.68	>10.0	46.7	6.5	1.54	4	<10	30	0.5	<2	4.44	<0.5	12	6	421
200865		3.72	0.178		<0.2	1.77	5	<10	150	0.6	<2	4.37	<0.5	8	8	24
200866		3.42	0.771		0.2	1.63	<2	<10	90	0.7	<2	4.00	<0.5	7	5	71
200867		3.10	0.049		0.3	1.38	2	<10	140	0.5	<2	3.30	<0.5	7	7	560
200868		3.20	0.044		<0.2	1.30	4	<10	130	<0.5	<2	3.31	<0.5	6	3	58
200869		1.70	0.022		0.2	1.43	9	<10	80	0.6	<2	3.56	<0.5	7	6	59
200870		1.54	0.029		<0.2	1.40	<2	<10	70	0.6	<2	3.76	<0.5	7	4	45
200871		0.76	0.020		<0.2	1.27	4	<10	170	0.5	<2	3.12	<0.5	7	9	59
200872		2.02	0.015		<0.2	1.72	4	<10	70	0.7	<2	4.01	<0.5	10	15	31
200873		3.64	0.013		<0.2	1.30	<2	<10	50	0.5	<2	3.06	<0.5	8	8	98
200874		3.00	0.071		0.3	1.36	11	<10	70	<0.5	<2	2.75	<0.5	9	8	78
200875		5.30	0.011		0.3	2.43	6	<10	130	0.5	<2	2.42	<0.5	22	21	313
200876		4.38	0.019		0.3	2.49	10	<10	100	0.5	<2	2.36	<0.5	25	13	404
200877		4.84	0.020		0.3	2.32	7	<10	100	0.5	<2	3.24	<0.5	27	12	353
200878		3.64	0.036		<0.2	2.61	6	<10	80	<0.5	<2	2.78	<0.5	20	32	156
200879		1.72	0.032		<0.2	2.18	17	<10	110	<0.5	<2	2.94	<0.5	12	16	161
200880		1.34	0.032		0.2	2.24	7	<10	80	<0.5	<2	3.03	<0.5	14	15	150
200881		3.84	0.034		0.3	2.30	5	<10	100	<0.5	<2	3.09	<0.5	31	11	319
200882		3.14	0.066		<0.2	1.88	7	<10	90	<0.5	<2	2.01	<0.5	31	11	222
200883		3.16	0.018		0.3	2.23	7	<10	120	<0.5	<2	2.11	<0.5	28	14	239
200884		1.98	0.025		0.3	1.73	8	<10	90	<0.5	<2	1.85	<0.5	30	12	192
200885		1.78	0.022		0.2	1.29	2	<10	40	<0.5	<2	0.85	<0.5	22	5	229
200886		3.20	0.025		0.6	2.91	<2	<10	110	0.5	<2	2.13	<0.5	18	21	627
200887		4.02	0.018		0.3	3.25	3	<10	90	0.6	<2	1.79	<0.5	22	25	392
200888		3.48	0.023		0.6	3.06	2	<10	70	0.5	<2	2.03	<0.5	36	17	557
200889		1.90	0.015		0.3	2.83	<2	<10	140	0.5	<2	2.39	<0.5	28	23	400
200890		1.92	0.013		0.6	2.74	4	<10	110	0.5	<2	2.45	<0.5	30	19	415
200891		3.58	0.016		0.3	1.84	<2	<10	160	<0.5	<2	1.97	<0.5	16	15	284
200892		3.32	0.011		0.2	2.26	2	<10	140	<0.5	<2	2.12	<0.5	20	24	306
200893		3.50	0.017		0.5	2.64	3	<10	130	<0.5	<2	2.47	<0.5	40	27	616
200894		3.84	0.014		0.3	2.80	<2	<10	120	<0.5	<2	2.87	<0.5	23	24	380
200895		3.70	0.012		0.3	2.60	5	<10	80	<0.5	<2	2.74	<0.5	31	22	297

ALS Chemex
EXCELLENCE IN ANALYTICAL CHEMISTRY
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To: **EQUITY ENGINEERING LTD.**
700 - 700 W. PENDER STREET
VANCOUVER BC V6C 1G8

