ASSESSMENT REPORT

on the

DILLARD PROPERTY

MTO Events # 5010962 + 5146768

SIMILKAMEEN MINING DIVISION,
British Columbia
Latitude 49°45' N, Longitude 120°25' W

Prepared for Operator:

FJORDLAND EXPLORATION INC.
1100 – 1111 Melville Street
Vancouver, B.C., Canada V6E 3V6

By:

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B.Sc., P .Geo.

9 December, 2011
Vancouver, B.C.

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1. SUMMARY

This report covers MTO Events 5010962 and 5146768 dated 19 September 2011 and 01 December 2011 respectively.

Between 11 September and 19 September 2011 a program consisting of prospecting and rock geochemistry was completed on the Dillard Property. The total cost of the survey was $30,150.[14].

The Dillard Property is located 50 kilometres west of Peachland and 47 kilometres southeast of Merritt in south-central British Columbia. At the date of this report, the Property consists of 17 mineral tenures with a total area of 2,192.2 hectares.

The western portion of the property is underlain by Upper Triassic-aged Nicola Group intermediate to mafic volcanic rocks, with local interbeds of argillite and minor limestone. Granite and quartz diorite of the Middle Jurassic Osprey Lake batholith intrude the Nicola volcanic rocks at the eastern portion of the property.

Prospecting and rock sampling has verified the distribution of copper and gold in the historic soil geochemistry programs. A total of 12 samples graded over 0.1% Cu and 2 rock samples graded 3.3 and 28.9 g/t Au.

The property has excellent potential to host economic deposits of porphyry-style mineralization.

A program of Induced Polarization and Magnetics over the anomalous zones is recommended over the two zones of mineralization to determine a mineralizing trend in preparation for drill testing. A geological, geochemical and geophysical database should be compiled from historic reporting prior to additional testing.

The cost of the next phase of exploration is estimated to be $80,000.

2.0 PROPERTY LOCATION, SIZE, ACCESS AND PHYSIOGRAPHY

The Dillard Property is located 50 kilometres west of Peachland and 47 kilometres southeast of Merritt in south-central British Columbia (Figure 1). The Property is located in the Similkameen Mining Division of south-central British Columbia, centred on latitude 49°45'N and longitude 120°25' W within NTS map areas 92H/9W and 16W (Figure 2).

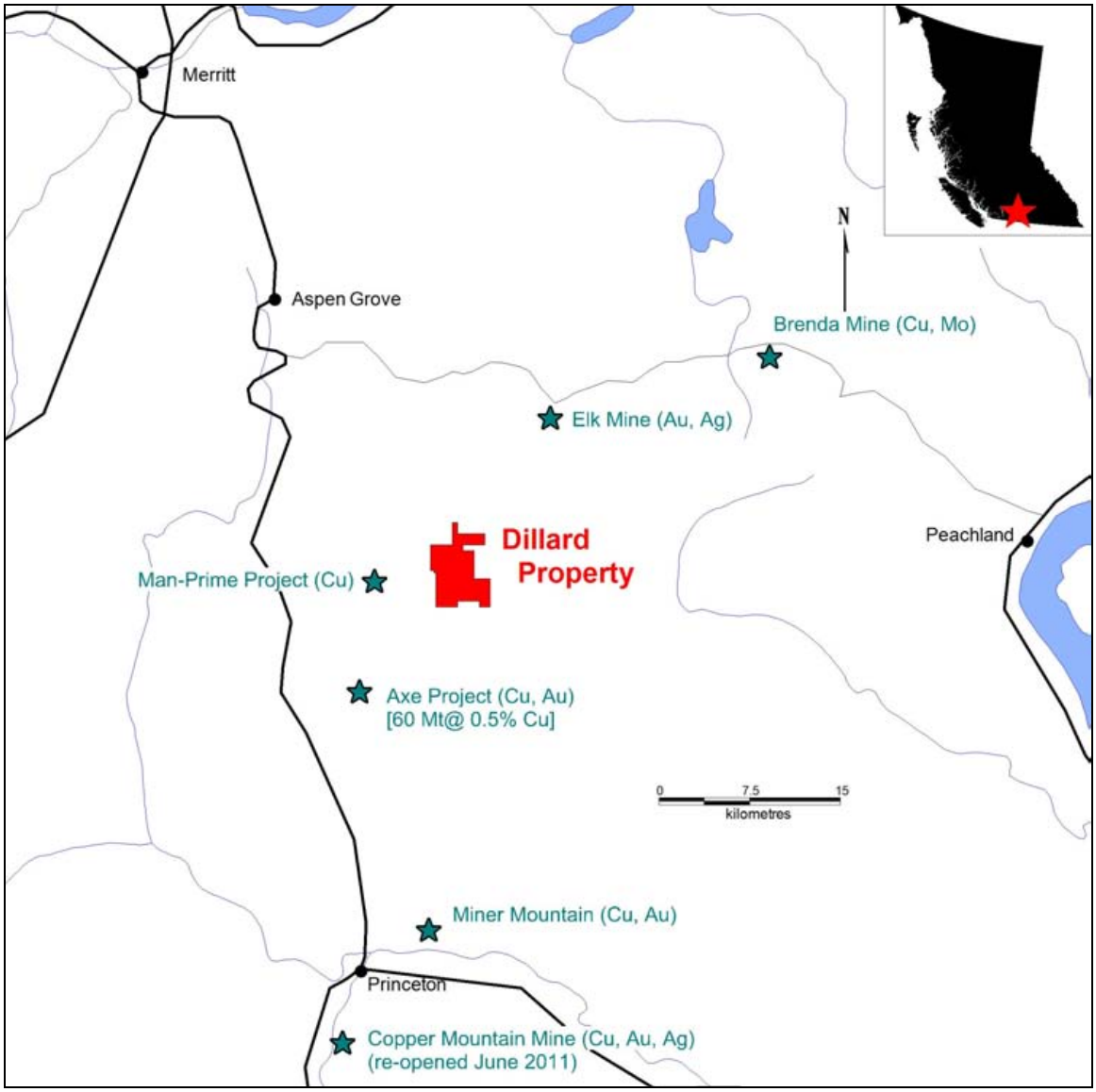


Figure 1: Location Map

Good gravel forest service roads provide access from Peachland and from the Princeton-Merritt highway (5A) via the Dillard Main Forest Service road, and from the Coquihalla highway (97C) via the Shrimpton Creek road south from the Loon Lake Road exit. Several logging roads traverse the property providing excellent access.

The claims cover an area of 2,179 hectares in rolling, hilly terrain on a broad uplands plateau. Elevations range from 1740 m asl in the center of the property to 1310 m in the southeast and 1430 m in the north. Small streams drain the property in all directions

from the high elevation point in the centre of the property. Outcrop exposures are moderately abundant and till cover overall is relatively shallow. Mature stands of spruce, balsam, fir and pine have been recently logged over at least 50% of the property. Annual temperatures range from -20° c to 30° C and precipitation is low to moderate. The area is basically snow-free from late June through October.