Total # Pages: 4 (A - C)
Finalized Date: 29-JUN-2007
Account: EIARG

Project: JW

CERTIFICATE OF ANALYSIS VA07061791

Sample Description	Method Analyte Units LOR	ME-ICP41 Fe % 0.01	ME-ICP41 Ga ppm 10	ME-ICP41 Hg ppm 1	ME-ICP41 K % 0.01	ME-ICP41 La ppm 10	ME-ICP41 Mg % 0.01	ME-ICP41 Mn ppm 5	ME-ICP41 Mo ppm 1	ME-ICP41 Na % 0.01	ME-ICP41 Ni ppm 1	ME-ICP41 P ppm 10	ME-ICP41 Pb ppm 2	ME-ICP41 S % 0.01	ME-ICP41 Sb ppm 2	ME-ICP41 Sc ppm 1
200856		2.45	<10	1	0.37	10	0.73	285	7	0.03	2	1390	5	0.83	<2	3
200857		5.22	10	1	0.14	10	1.59	323	<1	0.05	8	1080	2	0.50	<2	6
200858		5.29	10	1	0.07	10	1.43	308	<1	0.07	8	1910	<2	0.55	<2	6
200859		7.05	10	1	0.08	<10	1.51	316	1	0.06	10	2020	4	0.33	2	6
200860		7.93	10	1	0.08	<10	1.64	340	1	0.07	9	1960	2	0.31	<2	7
200861		4.75	10	1	0.14	10	1.27	360	<1	0.04	7	2050	3	1.05	<2	5
200862		4.40	10	2	0.15	10	1.55	417	1	0.05	5	2260	2	0.61	<2	6
200863		5.12	10	<1	0.33	10	1.43	468	3	0.04	7	2490	2	2.13	2	6
200864		4.67	10	<1	0.45	10	0.97	500	9	0.02	5	2200	6	3.52	<2	5
200865		5.20	10	2	0.23	10	1.53	488	1	0.06	8	1800	<2	0.46	<2	9
200866		4.38	10	2	0.08	10	1.56	395	4	0.05	8	2310	<2	1.55	4	7
200867		3.50	10	1	0.16	10	1.31	415	3	0.03	4	1920	<2	0.74	<2	5
200868		3.56	10	<1	0.13	10	1.22	396	1	0.04	5	2090	<2	0.92	<2	4
200869		4.83	10	1	0.15	10	1.23	382	2	0.05	8	2070	4	0.86	<2	6
200870		4.53	10	<1	0.13	10	1.22	383	4	0.04	8	2170	2	0.92	2	7
200871		5.00	10	1	0.15	10	1.16	362	2	0.05	10	1660	2	0.79	<2	6
200872		8.85	10	1	0.08	10	1.73	561	2	0.06	13	870	3	0.28	<2	13
200873		6.19	10	1	0.08	10	1.30	402	2	0.04	11	1610	<2	0.74	<2	6
200874		5.19	10	1	0.12	10	1.15	365	1	0.07	11	2220	6	0.40	<2	5
200875		4.61	10	1	1.76	<10	2.30	714	3	0.05	13	1960	3	1.60	<2	6
200876		4.97	10	<1	1.64	<10	2.32	696	1	0.03	11	2320	2	2.36	3	6
200877		4.99	10	1	1.23	10	2.19	779	1	0.06	11	2280	<2	2.85	<2	10
200878		5.61	10	1	1.94	<10	2.59	1050	4	0.05	17	2120	<2	3.26	4	6
200879		5.21	10	1	1.50	<10	2.18	915	10	0.06	10	2080	2	3.57	<2	6
200880		5.62	10	1	1.63	<10	2.29	960	10	0.04	11	2080	<2	4.06	<2	6
200881		5.83	10	1	1.26	<10	2.30	1160	5	0.06	12	2200	<2	4.02	<2	5
200882		6.28	10	1	1.12	<10	2.07	729	6	0.04	11	1990	<2	5.31	<2	7
200883		4.35	10	1	1.46	<10	2.28	771	9	0.08	10	2280	<2	2.77	<2	5
200884		4.07	10	2	1.26	<10	1.79	627	10	0.06	11	1990	<2	3.31	<2	4
200885		4.33	<10	1	0.82	<10	1.16	354	12	0.09	3	1520	<2	4.16	<2	3
200886		5.38	10	1	2.26	<10	2.84	1210	18	0.08	13	2400	<2	2.06	<2	7
200887		6.56	10	2	2.64	<10	2.85	1070	2	0.06	15	2280	3	1.92	<2	5
200888		6.06	<10	<1	2.41	<10	2.67	1030	10	0.06	14	2270	<2	2.41	<2	6
200889		5.26	10	1	2.28	<10	2.49	956	33	0.06	15	2250	<2	1.56	<2	7
200890		5.33	10	1	2.22	<10	2.41	947	39	0.06	14	2270	<2	1.81	<2	6
200891		3.23	10	1	1.30	<10	1.75	819	2	0.07	10	2040	<2	1.17	<2	5
200892		3.90	10	2	1.54	<10	2.25	1110	9	0.09	16	2390	2	1.86	<2	6
200893		4.96	10	1	2.08	<10	2.68	1380	8	0.06	16	2220	<2	2.80	<2	7
200894		4.80	10	1	2.25	<10	2.68	1520	5	0.06	14	2110	<2	1.91	<2	6
200895		4.81	<10	1	1.85	<10	2.51	1340	5	0.06	15	2240	<2	2.33	2	8

CERTIFICATE OF ANALYSIS VA07061791

Sample Description	Method	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Sr	Th	Ti	Tl	U	V	W	Zn
	Units	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm
	LOR	1	20	0.01	10	10	1	10	2
200856		546	<20	0.01	<10	<10	40	<10	13
200857		424	<20	0.11	<10	<10	185	<10	22
200858		334	<20	0.13	<10	<10	185	<10	20
200859		318	<20	0.14	<10	<10	230	<10	21
200860		322	<20	0.16	<10	<10	257	<10	22
200861		546	<20	0.12	<10	<10	181	<10	22
200862		470	<20	0.14	<10	<10	168	<10	30
200863		512	<20	0.06	<10	<10	172	10	26
200864		703	<20	0.02	<10	<10	82	<10	18
200865		556	<20	0.10	<10	<10	181	<10	22
200866		653	<20	0.14	<10	<10	177	<10	20
200867		395	<20	0.13	<10	<10	151	<10	22
200868		393	<20	0.11	<10	<10	159	<10	20
200869		384	<20	0.12	<10	<10	222	<10	21
200870		414	<20	0.12	<10	<10	217	<10	21
200871		609	<20	0.16	<10	<10	242	<10	19
200872		488	<20	0.22	<10	<10	414	<10	27
200873		417	<20	0.17	<10	<10	275	<10	20
200874		463	<20	0.18	<10	<10	220	<10	22
200875		200	<20	0.25	<10	<10	157	<10	41
200876		267	<20	0.25	<10	<10	159	<10	42
200877		213	<20	0.23	<10	<10	168	<10	31
200878		164	<20	0.24	<10	<10	130	<10	34
200879		137	<20	0.22	<10	<10	134	<10	31
200880		128	<20	0.21	<10	<10	135	<10	32
200881		167	<20	0.21	<10	<10	124	<10	38
200882		92	<20	0.20	<10	<10	132	<10	30
200883		113	<20	0.23	<10	<10	134	<10	37
200884		86	<20	0.20	<10	<10	103	<10	30
200885		67	<20	0.18	<10	<10	98	<10	20
200886		129	<20	0.28	<10	<10	168	<10	49
200887		147	<20	0.28	<10	<10	187	<10	47
200888		124	<20	0.27	<10	<10	156	<10	43
200889		158	<20	0.26	<10	<10	152	<10	39
200890		155	<20	0.25	<10	<10	148	<10	38
200891		129	<20	0.21	<10	<10	113	<10	31
200892		132	<20	0.24	<10	<10	141	<10	38
200893		128	<20	0.24	<10	<10	138	<10	45
200894		148	<20	0.25	<10	<10	147	<10	43
200895		151	<20	0.24	<10	<10	139	<10	42

To: **EQUITY ENGINEERING LTD.**
700 - 700 W. PENDER STREET
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CERTIFICATE OF ANALYSIS VA07061791

Method	WEI-21	Au-AA23	Au-GRA21	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
Analyte	Recvd Wt.	Au	Au	Ag	Al	As	B	Ba	Be	Bi	Ca	Cd	Co	Cr	Cu
Units	kg	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
Sample Description LOR	0.02	0.005	0.05	0.2	0.01	2	10	10	0.5	2	0.01	0.5	1	1	1
200896	3.92	<0.005		<0.2	2.91	2	<10	130	<0.5	<2	3.32	<0.5	16	24	151
200897	3.38	0.006		<0.2	2.81	<2	<10	120	<0.5	<2	3.26	<0.5	13	27	114
200898	3.86	0.017		0.2	2.49	10	<10	100	<0.5	<2	3.10	<0.5	23	20	232
200899	1.84	0.014		0.2	2.42	7	<10	60	<0.5	<2	3.10	<0.5	27	17	142
200900	1.56	0.014		<0.2	2.29	<2	<10	60	<0.5	<2	3.11	<0.5	26	13	149
200901	3.90	0.046		0.3	2.55	6	<10	40	<0.5	<2	2.07	<0.5	31	11	483
200902	5.60	0.038		0.4	2.83	6	<10	70	<0.5	<2	3.39	<0.5	28	19	549
200903	1.86	0.048		0.4	2.52	8	<10	60	<0.5	<2	2.77	<0.5	24	11	383
200904	3.62	0.040		0.4	2.49	2	<10	50	<0.5	<2	2.70	<0.5	27	11	518
200905	3.32	0.039		0.4	2.49	10	<10	100	<0.5	<2	2.90	<0.5	22	12	414
200906	3.56	0.031		0.2	1.67	3	<10	120	<0.5	<2	2.85	<0.5	18	7	269
200907	0.82	0.011		<0.2	0.95	<2	<10	60	<0.5	<2	2.55	<0.5	3	4	6
200908	3.92	0.010		0.2	2.03	8	<10	150	<0.5	<2	2.85	<0.5	12	10	173
200909	1.90	0.008		<0.2	2.02	8	<10	140	<0.5	<2	2.61	<0.5	17	11	177
200910	1.30	0.015		<0.2	1.66	<2	<10	120	<0.5	<2	2.44	<0.5	14	9	179
200911	3.98	0.018		0.2	2.24	4	<10	190	<0.5	4	2.84	<0.5	20	15	343
200912	3.52	0.021		0.2	1.80	<2	<10	150	<0.5	3	2.95	<0.5	17	7	369
200913	3.60	0.024		0.5	2.52	<2	<10	250	0.5	3	2.47	<0.5	12	22	603
200914	3.22	0.029		0.4	2.44	<2	<10	250	0.5	4	2.14	<0.5	16	9	629
200915	3.10	0.056		0.3	2.11	2	<10	180	<0.5	3	2.06	<0.5	16	5	645
200916	3.44	0.054		0.3	1.74	5	<10	140	<0.5	3	2.08	<0.5	23	5	621
200917	3.44	0.037		<0.2	1.68	9	<10	160	<0.5	<2	1.75	<0.5	11	6	88
200918	2.86	0.031		<0.2	2.59	<2	<10	220	<0.5	4	1.85	<0.5	13	17	535
200919	1.92	0.112		0.2	2.33	<2	<10	50	<0.5	4	2.02	<0.5	30	23	202
200920	0.10	<0.005		<0.2	0.01	<2	<10	10	<0.5	<2	0.01	<0.5	<1	<1	1
200921	2.44	0.117		0.5	2.11	12	<10	30	<0.5	3	1.96	<0.5	17	11	796
200922	3.22	0.034		0.3	2.27	<2	<10	220	<0.5	<2	2.29	<0.5	11	11	695
200923	3.26	0.008		<0.2	1.75	<2	<10	150	<0.5	<2	2.13	<0.5	11	6	95
200924	3.10	0.008		<0.2	2.09	<2	<10	160	<0.5	2	2.20	<0.5	18	10	112
200925	3.68	0.018		<0.2	2.35	4	<10	170	0.5	<2	2.39	<0.5	13	13	89
200926	3.50	0.007		<0.2	2.69	<2	<10	150	0.6	2	2.52	<0.5	6	26	90
200927	3.38	<0.005		<0.2	2.68	<2	<10	160	0.6	<2	2.59	<0.5	9	18	91
200928	3.42	0.021		<0.2	2.14	<2	<10	150	<0.5	2	1.95	<0.5	21	19	88
200929	1.40	0.017		0.2	2.16	3	<10	100	0.5	4	2.22	<0.5	21	20	105
200930	1.30	0.010		<0.2	2.07	5	<10	100	<0.5	2	2.18	<0.5	23	20	138
200931	4.88	0.014		<0.2	1.94	<2	<10	50	<0.5	<2	3.20	<0.5	27	28	207
200932	1.74	0.013		<0.2	2.09	<2	<10	90	0.6	2	3.20	<0.5	16	18	145
200933	2.96	0.015		<0.2	2.18	6	<10	130	<0.5	<2	1.87	<0.5	27	20	259
200934	2.60	0.022		<0.2	2.45	<2	<10	160	0.5	3	2.12	<0.5	22	19	293
200935	2.78	0.039		0.3	1.61	<2	<10	100	<0.5	2	1.24	<0.5	35	4	544