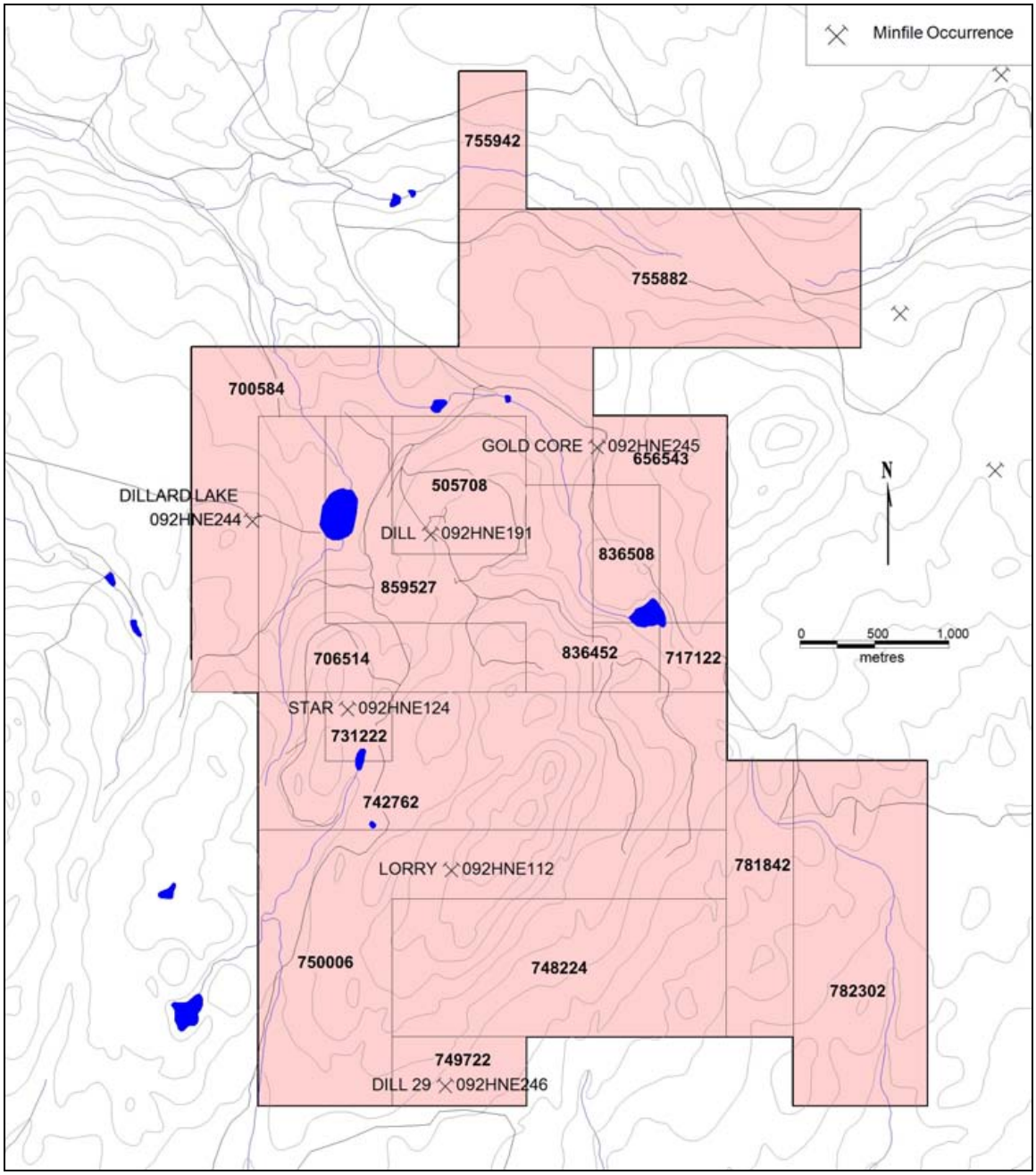


Figure 2: Claim Map

At the date of this report, the Dillard Property consists of 17 mineral tenures with a total area of 2,192.2 hectares. The property is surrounded on all sides by competitor claims. A listing of all tenures follows on Table 1. Expiry dates on tenures are valid pending the acceptance of this report.

Tenure	Issue Date	Good To	CLAIM_NAME	Area (ha)
505708	2005/02/03	2014/07/31		83
656543	2009/10/21	2014/07/31		83
700584	2010/01/16	2014/07/31		230
706514	2010/02/18	2014/07/31		146
717122	2010/03/06	2014/07/31	DRILL	21
731222	2010/03/20	2014/07/31		21
742762	2010/04/07	2014/07/31		271
748224	2010/04/14	2014/07/31		209
749722	2010/04/16	2014/07/31		42
750006	2010/04/16	2014/07/31		271
755882	2010/04/24	2014/07/31		250
755942	2010/04/24	2014/07/31		42
781842	2010/05/30	2014/07/31		84
782302	2010/05/31	2014/07/31		209
836452	2010/10/22	2014/07/31		21
836508	2010/10/23	2014/07/31		42
859527	2011/06/26	2014/07/31		167

Table 1: List of Tenures

The claims are owned by Mike Adam and Frank LaRoche of Kamloops B.C. and optioned to Fjordland Exploration Inc. Fjordland is a public company incorporated in Canada, with offices at #1100-1111 Melville Street, Vancouver, BC, Canada, V6E 3V6.

3.0 HISTORY

ARIS assessment filings report that the Dill Property and surrounding area were explored for copper, gold and silver from the early 1960's. The bulk of the property was explored for porphyry-style copper-gold mineralization in the 1970's and 1980's by Noranda Exploration, Cominco Ltd and Placer Dome (Fairfield Minerals ownership). Numerous surveys including soil geochemistry, and geophysical surveys including ground magnetics, VLF-EM and IP, and trenching were completed.

On the eastern portion of the property, at the contact with the L Triassic-aged Nicola volcanics and Jurassic-aged granitics, a small property was owned and operated by Harold Adams. From 1989 to 1999, 2 drill programs drilled 4 diamond drill holes (335.9 m) testing the vein-style gold potential similar to the nearby Siwash vein deposit of Almaden Minerals' Elk property.

A summary of exploration follows on Table 2.

ARIS #	Year	Company	Description
2356	1970	Pagent Mines	Photo-Geological Survey
4341	1972	Noranda Expl	Soil Geochemistry, Ground Mag
4491	1973	Noranda Expl	Soil Geochemistry
9429	1981	Cominco Ltd	Soil Geochemistry, Ground Mag + VLF-EM
11605	1983	Cominco Ltd	Soil Geochemistry
18410	1988	Fairfield Minerals	Soil Geochemistry
19335	1989	Harold Adams	2 BQ DDH (152.4m)
19593	1989	Fairfield Minerals/Placer Dome	Soil Geochemistry, IP, Trenching
25992	1999	Harold Adams	2 AQ DDH (183.5m)
29817	2008	John Kerr	Soil + Silt Geochemistry

Table 2: Historic Exploration Surveys

4.0 GEOLOGICAL SETTING

The Dill property is situated in the Intermontane tectonic belt in south-central British Columbia. The regional geology is shown on the northeastern part of GSC Map 41-1989, Hope (1:250,000), compiled by J.W.H.Monger (1984-89). The west side of the property is underlain by the Nicola Group, a package of intermediate to mafic volcanic rocks with minor limestone and sandstone intruded by consanguineous bodies of monzonite, diorite, granodiorite and dacite. These rocks have a genetic association with tectonic activity along the Summer's Creek and Allison fault systems which dominated the geology of the region in Late Triassic time (Preto, 1979). To the east, these rocks are in contact with granite and granodiorite of the Upper Jurassic osprey Lake batholith (Figure 3).

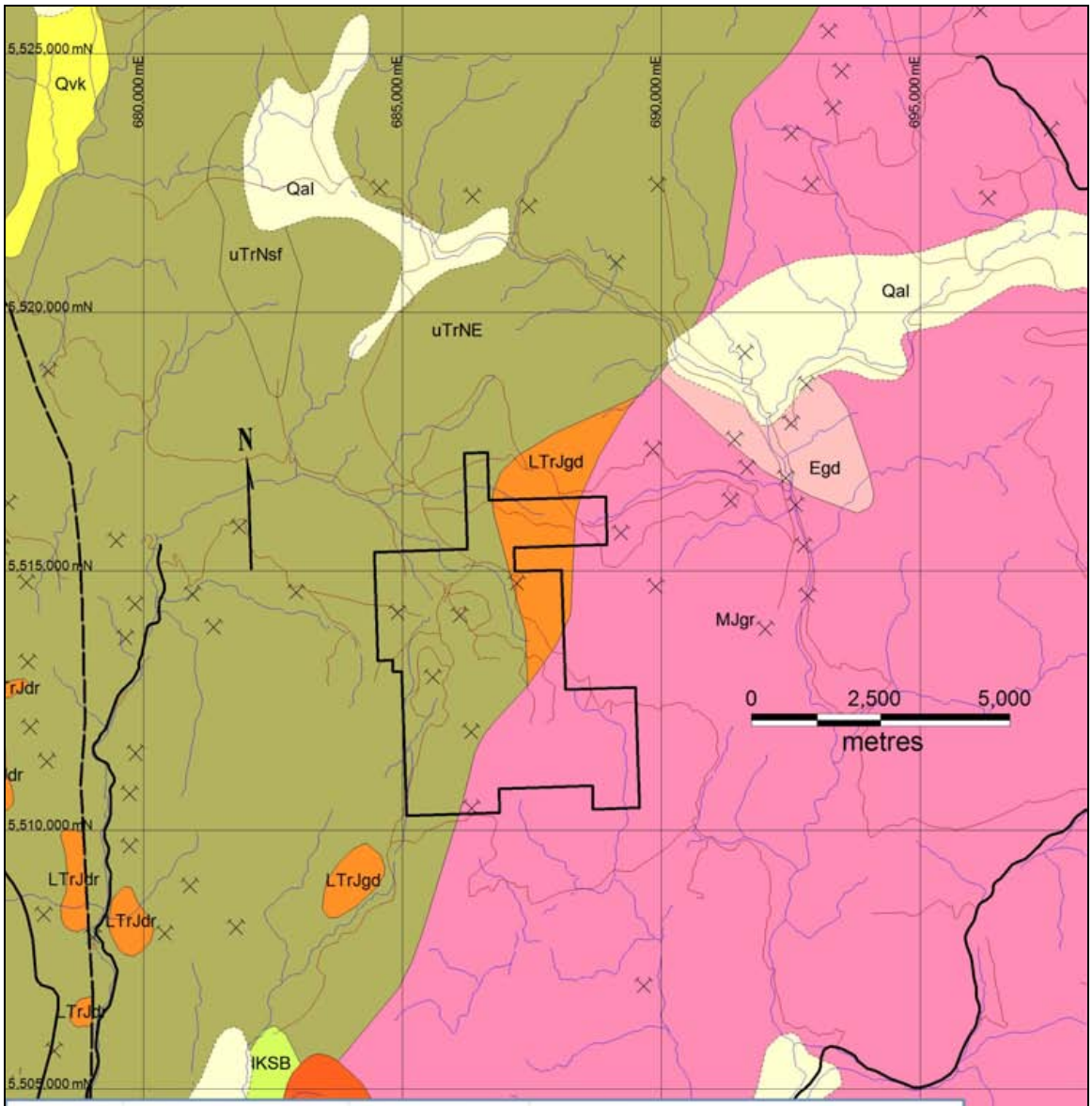
4.1 Property Geology

The Dillard property regional geology is shown on the northern part of GSC Map 88819, Princeton, mapped (H.M. A. Rice, 1939-1944).