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CERTIFICATE OF ANALYSIS VA07061791

Method	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
Analyte	Fe	Ga	Hg	K	La	Mg	Mn	Mo	Na	Ni	P	Pb	S	Sb	Sc
Units	%	ppm	ppm	%	ppm	%	ppm	ppm	%	ppm	ppm	ppm	%	ppm	ppm
Sample Description LOR	0.01	10	1	0.01	10	0.01	5	1	0.01	1	10	2	0.01	2	1
200896	4.51	10	2	2.22	<10	2.68	1360	4	0.05	16	2180	<2	1.24	<2	6
200897	4.33	10	1	2.00	<10	2.70	1440	<1	0.06	13	2270	<2	1.73	<2	6
200898	4.36	10	1	1.57	<10	2.38	1620	<1	0.07	15	2070	<2	2.43	3	5
200899	5.44	10	<1	1.43	<10	2.35	1480	3	0.06	12	2310	<2	3.97	2	5
200900	5.00	10	2	1.49	<10	2.21	1450	2	0.05	10	2210	<2	3.51	<2	5
200901	6.24	10	1	1.49	<10	2.58	1180	1	0.05	9	2300	<2	4.67	<2	4
200902	5.52	<10	1	1.67	<10	2.66	1650	1	0.05	13	2410	<2	3.18	<2	6
200903	5.43	<10	1	1.53	<10	2.32	1600	<1	0.04	10	2330	<2	3.39	<2	4
200904	5.63	10	1	1.58	<10	2.31	1500	<1	0.06	9	2460	<2	3.93	<2	5
200905	5.17	<10	2	1.58	<10	2.38	1460	<1	0.05	9	2460	<2	3.46	2	5
200906	3.81	<10	<1	1.03	<10	1.46	1030	1	0.06	6	1590	<2	2.63	<2	3
200907	1.63	<10	1	0.49	<10	0.67	579	<1	0.08	<1	830	<2	0.60	<2	2
200908	3.79	10	<1	1.46	<10	1.86	936	<1	0.06	7	1770	2	2.00	<2	5
200909	3.96	<10	1	1.48	<10	1.84	944	5	0.05	7	2020	<2	2.00	<2	5
200910	3.45	10	<1	1.20	10	1.49	840	5	0.05	6	1780	<2	1.72	<2	3
200911	5.04	10	<1	1.56	10	1.97	1075	9	0.08	13	1860	<2	2.50	3	5
200912	4.34	10	<1	1.17	10	1.46	802	2	0.06	6	1610	<2	1.84	<2	3
200913	5.67	10	<1	1.90	10	2.00	1105	1	0.06	12	1850	2	0.68	<2	5
200914	6.39	10	<1	1.93	10	1.97	1120	<1	0.06	10	2140	<2	0.86	<2	4
200915	6.75	10	<1	1.54	10	1.65	1235	7	0.06	4	1970	<2	2.19	<2	4
200916	5.20	10	<1	1.19	10	1.46	805	6	0.05	7	1630	<2	3.13	<2	4
200917	4.79	10	<1	1.11	10	1.34	614	5	0.07	3	1580	<2	2.58	<2	4
200918	5.54	10	<1	2.04	<10	2.32	919	5	0.05	18	2060	<2	1.62	<2	4
200919	7.93	10	<1	1.69	<10	2.25	994	12	0.06	18	1900	<2	6.06	<2	5
200920	0.02	<10	<1	<0.01	<10	0.01	<5	<1	0.01	<1	20	<2	0.02	<2	<1
200921	9.11	10	<1	1.38	10	2.01	719	9	0.06	10	1920	<2	7.44	<2	5
200922	5.31	10	2	1.41	10	1.93	777	9	0.07	8	1930	<2	1.08	<2	5
200923	4.34	10	<1	1.02	10	1.42	708	3	0.06	5	1780	<2	0.90	<2	3
200924	4.68	10	<1	1.40	10	1.72	978	4	0.07	8	1830	<2	1.86	<2	4
200925	4.45	10	<1	1.59	10	2.16	945	6	0.07	12	2140	2	1.27	<2	6
200926	5.80	10	<1	1.11	10	2.46	872	4	0.06	19	2000	<2	0.34	<2	9
200927	5.15	10	<1	1.04	10	2.54	895	2	0.06	13	2280	<2	0.44	2	9
200928	5.06	<10	<1	0.54	10	1.81	866	3	0.07	16	1760	2	2.03	<2	5
200929	5.33	10	<1	0.28	10	1.93	931	6	0.07	14	2040	4	0.64	<2	7
200930	5.18	10	<1	0.27	10	1.82	889	7	0.07	11	2000	<2	0.88	<2	7
200931	5.46	10	<1	0.31	10	2.11	935	5	0.05	19	2070	3	1.85	<2	10
200932	5.23	10	<1	0.88	10	2.04	728	2	0.05	14	1920	<2	2.30	<2	12
200933	4.99	10	<1	1.40	<10	2.02	691	5	0.08	13	2070	<2	2.02	<2	5
200934	5.15	10	<1	1.60	10	2.12	814	7	0.09	12	2040	<2	1.34	<2	6
200935	4.46	10	<1	1.02	<10	1.44	490	11	0.07	6	2010	<2	2.95	<2	4