The western portion of the property is underlain by Upper Triassic Nicola Group intermediate to mafic volcanic rocks, with local interbeds of argillite and minor limestone. These rocks occupy a shallow embayment in a Jurassic, reddish, coarse grained granite batholith which underlies the eastern part of the property .

Exposures on the property consist predominantly of dark green, blocky Nicola Group volcanic rocks. These are andesitic to basaltic flows and fragmental units which are contemporaneous with zones of diorite to granodiorite composition, considered to be sub-volcanic feeders.

Volcanics comprise predominantly light green to almost black, fine-grained andesite containing 0-70 percent augite/hornblende and plagioclase phenocrysts. Less common, volcanic breccia consists of sparse to concentrated clasts of granodiorite, monzonite, syenite and volcanics up to 15 cm, but typically 0.5 - 3 cm, with diffuse margins



STRAT_UNIT	STRAT_AGE	GROUP_SUIT	ROCK_TYPE
Qvk	Pleistocene to Holocene		alkaline volcanic rocks
Egd	Eocene		granodioritic intrusive rocks
Kgr	Cretaceous		granite, alkali feldspar granite intrusive rocks
IKSB	Lower Cretaceous	Spences Bridge Group	undivided volcanic rocks
MJgr	Middle Jurassic		granite, alkali feldspar granite intrusive rocks
LTrJgd	Late Triassic to Early Jurassic		granodioritic intrusive rocks
LTrJdr	Late Triassic to Early Jurassic		dioritic intrusive rocks
uTrNsf	Upper Triassic	Nicola Group	mudstone, siltstone, shale fine clastic sedimentary rocks
uTrNC	Upper Triassic	Nicola Group	andesitic volcanic rocks
uTrNE	Upper Triassic	Nicola Group	basaltic volcanic rocks

Minfile Occurrences

Figure 3: Dillard Geology

supported by a dark green to black fine-grained matrix with local augite/hornblende phenocrysts. Coarser grained andesites, trachyandesites and fine-grained tuffs are occasionally present.

The majority of intrusive rocks encountered to the west of the property are monzonites and monzodiorites with lesser diorites and dacites. Hornblende-plagioclase porphyry textures are common with phenocrysts typically measuring 1 - 3 mm. Groundmass textures range from aphanitic to moderately coarse (1 - 5 mm) interlocking crystals.

Granite and quartz diorite of the Middle Jurassic Osprey Lake batholith intrude the Nicola volcanic rocks at the eastern portion of the property. A number of fine grained siliceous dykes may represent a marginal feature of the Osprey Lake batholith.

Most Nicola rocks are massive, non-foliated and weakly metamorphosed, locally grading to greenschist facies. Local strong fracturing and shearing are accompanied in places by intense argillic, propylitic and carbonate alteration. Silicification, quartz and calcite veining and masses of epidote and garnet are developed locally.

Extensive hematite probably resulted from near-surface oxidation of magnetite and as a by-product of hydrothermal alteration at depth. Argillic alteration is concentrated in zones of faulting or strong fracturing. original textures are destroyed resulting in an orange to green incompetent aggregate of sericite, K-feldspar and clay minerals. Secondary quartz and wispy narrow veinlets of MnO(3) are common.

Potassic alteration has been observed in three different modes; pervasive salmon-pink coloured K-feldspar throughout the host rock (usually volcanic) typically less than a metre wide; thin orange to green selvages along carbonate-epidote stringers, commonly with diffuse outer margins; or K-feldspar filling microfractures and crackle breccia, visible in thin section and stained slab.

Silicification is typically confined to quartz + calcite, epidote veins and stringers, however, local increases in hardness of host rocks have been observed over narrow widths. Local albite flooding occurs proximal to carbonate and epidote stringers.

4.2 Mineralization

Two modes of mineralization are evident on the property.

Porphyry style mineralization is present as chalcopyrite in fracture coatings, fine veinlets, disseminations, and masses. Significant concentrations of chalcopyrite (up to 1% locally) is hosted by both intrusive and volcanic rocks. Pyrite occurs in veinlets ± carbonate, sericite, epidote, quartz) and as fine to medium-grained disseminations. Mineralization is associated with early carbonate-sericite-(quartz) filled breccia. Potassic alteration coincides with high chalcopyrite concentrations. Enriched gold values (up to 710 ppb over 3.1 m) correlate well with increased quartz + calcite, epidote veinlets. Significant samples taken during the 2011 program included one sample grading 1.6% Cu and 11 additional samples grading > 0.1% Cu from fractured, silicified volcanics.

Mineralization in the eastern portion of the property is typical of Mesothermal-type vein gold deposits similar in nature to the Siwash vein deposit of Almaden Minerals. This type of deposit is possibly indicated by gold-in-soils geochemistry in the eastern portion of the

claims. Historic drilling in the area delineated white carbonate altered zones and silicified zones with quartz-carbonate veinlets ± breccia containing 1-5% pyrite and traces of galena. Historic drilling from these zones did not contain anomalous gold or silver values, but several samples gave weak arsenic, copper, lead and zinc values. A rock sample from the 2011 prospecting program returned 3.3 g/t Au and 28.9 g/t Au from two areas sampling small quartz veins in diorite.

A total of 6 MINFILE occurrences are located on the property as shown on Table 3.

Minfile #	Name	Type	Commodity
092HNE191	DILL	Prospect	Au, Cu, Ag
092HNE246	DILL 29	Showing	Cu
092HNE244	DILLARD LAKE	Showing	Cu
092HNE245	GOLD CORE	Showing	Zn, Pb, Ag
092HNE112	LORRY	Showing	Cu
092HNE124	STAR	Showing	Cu

Table 3: Minfile Occurrences

5.0 2011 EXPLORATION PROGRAM

Property visits were made by the author and representatives from Fjordland on 3 August and 6 October 2011. Between 11 September and 19 September 2011 a program consisting of prospecting and rock geochemistry was completed on the Dillard property by Mike Adam and Frank LaRoche of Kamloops, BC.

A total of 88 rocks were collected by the author and prospectors. Prospective samples were chipped from outcroppings or from surface (float) debris believed to have originated nearby. Locations were determined in the field using GPS handheld units. A description including sample type, rock type, mineralization observed and significant analytical values was recorded for each sample (Appendix A).

No sample preparation was conducted by an employee, officer, director or associate of Fjordland prior to delivery to the laboratory for analyses. Samples were shipped by the prospectors to the offices of AGAT Laboratories located at 120 - 8600 Glenlyon Parkway, Burnaby, BC.

Samples were analyzed for a 51-element suite of elements. Sample analyses, preparation methods, and QAQC protocols are described in Appendix C. Analytical certificates are located in Appendix B.

The focus of prospected was aided by a compilation of historic soil sampling programs. Locations of samples including sample identification numbers are presented on Figure 4.

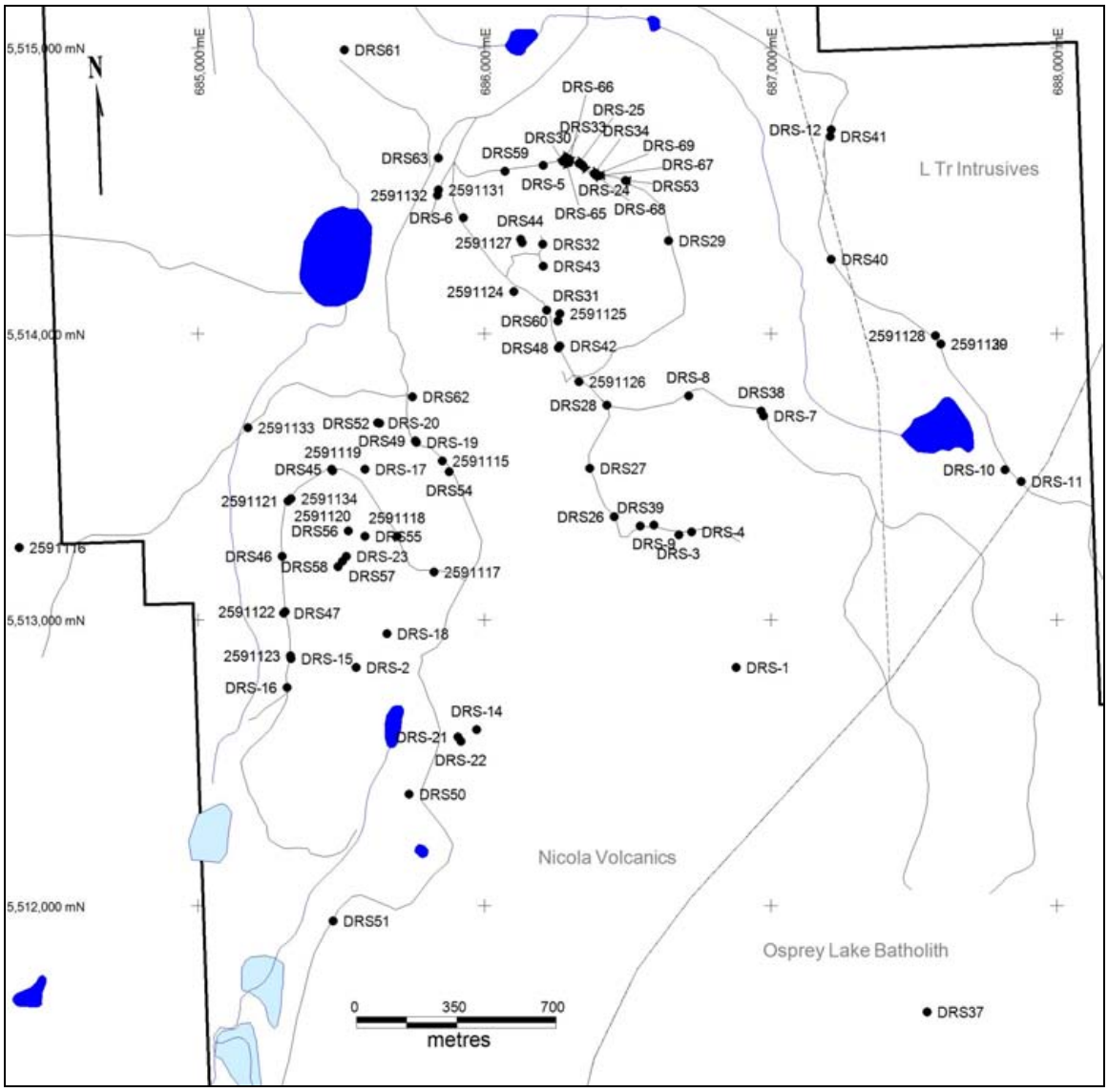


Figure 4: 2011 Rock Sample Location Map

All laboratory sample duplicates, blanks and standards were within expected tolerances and no indications of problems with the analyses were observed.

Results for copper and gold in rock sample are presented on Figures 5 and 6. Background contour values for copper and gold are also presented on the figures for comparative effect.