CERTIFICATE OF ANALYSIS VA07061791

	Method	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41	ME-ICP41
	Analyte	Sr	Th	Ti	Tl	U	V	W	Zn
	Units	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm
Sample Description	LOR	1	20	0.01	10	10	1	10	2
200896		197	<20	0.25	<10	<10	145	<10	44
200897		159	<20	0.24	<10	<10	132	<10	44
200898		146	<20	0.22	<10	<10	130	<10	40
200899		151	<20	0.21	<10	<10	138	<10	36
200900		152	<20	0.21	<10	<10	137	<10	36
200901		115	<20	0.20	<10	<10	130	<10	36
200902		132	<20	0.23	<10	<10	169	<10	41
200903		123	<20	0.22	<10	<10	143	<10	41
200904		127	<20	0.22	<10	<10	151	<10	41
200905		146	<20	0.23	<10	<10	155	<10	40
200906		134	<20	0.18	<10	<10	107	<10	23
200907		102	<20	0.13	<10	<10	79	<10	11
200908		139	<20	0.21	<10	<10	125	<10	26
200909		144	<20	0.20	<10	<10	122	<10	28
200910		107	<20	0.15	<10	<10	99	<10	25
200911		147	<20	0.22	<10	<10	139	<10	29
200912		149	<20	0.17	<10	<10	117	<10	23
200913		165	<20	0.24	<10	<10	150	<10	34
200914		142	<20	0.23	<10	10	167	<10	37
200915		121	<20	0.24	<10	<10	160	<10	35
200916		97	<20	0.17	<10	<10	118	<10	24
200917		91	<20	0.17	<10	<10	121	<10	21
200918		107	<20	0.20	<10	10	140	10	30
200919		89	<20	0.19	<10	<10	127	10	32
200920		<1	<20	<0.01	<10	<10	<1	<10	2
200921		81	<20	0.20	<10	<10	137	<10	28
200922		141	<20	0.23	<10	<10	149	<10	29
200923		139	<20	0.19	<10	<10	125	<10	27
200924		157	<20	0.21	<10	<10	126	<10	28
200925		151	<20	0.20	<10	<10	129	<10	32
200926		203	<20	0.20	<10	<10	155	<10	35
200927		217	<20	0.21	<10	<10	152	<10	40
200928		167	<20	0.19	<10	<10	105	<10	54
200929		172	<20	0.21	<10	<10	141	<10	56
200930		174	<20	0.21	<10	<10	134	<10	54
200931		144	<20	0.18	<10	<10	171	<10	31
200932		148	<20	0.21	10	<10	170	<10	25
200933		114	<20	0.23	<10	10	155	<10	25
200934		141	<20	0.25	<10	<10	164	<10	27
200935		105	<20	0.23	<10	10	135	10	25