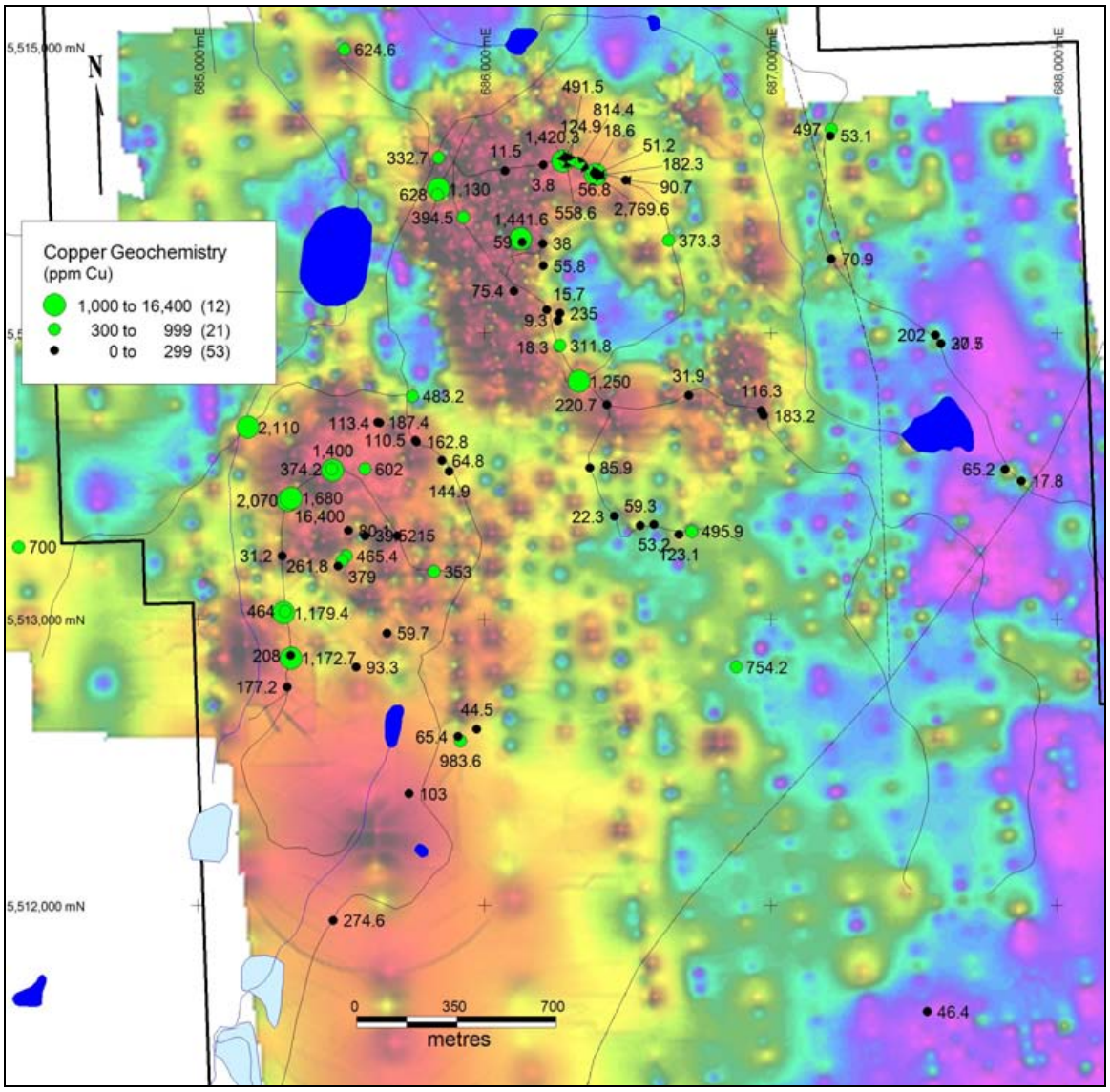


Figure 5: Copper Geochemistry Results

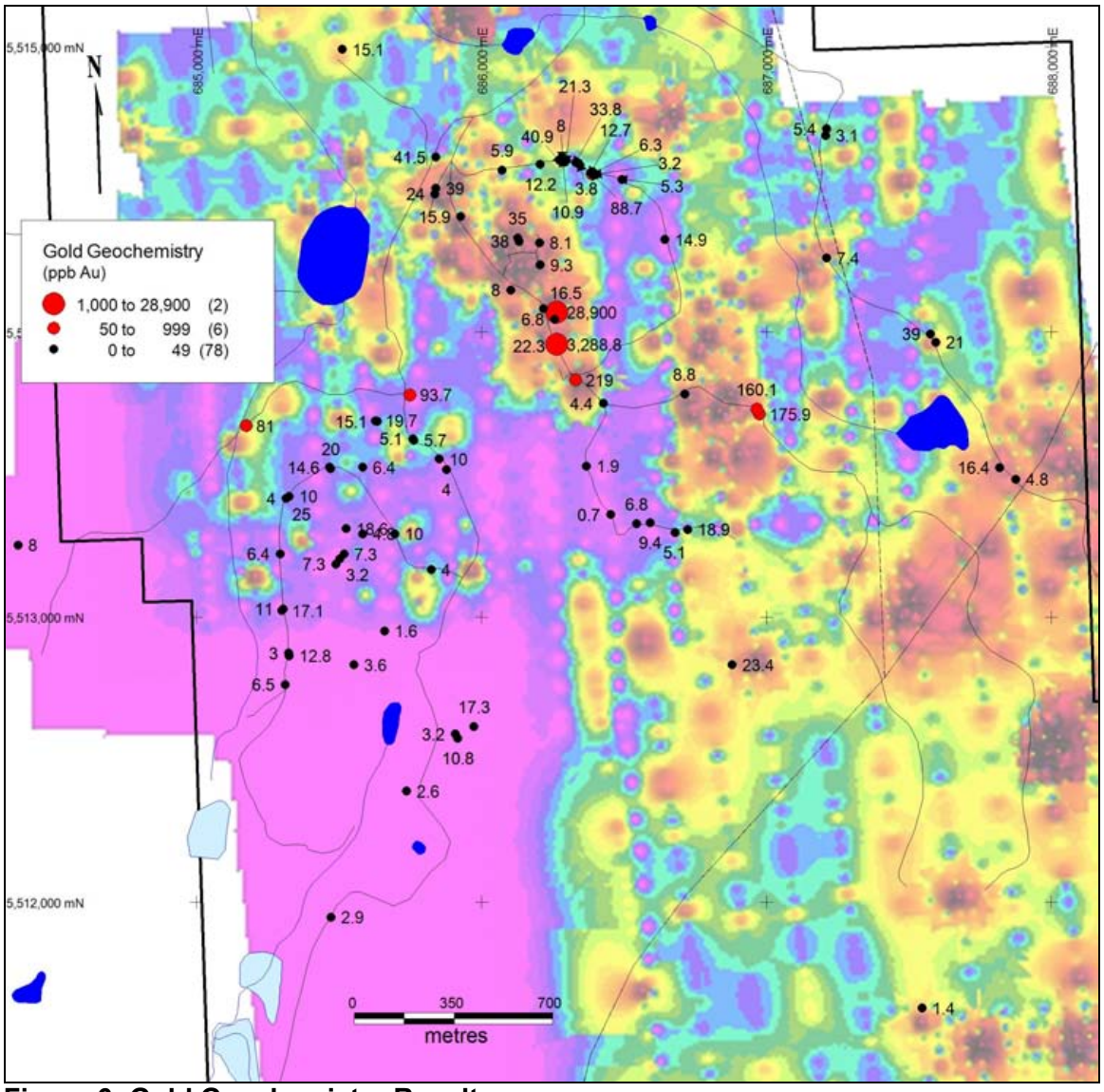


Figure 6: Gold Geochemistry Results

6.0 INTERPRETATION AND CONCLUSIONS

The distribution of copper and gold in the historic soil geochemistry programs are confirmed by the current prospecting and rock sampling program. A total of 12 samples graded over 0.1% Cu and 2 rock samples graded 3.3 and 28.9 g/t Au.

All three zones of copper mineralization, defined by historic copper-in-soil anomalies, have all had rock samples taken from them grading in excess of 0.1% copper. The gold-in-soils anomaly, coincident with the northernmost copper-in-soils anomaly and extending 800 m, was the focus of the historic trenching activities.

The property has excellent potential to host economic deposits of porphyry-style mineralization.

7.0 RECOMMENDATIONS

A program of Induced Polarization and Magnetics over the anomalous zones is recommended over the three zones of mineralization to determine a mineralizing trend in preparation for drill testing. A geological, geochemical and geophysical database should be compiled from historic reporting prior to additional testing.

The cost of the next phase of exploration is estimated to be $80,000.

8.0 STATEMENT OF EXPENDITURES

A total of 28 man-days worked.

Item	Description		Total
Mike Adam	Prospector	$	7,109.69
Frank LaRoche	Prospector	$	7,098.03
John Peters	Geologist	$	5,250.00
Tom Schroeter	Geologist	$	3,039.20
Truck Rental		$	589.48
Meals		$	122.72
Analytical		$	2,700.10
Report Writing		$	1,500.00
Subtotal		$	27,409.22
Office Overhead (10%)		$	2,740.92
Total		$	30,150.14
MTO Filing - Event 5010962	19-Sep-11	$	7,778.78
MTO Filing - Event 5146768	01-Dec-11	$	22,371.36

Table 4: Statement of Expenditures

9.0 REFERENCES

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Rowe, J.D. (1989): Geochemical Report On the DILL #1 Mineral Claim. Fairfield Minerals Ltd. Assessment Report 18410.

10.0 AUTHOR'S STATEMENT OF QUALIFICATIONS – L. John Peters

I, **L. John Peters, P.Geo** do hereby certify that:

a. I am a consulting geologist with addresses at 6549 Portland Street, Burnaby, BC, Canada, V5E 1A1.

b. I graduated with a Bachelor of Science degree (Geology) from the University of Western Ontario in 1984.

c. I am a Professional Geoscientist (P.Geo.) in good standing with the Association of Professional Engineers and Geoscientists of British Columbia (#19010).

d. I have worked as a geologist for a total of 26 years since my graduation from university.

e. I am responsible for the preparation of all sections of the technical report titled "Assessment Report on the Dillard Property" and dated 9 December 2011 relating to the Dillard Property. Work on the Dillard Property was completed by experienced prospectors under my supervision and I represent Fjordland as Exploration Manager.

f. I was not involved in any of the historic work programs on the Dillard Property, however, I have been involved in all aspects of Fjordland's exploration activities on the Property since 2011.

g. I am not aware of any material fact or material change with respect to the subject matter of the Technical Report that is not reflected in the Technical Report, the omission to disclose which makes the Technical Report misleading.

Dated this 9th day of December 2011.

"Lawrence John Peters"

Appendix A:
Rock Sample Descriptions

2011 Dillard Property Rock Sample Descriptions

ID	Easting	Northing	Sample Type	Rock Description	Au ppb	Cu	Mo	Pb	Zn	Ag
2591120	685314	5513416	Outcrop	silic volc + cpy in fract	25	16400	56.9	27.4	53.7	7.4
DRS-68	686389	5514554	Outcrop chip sample roadcut	Green silicified rock, chalcopyrite, pyrite, malachite,epidote	88.7	2769.6	0.3	3.2	16	1.8
2591133	685174	5513671	Outcrop	CPY + Mal in andesite	81	2110	12.1	6.8	28.5	1.7
2591121	685314	5513416	Outcrop	fg volc + cpy (50m east of 8)	4	2070	3.8	6.8	81	0.8
2591134	685324	5513424	Outcrop	re-sample Dill 8 from prev visit	10	1680	1.4	4.9	68.1	0.6
DRS44	686126	5514330	Subcrop on old trench	andesite - mal, chalco with small bleb of bornite from subcrop, on old trench.	35	1441.6	4.1	2	23	1
DRS30	686274	5514602	subcrop on new roadcut	abundant chalco,malachite,Py.epidote from andesite	40.9	1420.3	17.6	2.1	38	1
2591119	685470	5513522	Outcrop	fg volc w cpy/py in frac + magnetite veinlets	20	1400	<0.5	5	58.1	0.2
2591126	686330	5513832	Outcrop	Porph andes + cpy+mal	219	1250	40.2	3.6	66.2	2.3
DRS47	685298	5513023	Outcrop	andesite with malachite,chalco, tenerite, distintive tremolite-actinolite.	17.1	1179.4	<0.1	6.7	88	0.3
DRS-15	685325	5512865	Outcrop chip sample	Fine grained andesite, epidote, tremolite, tenorite, malachite	12.8	1172.7	0.4	13.9	107	1.1
2591131	685840	5514504	Outcrop	Intr tourmaline + cpy + tenarite (float?)	39	1130	7.2	3.5	23.2	0.4
DRS-22	685917	5512576	Subcrop rock sample roadcut	Dark andesite, pyrite, malachite, white potasic powder	10.8	983.6	0.1	295	71	0.8
DRS-25	686330	5514596	Subcrop rock sample	Blocky andesite, chalcopyrite, pyrite, epidote	33.8	814.4	0.5	3.7	30	0.6
DRS-1	686880	5512833	Surface outcrop chip rock sample	Green andesite, pyrite disseminated through rock iron staining.	23.4	754.2	0.2	4.5	41	0.6
2591116	684374	5513253	Outcrop	Hem volcanics w cpy/mal	8	700	0.8	4.1	56.4	<0.2
2591132	685835	5514485	Outcrop	Fine-gr volc + malachite - OC	24	628	0.9	4.7	47.8	<0.2
DRS61	685511	5514992	Subcrop	andesite - malachite staining.	15.1	624.6	0.1	1.1	53	1
DRS-17	685583	5513526	Outcrop chip sample	Green andesite, with black feldspar,malachite, minor chalcopyrite, pyrite	6.4	602	0.7	4.5	25	0.4
DRS-65	686286	5514599	Subcrop rock sample	Andesite blocky fractured, malachite, chalcopyrite, tenorite, epidote, lightly rusted	10.9	558.6	0.3	2.2	73	0.2
DRS-12	687211	5514713	Subcrop rock sample	Silicified quartz, pyrite, epidote, heavy rust staining	5.4	497	0.3	1249.3	2526	3.9
DRS-4	686723	5513308	Subcrop rock sample roadcut	Andesite, silicified volcanics with quartz, massive pyrite.	18.9	495.9	0.3	2.5	16	0.5
DRS-66	686297	5514602	Subcrop rock sample roadcut	Andesite, blocky fractured, silicified quartz	21.3	491.5	0.4	2	28	0.4
DRS62	685748	5513780	Subcrop	malachite, chalco, some quartz from a 1 cm quartz vein in diorite.	93.7	483.2	0.3	4.7	34	0.9
DRS-23	685517	5513221	Outcrop chip sample	Andesite, tremolite, chalcopyrite, pyrite	7.3	465.4	0.3	13.5	115	0.7
2591122	685303	5513029	Outcrop	fg volc + ep +frac cpy, magnetic + tremolite	11	464	0.5	5.4	112	<0.2
DRS-6	685926	5514405	Subcrop rock sample roadcut	Fine grained andesite,epidote,pyrite disseminated through rock iron staining.	15.9	394.5	0.8	0.8	55	0.3
DRS57	685503	5513204		andesite - malachite,chalco,tenorite, tremolite.	3.2	379	0.7	17.6	98	0.5
DRS45	685466	5513526	Outcrop	quartz vein in andesite - malachite, chalco,with specular hematite.	14.6	374.2	1	2.3	43	0.1
DRS29	686643	5514325	angular float on new roadcut.	rusty with abundant Py, malachite,azurite	14.9	373.3	0.4	1	78	0.5
2591117	685824	5513167	Outcrop	Andesite volc w cpy	4	353	<0.5	3.9	65.5	<0.2
DRS63	685839	5514614	Outcrop	andesite - very rusty with a small bleb of chalco, Py.	41.5	332.7	0.5	0.7	95	0.5
DRS42	686263	5513957		small one cm. quartz vein in (diorite?).	3288.8	311.8	2.2	4.8	12	1.9
DRS51	685472	5511948	Outcrop	malachite with minor chalco in andesite	2.9	274.6	0.2	6.7	84	0.6
DRS58	685488	5513186		andesite with malachite, tenorite,tremolite.	7.3	261.8	0.2	12.2	83	0.1
2591125	686264	5514070	Outcrop	cpy in qtz vein	28900	235	3	14.9	39.6	11.4
DRS28	686427	5513750	Subcrop	rusty with some Py, from andesite	4.4	220.7	1	0.9	26	0.1
2591118	685695	5513292	Outcrop	diss cpy in fg intrusive	10	215	1	8.8	92.6	0.8
2591123	685323	5512874	Outcrop	volc + cpy-mal-tenarite in fracts	3	208	0.9	7.7	104	<0.2
2591128	687575	5513994	Outcrop	Volcanics, py-rich	39	202	0.6	4	27.9	<0.2
DRS-20	685627	5513689	Subcrop rock sample	Andesite, pyrite, light malachite staining	19.7	187.4	5.3	4	28	0.3
DRS-7	686975	5513712	Outcrop chip sample roadcut	Massive sulphide in silicified quartz, heavy iron staining	175.9	183.2	12.8	8.2	7	3.8
DRS-67	686402	5514553	Outcrop chip sample roadcut	Rusty silicified zone, white potassic powder quartz	3.2	182.3	0.8	0.7	32	<0.1
DRS-16	685310	5512764	Rust clay zone grab sample	Rusty carbonate, silicified quartz,	6.5	177.2	0.4	8.4	96	0.8
DRS-19	685762	5513621	Subcrop rock sample	Andesite, heavy rust, highly fractured silicified	5.7	162.8	0.4	2.2	69	0.3
DRS54	685877	5513518	Outcrop	4 cm. quartz vein srike 40 degrees, dip 40 degrees N.E. in andesite, py, rusty in andesite	4	144.9	<0.1	1.5	42	0.2
DRS33	686283	5514614	Float	very rusty,Py, silicified, source looks to be close by.	8	124.9	0.3	2.1	44	0.2
DRS-3	686679	5513297	Subcrop rock sample	Fine grained andesite,epidote,pyrite disseminated through rock iron staining.	5.1	123.1	0.4	5.3	47	0.2
DRS38	686967	5513730	Float boulder	angular andesite float boulder with quartz, Py, source looks to be close by.	160.1	116.3	0.5	162.5	164	19.9

ID	Easting	Northing	Sample Type	Rock Description	Au ppb	Cu	Mo	Pb	Zn	Ag
DRS52	685634	5513687	Outcrop	up to 15 cm. quartz vein srike 55 degrees, dip 45 degrees N.E. in andesite, abundant Py, malachite, minor amount of chalco, tremolite visable on wall rock.	15.1	113.4	1.6	7.9	12	0.3
DRS49	685758	5513626		rusty silicified, Py in andesite.	5.1	110.5	1.4	1.4	25	0.1
DRS50	685736	5512390		minor rusty andesite.	2.6	103	<0.1	2	64	0.1
DRS-2	685551	5512833	Surface outcrop chip rock sample	Fine grained andesite, pyrite disseminated through rock iron staining.	3.6	93.3	0.2	2.6	70	0.1
DRS53	686493	5514536	Outcrop	Rock sample, rusty,Py, in diorite	5.3	90.7	0.2	1	25	<0.1
DRS27	686367	5513530	Outcrop	very rusty , PY, from andesite	1.9	85.9	1.2	1.2	29	<0.1
DRS56	685525	5513310	Subcrop	andesite - small amount of malachite and chalco, visable tremolite, rusty with minor Py.	18.6	80.1	0.5	5.6	49	0.1
2591124	686103	5514148	Outcrop	quartz stockwork	8	75.4	0.6	1.6	80	<0.2
DRS40	687211	5514260		dark green andesite with what looks some argillite	7.4	70.9	0.1	2.9	112	0.2
DRS-21	685907	5512590	Subcrop rock sample	Dark andesite, pyrite, light rust, epidote	3.2	65.4	0.1	1.1	54	0.1
DRS-10	687818	5513525	Quartz vein chip sample	Quartz,heavy rust staining, minor pyrite	16.4	65.2	2.3	0.9	3	0.2
2591115	685852	5513555	Outcrop	Andesite volc w cpy	10	64.8	4.1	6.6	51.8	<0.2
DRS-18	685660	5512952	Subcrop rock sample	Diorite, pyrite, light rust staining, silicified quartz	1.6	59.7	0.5	22.3	97	0.2
DRS39	686544	5513328	Outcrop	very rusty,black sooty from outcrop, rock type unknown.	6.8	59.3	0.3	699.8	>10000	0.2
2591127	686131	5514319	Outcrop	Porph andes + cpy+bornite+py	38	59	1.1	2	36.3	0.2
DRS-24	686345	5514586	Subcrop rock sample	Silicified rusty zone, quartz rich pyrite, black carbonate	3.8	56.8	0.1	1.6	41	<0.1
DRS43	686205	5514236	Rubble pile from old trench	andesite with quartz attached from old trench rubble pile.	9.3	55.8	0.1	1.5	15	<0.1
DRS-9	686592	5513332	Subcrop rock sample	Andesite,epidote,light pyrite, rust staining	9.4	53.2	0.9	1.6	52	<0.1
DRS41	687207	5514690	Outcrop	Rusty looking, minor Py,andesite	3.1	53.1	0.1	3	125	0.1
DRS-69	686389	5514554	Outcrop chip sample roadcut	Green silicified rock, quartz, pyrite, light rust	6.3	51.2	1	0.6	17	<0.1
DRS37	687547	5511629	Subcrop on log landing	rusty coarse grained granite	1.4	46.4	1.5	114.6	109	0.2
DRS-14	685973	5512617	Quartz vein chip sample	One foot wide quartz vein, heavy rust, pyrite.	17.3	44.5	2.1	3.3	15	0.7
DRS55	685582	5513292	Outcrop	minor Py, tremolite, possibly malachite, rock has distintive green hue, in andesite	4.8	39.5	0.5	9.2	59	0.2
DRS32	686204	5514314	Subcrop	small quartz vein in diorite	8.1	38	0.2	1.1	19	<0.1
2591129	687594	5513965	Outcrop	Qtz-carb vein + py, cpy	21	37.7	<0.5	1.2	64.6	<0.2
DRS-8	686713	5513783	Quartz vein chip sample	Quartz, chalcopyrite,pyrite, light rust staining	8.8	31.9	2.2	2	2	<0.1
DRS46	685293	5513222	Subcrop	quartz with iron carbonate from subcrop, numerous rusty boulders nearby in road ditch.	6.4	31.2	0.3	15.8	124	0.1
DRS26	686454	5513360	angular float boulder	numerous quartz veinlets throughout with Py, possibly some chalco.	0.7	22.3	0.3	1.3	46	<0.1
259113C	687594	5513965	Outcrop	Silicified volcanics + sulphide	21	20.5	<0.5	4.5	34.8	<0.2
DRS34	686384	5514560	taken over 6 meters from similar float rocks	very rusty , PY, silicified, small bleb of chalco	12.7	18.6	2.4	12.1	95	0.6
DRS48	686259	5513951	Outcrop	4 cm. quartz vein in diorite, py	22.3	18.3	0.9	1.3	12	<0.1
DRS-11	687875	5513484	Subcrop rock sample	Granodiorite with quartz veins	4.8	17.8	0.8	2.9	14	<0.1
DRS31	686217	5514083	Boulder float	Py,chalco,quartz, andesite	16.5	15.7	0.1	0.8	6	<0.1
DRS59	686072	5514568		diorite with small quartz vein, rusty,Py.	5.9	11.5	2.8	3.4	20	<0.1
DRS60	686256	5514045	Float	rusty 15 cm. wide quartz vein	6.8	9.3	1.6	0.5	12	<0.1
DRS-5	686206	5514589	Float	Rough edged quartz boulder, iron staining	12.2	3.8	0.1	0.2	1	<0.1
DRS-13	686663	5516924	Float	Coarse grain granite with pink feldspar, quartz veining	1.8	2.9	8.7	4.4	28	<0.1
DRS35	687918	5512500	Subcrop	coarse grained granite with some quartz taken over 6 meters	2.5	2.4	0.5	8.6	34	<0.1
DRS36	688040	5512549	Outcrop	coarse grained red color granite	1.1	1.7	0.5	5.4	12	<0.1

Appendix B:
Laboratory Certificates

Acme Analytical Laboratories (Vancouver) Ltd.
1020 Cordova St. East Vancouver BC V6A 4A3 Canada

www.acmelab.com

Client: **Fjordland Exploration Inc.**
11th Floor, 1111 Melville Street
Vancouver BC V6E 3V6 Canada

Submitted By: John Peters
Receiving Lab: Canada-Vancouver
Received: October 12, 2011
Report Date: October 29, 2011
Page: 1 of 3

CERTIFICATE OF ANALYSIS VAN11005474.1

CLIENT JOB INFORMATION

Project: Dillard
Shipment ID:
P.O. Number
Number of Samples: 38

SAMPLE PREPARATION AND ANALYTICAL PROCEDURES

Method Code	Number of Samples	Code Description	Test Wgt (g)	Report Status	Lab
R200-250	38	Crush, split and pulverize 250 g rock to 200 mesh			VAN
1DX2	38	1:1:1 Aqua Regia digestion ICP-MS analysis	15	Completed	VAN

SAMPLE DISPOSAL

STOR-PLP Store After 90 days Invoice for Storage
STOR-RJT Store After 90 days Invoice for Storage

Acme does not accept responsibility for samples left at the laboratory after 90 days without prior written instructions for sample storage or return.