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Page: 4 - A
Total # Pages: 4 (A - C)
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CERTIFICATE OF ANALYSIS VA07061791

Sample Description	Method Analyte Units LOR	WEI-21 Recvd Wt. kg 0.02	Au-AA23 Au ppm 0.005	Au-GRA21 Au ppm 0.05	ME-ICP41 Ag ppm 0.2	ME-ICP41 Al % 0.01	ME-ICP41 As ppm 2	ME-ICP41 B ppm 10	ME-ICP41 Ba ppm 10	ME-ICP41 Be ppm 0.5	ME-ICP41 Bi ppm 2	ME-ICP41 Ca % 0.01	ME-ICP41 Cd ppm 0.5	ME-ICP41 Co ppm 1	ME-ICP41 Cr ppm 1	ME-ICP41 Cu ppm 1
200936		1.58	0.057		0.5	1.18	<2	<10	80	<0.5	4	1.17	<0.5	26	7	721

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Account: EIARG

Project: JW

CERTIFICATE OF ANALYSIS VA07061791

Sample Description	Method Analyte Units LOR	ME-ICP41 Fe % 0.01	ME-ICP41 Ga ppm 10	ME-ICP41 Hg ppm 1	ME-ICP41 K % 0.01	ME-ICP41 La ppm 10	ME-ICP41 Mg % 0.01	ME-ICP41 Mn ppm 5	ME-ICP41 Mo ppm 1	ME-ICP41 Na % 0.01	ME-ICP41 Ni ppm 1	ME-ICP41 P ppm 10	ME-ICP41 Pb ppm 2	ME-ICP41 S % 0.01	ME-ICP41 Sb ppm 2	ME-ICP41 Sc ppm 1
200936		3.73	<10	<1	0.48	<10	1.06	403	4	0.07	7	1700	3	2.70	<2	3

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CERTIFICATE OF ANALYSIS VA07061791

Sample Description	Method Analyte Units LOR	ME-ICP41 Sr ppm 1	ME-ICP41 Th ppm 20	ME-ICP41 Ti % 0.01	ME-ICP41 Tl ppm 10	ME-ICP41 U ppm 10	ME-ICP41 V ppm 1	ME-ICP41 W ppm 10	ME-ICP41 Zn ppm 2
200936		89	<20	0.19	<10	<10	99	<10	20

Appendix E: Quality Control / Quality Assurance

QUALITY CONTROL / QUALITY ASSURANCE

I Chain of Custody

All JW core samples were packed in rice sacks and sealed with uniquely-numbered non-resealable security straps. Rice sacks were trucked to ALS Chemex Labs in North Vancouver. ALS Chemex reported that all bags were received in good condition, with all security straps intact, and with no evidence of tampering.

II Blanks

Blanks are samples which are known to be barren of mineralization and are inserted into the sample stream to determine whether contamination has occurred after sample collection.

a. Core Blanks

Two blanks were inserted into the core sample sequence and submitted for analysis. The blank material comprised commercially available silica sand of the type used in swimming pool filters. Blanks were inserted into the sample series in the field.

	Au (ppb)	Ag (ppm)	As (ppm)	Bi (ppm)	Cu (ppm)	Mo (ppm)	Pb (ppm)	Sb (ppm)	Zn (ppm)
200780	<5	<0.2	2	<2	3	<1	<2	<2	2
200920	<5	<0.2	<2	<2	1	<1	<2	<2	2

Results of JW core blanks indicate the silica sand was suitably devoid of metals and served as good blank material. Secondly, the consistently low values for all metals of interest (see table above), indicate that contamination of the core samples did not take place in the field, or in the lab.

III Field Duplicates

Field duplicates are collected and analysed as two separate samples from the same field location. They are used to measure the reproducibility of sampling, which includes both laboratory variation and sample variation. A total of 21 duplicate core samples (approximately every 20th sample) were collected during the 2007 JW drill program, as two quarters of a single interval, and submitted for analysis.

Most of the elements of interest are reproducible at a level of 20% (Ag, Mo, Pb, Zn), 30% (Au) or 40% (Cu) precision. However, As is only reproducible at a level of 160%. These data indicate that the "nugget effect" (variable metal distribution within samples), should not be a major issue for the main metals of interest (Au, Cu, Ag and Mo) on the JW property, but could pose problems in As analysis.

IV Conclusions

- No interference with samples took place between the JW property and the analytical laboratory.
- The blanks indicate that no sample contamination took place prior to analysis.
- The field duplicates indicate that core samples are reproducible at satisfactory levels of precision for all elements of interest except arsenic. The "nugget effect" should not be a concern for any of the elements of economic interest.
- Although not presented here, ALS Chemex carries out a full QA/QC protocol, including blanks, duplicates and standards, on laboratory handling and analysis of samples and satisfy themselves that results are satisfactory, prior to issuing certificates.