ADDITIONAL COMMENTS

Invoice To: Fjordland Exploration Inc.
11th Floor, 1111 Melville Street
Vancouver BC V6E 3V6
Canada

CC:

BRITISH COLUMBIA CERTIFIED ASSAYER
CLARENCE LEONG
GENERAL MANAGER

This report supersedes all previous preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates final approval; preliminary reports are unsigned and should be used for reference only.
All results are considered the confidential property of the client. Acme assumes the liabilities for actual cost of analysis only. Results apply to samples as submitted.
"*" asterisk indicates that an analytical result could not be provided due to unusually high levels of interference from other elements.

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Client: **Fjordland Exploration Inc.**
11th Floor, 1111 Melville Street
Vancouver BC V6E 3V6 Canada

Project: Dillard
Report Date: October 29, 2011

CERTIFICATE OF ANALYSIS VAN11005474.1

Method		WGHT	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
Analyte		Wgt	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P
Unit		kg	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppb	ppm	ppm	ppm	ppm	ppm	ppm	%	%
MDL		0.01	0.1	0.1	0.1	1	0.1	0.1	0.1	1	0.01	0.5	0.5	0.1	1	0.1	0.1	0.1	2	0.01	0.001
DRS26	Rock	1.94	0.3	22.3	1.3	46	<0.1	44.7	46.9	2210	4.72	2.5	0.7	0.1	116	<0.1	0.2	0.4	71	6.71	0.090
DRS27	Rock	2.13	1.2	85.9	1.2	29	<0.1	11.6	20.8	324	1.89	7.7	1.9	0.6	14	<0.1	0.4	0.2	71	0.66	0.087
DRS28	Rock	2.25	1.0	220.7	0.9	26	0.1	89.1	16.4	382	2.15	1.6	4.4	0.1	47	<0.1	<0.1	<0.1	70	1.42	0.093
DRS29	Rock	1.88	0.4	373.3	1.0	78	0.5	48.2	34.4	786	4.01	2.7	14.9	0.2	41	<0.1	<0.1	<0.1	121	1.06	0.113
DRS30	Rock	2.15	17.6	1420	2.1	38	1.0	21.9	3.8	1879	3.33	9.7	40.9	1.0	80	0.2	0.5	<0.1	101	6.60	0.164
DRS31	Rock	1.91	0.1	15.7	0.8	6	<0.1	3.5	3.2	109	0.89	0.7	16.5	0.2	17	<0.1	<0.1	<0.1	44	0.54	0.062
DRS32	Rock	1.35	0.2	38.0	1.1	19	<0.1	7.8	8.1	298	2.08	1.1	8.1	0.4	42	<0.1	0.1	0.1	84	0.63	0.097
DRS33	Rock	1.55	0.3	124.9	2.1	44	0.2	58.8	26.7	1025	3.66	59.0	8.0	0.2	58	0.3	8.0	<0.1	113	3.08	0.031
DRS34	Rock	1.71	2.4	18.6	12.1	95	0.6	29.8	41.4	1484	4.76	12.0	12.7	0.3	29	0.4	0.6	1.6	24	2.03	0.072
DRS35	Rock	1.94	0.5	2.4	8.6	34	<0.1	0.8	1.2	235	0.57	<0.5	2.5	12.5	7	0.2	0.2	<0.1	6	0.05	0.012
DRS36	Rock	1.49	0.5	1.7	5.4	12	<0.1	0.8	1.7	238	0.69	<0.5	1.1	7.3	9	<0.1	0.2	<0.1	9	0.08	0.025
DRS37	Rock	1.75	1.5	46.4	114.6	109	0.2	1.7	3.6	526	1.49	5.2	1.4	13.9	10	0.2	0.8	0.1	14	0.09	0.029
DRS38	Rock	2.22	0.5	116.3	162.5	164	19.9	12.5	9.9	2756	2.79	63.8	160.1	0.2	167	1.5	8.4	9.8	29	13.96	0.038
DRS39	Rock	1.44	0.3	59.3	699.8	>10000	0.2	55.3	35.2	9836	11.49	5.7	6.8	0.4	37	21.0	2.4	0.1	70	2.65	0.091
DRS40	Rock	1.89	0.1	70.9	2.9	112	0.2	62.2	21.4	685	3.71	5.0	7.4	0.3	32	<0.1	0.2	<0.1	123	1.19	0.119
DRS41	Rock	2.27	0.1	53.1	3.0	125	0.1	26.9	26.5	754	4.90	3.1	3.1	0.2	54	<0.1	0.1	<0.1	165	1.38	0.116
DRS42	Rock	1.62	2.2	311.8	4.8	12	1.9	13.2	25.7	85	6.46	9.9	3289	0.1	18	0.1	0.2	1.5	30	0.33	0.041
DRS43	Rock	2.16	0.1	55.8	1.5	15	<0.1	6.5	5.2	197	1.42	1.3	9.3	0.6	25	<0.1	0.1	<0.1	57	0.56	0.061
DRS44	Rock	1.49	4.1	1442	2.0	23	1.0	17.5	19.0	403	2.35	3.2	35.0	0.3	54	0.2	0.6	0.3	51	1.29	0.147
DRS45	Rock	1.86	1.0	374.2	2.3	43	0.1	3.2	12.4	837	3.29	1.8	14.6	1.0	29	0.3	0.2	<0.1	102	0.66	0.112
DRS46	Rock	1.61	0.3	31.2	15.8	124	0.1	4.5	30.3	2437	4.29	6.0	6.4	1.5	87	1.2	0.2	0.2	108	7.17	0.129
DRS47	Rock	1.93	<0.1	1179	6.7	88	0.3	7.6	21.0	1071	3.19	2.2	17.1	1.2	60	<0.1	0.3	<0.1	112	1.11	0.143
DRS48	Rock	1.62	0.9	18.3	1.3	12	<0.1	7.8	15.3	160	2.99	0.9	22.3	0.3	50	<0.1	<0.1	0.3	80	0.86	0.083
DRS49	Rock	1.54	1.4	110.5	1.4	25	0.1	6.4	12.3	289	2.42	1.6	5.1	0.2	56	<0.1	<0.1	0.1	76	0.67	0.094
DRS50	Rock	1.80	<0.1	103.0	2.0	64	0.1	15.8	21.9	608	3.33	0.8	2.6	0.5	100	<0.1	0.2	<0.1	103	0.97	0.124
DRS51	Rock	2.00	0.2	274.6	6.7	84	0.6	10.7	28.6	550	2.96	3.6	2.9	0.7	67	0.4	0.3	<0.1	98	0.96	0.156
DRS52	Rock	2.17	1.6	113.4	7.9	12	0.3	3.2	12.2	114	2.92	1.3	15.1	0.5	38	<0.1	0.2	0.2	52	0.32	0.079
DRS53	Rock	1.81	0.2	90.7	1.0	25	<0.1	24.0	14.6	319	2.64	3.2	5.3	0.3	84	<0.1	0.3	<0.1	94	0.94	0.102
DRS54	Rock	1.73	<0.1	144.9	1.5	42	0.2	9.5	16.4	1407	2.45	1.3	4.0	0.3	146	0.1	<0.1	<0.1	56	11.58	0.082
DRS55	Rock	1.78	0.5	39.5	9.2	59	0.2	9.6	12.1	430	2.98	3.7	4.8	0.8	82	0.1	0.5	0.3	69	0.87	0.131

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Project: Dillard
Report Date: October 29, 2011