Appendix F: Compact Disc

Report text, geochemical databases, drafting and plot files, photographs

Appendix G: Geologist’s Certificate

Rory Kutluoglu, B.Sc.
5-1214 W 7th Ave.
Vancouver, B. C., Canada
V6H 1G6
roryk@equityeng.bc.ca

I, Rory Kutluoglu, do hereby certify:

THAT I am a Geoscientist employed by Equity Engineering Ltd., with offices at #700-700 West Pender Street in the City of Vancouver, in the Province of British Columbia.

THAT I am a graduate of Lakehead University (2004) with a Bachelor of Science degree in Geology and I have practiced my profession continuously since 2003.

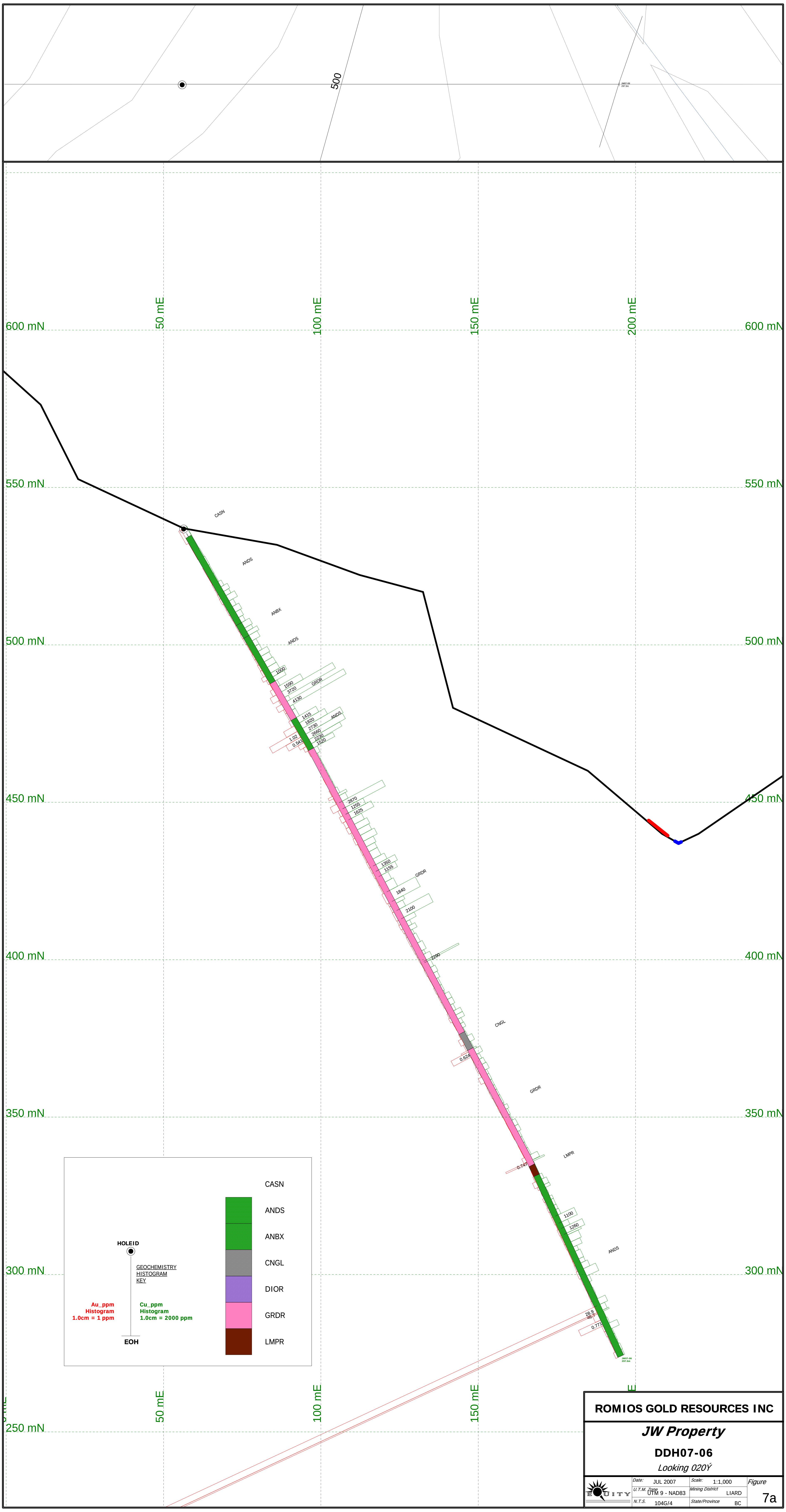
THAT I am presently a Project Geologist and have been so since April 2006.

THAT this report is based on publicly-available reports and maps and on fieldwork carried out under my direction in June 2007.

Dated at Vancouver, British Columbia, this 1st day of October, 2007.

Rory Kutluoglu, B.Sc. (Geology)

Appendix H: Maps



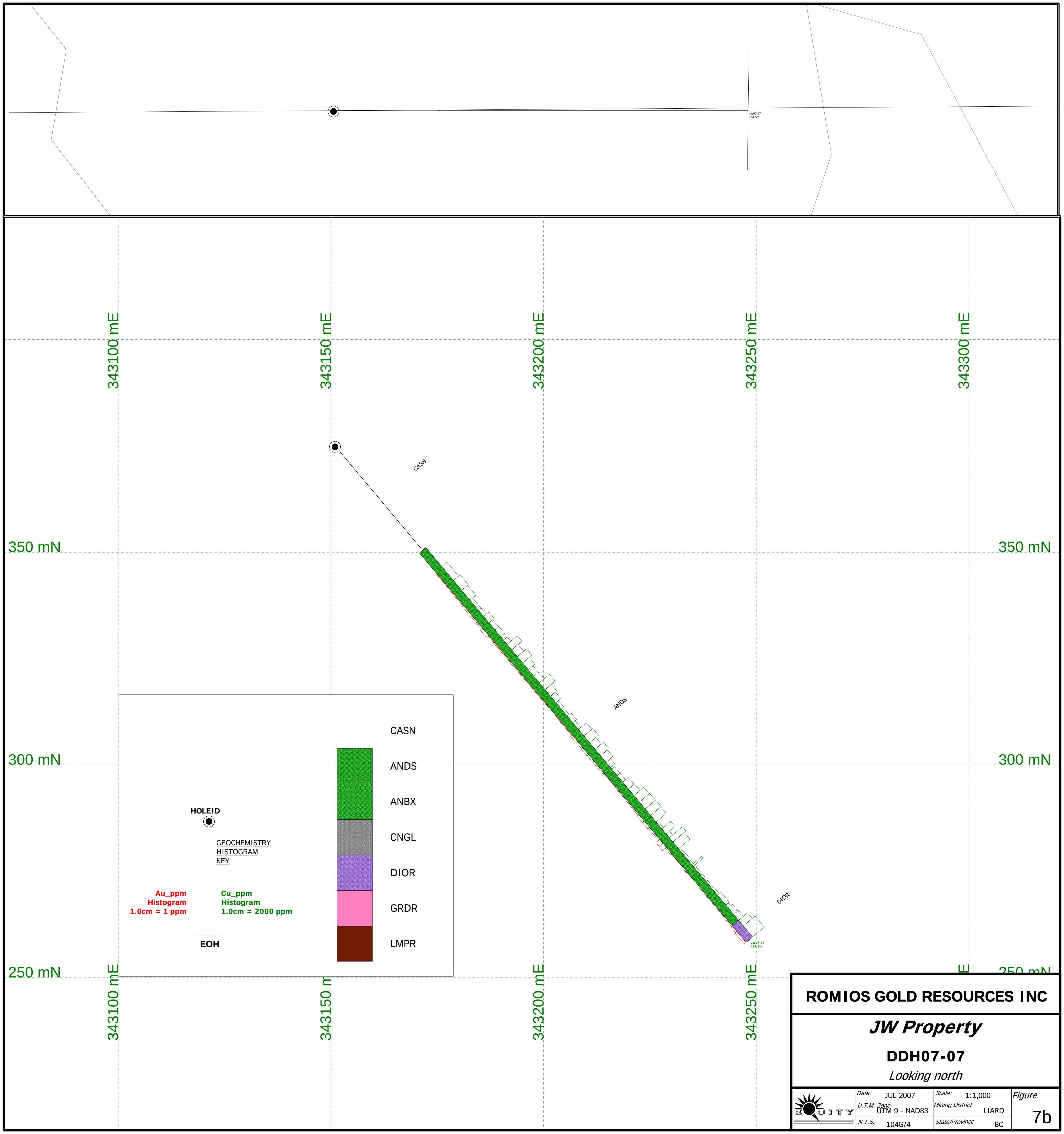
ROMIOS GOLD RESOURCES INC

JW Property

DDH07-06

Looking 020Ý

Date: JUL 2007	Scale: 1:1,000		Figure
U.T.M. Zone UTM 9 - NAD83	Mining District LIARD		7a
N.T.S. 104G/4	State/Province BC		



ROMIOS GOLD RESOURCES INC

JW Property

DDH07-07

Looking north

Date: JUL 2007	*Scale:* 1:1,000	*Figure*
U.T.M. Zone UTM 9 - NAD83	*Mining District* LIARD	7b
N.T.S. 104G/4	*State/Province* BC	

BRITISH COLUMBIA

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Confirmation

Recorder: AWMACK, HENRY JAMES (101159) **Submitter:** AWMACK, HENRY JAMES (101159)

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Event Number: 4156985

Work Start Date: 2007/MAY/22
Work Stop Date: 2007/JUN/15

Total Value of Work: $ 500000.00
Mine Permit No: MX-1-732

Work Type: Technical Work
Technical Items: Drilling

Summary of the work value:

Tenure #	Claim Name/Property	Issue Date	Good To Date	New Good To Date	# of Days For-ward	Area in Ha	Work Value Due	Sub-mission Fee
509237		2005/mar/18	2007/jul/09	2016/dec/31	3463	613.83	$ 42399.61	$ 2329.52

Total required work value: $ 42399.61

PAC name: Henry Awmack

Debited PAC amount:	$	0.00
Credited PAC amount:	$	457600.39
Total Submission Fees:	$	2329.52
Total Paid:	$	2329.52

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