CERTIFICATE OF ANALYSIS VAN11005474.1

Method		1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
Analyte		La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Hg	Sc	Tl	S	Ga	Se	Te
Unit		ppm	ppm	%	ppm	%	ppm	%	%	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm
MDL		1	1	0.01	1	0.001	1	0.01	0.001	0.01	0.1	0.01	0.1	0.1	0.05	1	0.5	0.2
DRS26	Rock	2	58	2.67	48	0.027	2	1.09	0.033	0.30	<0.1	0.01	16.6	<0.1	0.45	3	<0.5	<0.2
DRS27	Rock	2	17	0.60	58	0.091	<1	0.94	0.124	0.31	0.3	<0.01	3.8	0.3	0.13	4	<0.5	<0.2
DRS28	Rock	2	212	1.29	107	0.160	<1	1.24	0.096	0.60	<0.1	<0.01	4.6	0.1	0.05	3	0.5	<0.2
DRS29	Rock	2	89	2.00	166	0.245	<1	2.17	0.144	1.52	<0.1	<0.01	4.3	0.2	0.46	6	0.8	<0.2
DRS30	Rock	7	37	0.30	69	0.086	<1	1.09	<0.001	0.02	0.4	<0.01	5.3	<0.1	0.08	4	1.2	<0.2
DRS31	Rock	<1	6	0.28	49	0.021	2	0.93	0.084	0.25	1.1	0.06	1.5	<0.1	0.43	2	<0.5	<0.2
DRS32	Rock	2	13	0.88	77	0.124	5	1.36	0.080	0.57	0.3	0.15	2.1	0.1	0.11	5	<0.5	<0.2
DRS33	Rock	2	52	1.20	122	0.004	5	0.47	0.007	0.13	0.2	0.66	16.5	0.2	0.21	1	<0.5	<0.2
DRS34	Rock	4	6	0.47	40	0.002	3	0.44	0.014	0.25	<0.1	<0.01	7.3	<0.1	2.73	<1	1.5	0.2
DRS35	Rock	15	3	0.02	28	0.001	2	0.17	0.029	0.10	0.2	<0.01	0.3	<0.1	<0.05	<1	<0.5	<0.2
DRS36	Rock	11	2	0.02	34	0.001	2	0.19	0.034	0.11	0.1	<0.01	0.8	<0.1	<0.05	<1	<0.5	<0.2
DRS37	Rock	21	3	0.04	112	<0.001	3	0.29	0.021	0.14	0.1	<0.01	1.2	0.1	<0.05	<1	0.6	<0.2
DRS38	Rock	6	16	0.85	19	0.001	4	0.85	0.002	0.10	<0.1	0.02	4.1	<0.1	0.41	2	<0.5	7.5
DRS39	Rock	8	9	1.56	184	0.002	4	0.33	0.003	0.19	0.3	0.23	21.4	0.1	<0.05	1	0.5	<0.2
DRS40	Rock	2	68	1.93	283	0.141	2	1.80	0.112	0.55	0.2	0.01	5.9	0.2	<0.05	6	<0.5	<0.2
DRS41	Rock	2	37	1.85	292	0.148	<1	1.74	0.084	1.39	<0.1	<0.01	5.6	0.3	<0.05	6	<0.5	<0.2
DRS42	Rock	<1	6	0.15	9	0.039	<1	0.48	0.033	0.09	6.7	0.15	1.6	<0.1	6.14	2	5.2	3.5
DRS43	Rock	1	22	0.53	60	0.076	2	0.78	0.057	0.18	0.4	0.02	1.9	<0.1	0.21	2	<0.5	<0.2
DRS44	Rock	2	86	0.68	25	0.107	<1	0.80	0.028	0.09	0.2	<0.01	2.6	<0.1	0.43	3	0.8	<0.2
DRS45	Rock	8	2	0.87	30	0.052	<1	0.99	0.032	0.25	<0.1	0.02	2.4	<0.1	0.06	5	<0.5	<0.2
DRS46	Rock	10	3	2.31	103	0.024	5	0.75	0.029	0.27	<0.1	<0.01	4.7	<0.1	0.39	3	1.0	<0.2
DRS47	Rock	6	12	1.63	53	0.136	3	1.73	0.045	0.52	0.1	<0.01	3.3	<0.1	0.06	7	1.4	<0.2
DRS48	Rock	2	9	0.56	64	0.094	2	1.48	0.143	0.31	8.9	0.02	2.2	<0.1	1.37	4	0.7	0.2
DRS49	Rock	2	8	0.72	81	0.131	2	0.97	0.114	0.22	0.2	0.07	2.3	<0.1	0.45	4	1.0	<0.2
DRS50	Rock	4	16	1.72	311	0.165	3	1.79	0.053	1.03	<0.1	<0.01	2.1	<0.1	<0.05	6	0.6	<0.2
DRS51	Rock	5	8	1.61	129	0.169	4	1.86	0.068	1.02	0.1	<0.01	2.2	0.3	0.10	5	0.6	<0.2
DRS52	Rock	3	7	0.29	26	0.092	1	0.41	0.047	0.10	0.4	<0.01	1.1	<0.1	0.73	2	1.1	<0.2
DRS53	Rock	3	39	1.14	68	0.164	2	1.53	0.081	0.47	0.1	<0.01	2.6	<0.1	0.14	4	0.7	<0.2
DRS54	Rock	3	12	1.14	143	0.111	3	1.10	0.031	0.60	0.1	0.02	2.4	<0.1	0.43	4	0.9	<0.2
DRS55	Rock	5	14	0.91	68	0.134	1	1.20	0.055	0.10	0.3	<0.01	1.5	<0.1	0.13	5	0.7	<0.2

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11th Floor, 1111 Melville Street
Vancouver BC V6E 3V6 Canada

Project: Dillard
Report Date: October 29, 2011

CERTIFICATE OF ANALYSIS VAN11005474.1

Method		WGHT	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
Analyte		Wgt	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P
Unit		kg	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppb	ppm	ppm	ppm	ppm	ppm	ppm	%	%
MDL		0.01	0.1	0.1	0.1	1	0.1	0.1	0.1	1	0.01	0.5	0.5	0.1	1	0.1	0.1	0.1	2	0.01	0.001
DRS56	Rock	1.58	0.5	80.1	5.6	49	0.1	13.1	11.1	495	2.56	1.7	18.6	0.1	61	0.1	0.2	<0.1	89	0.86	0.094
DRS57	Rock	1.41	0.7	379.0	17.6	98	0.5	8.2	8.9	964	3.94	2.7	3.2	0.7	86	0.3	0.2	<0.1	88	1.36	0.154
DRS58	Rock	1.49	0.2	261.8	12.2	83	0.1	9.3	12.1	683	3.95	1.9	7.3	0.9	59	0.4	0.2	<0.1	106	0.91	0.140
DRS59	Rock	1.77	2.8	11.5	3.4	20	<0.1	9.3	14.0	243	2.21	1.8	5.9	0.3	61	<0.1	0.3	0.1	65	0.84	0.087
DRS60	Rock	1.34	1.6	9.3	0.5	12	<0.1	6.0	10.4	311	1.68	1.1	6.8	<0.1	23	<0.1	0.1	<0.1	8	1.58	0.005
DRS61	Rock	1.40	0.1	624.6	1.1	53	1.0	7.9	16.0	729	2.77	0.8	15.1	0.5	107	0.2	<0.1	<0.1	88	1.41	0.120
DRS62	Rock	1.48	0.3	483.2	4.7	34	0.9	7.6	31.2	575	2.86	9.9	93.7	0.7	50	0.1	0.3	0.4	72	1.34	0.129
DRS63	Rock	1.63	0.5	332.7	0.7	95	0.5	15.7	25.3	1239	4.82	1.9	41.5	0.4	35	<0.1	0.2	<0.1	174	0.73	0.116

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Project: Dillard
Report Date: October 29, 2011

CERTIFICATE OF ANALYSIS VAN11005474.1

	Method	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
	Analyte	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Hg	Sc	Tl	S	Ga	Se	Te
	Unit	ppm	ppm	%	ppm	%	ppm	%	%	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm
	MDL	1	1	0.01	1	0.001	1	0.01	0.001	0.01	0.1	0.01	0.1	0.1	0.05	1	0.5	0.2
DRS56	Rock	1	19	1.05	139	0.128	3	1.24	0.078	0.31	<0.1	<0.01	4.4	<0.1	<0.05	4	<0.5	<0.2
DRS57	Rock	7	18	1.08	61	0.139	4	1.58	0.053	0.19	0.2	<0.01	1.5	<0.1	<0.05	5	<0.5	<0.2
DRS58	Rock	6	13	1.11	129	0.150	3	1.37	0.064	0.63	0.1	<0.01	2.1	<0.1	<0.05	5	<0.5	<0.2
DRS59	Rock	2	30	0.68	47	0.133	3	0.94	0.064	0.16	0.8	0.06	2.8	<0.1	0.12	3	0.7	<0.2
DRS60	Rock	<1	6	0.53	149	<0.001	1	0.09	0.005	0.05	<0.1	0.03	0.3	<0.1	0.36	<1	<0.5	<0.2
DRS61	Rock	3	5	1.32	101	0.150	2	1.63	0.091	0.28	0.2	<0.01	2.6	<0.1	<0.05	5	<0.5	<0.2
DRS62	Rock	5	7	1.07	43	0.117	3	1.20	0.086	0.12	0.3	0.01	2.9	<0.1	0.69	5	1.3	0.6
DRS63	Rock	4	7	2.43	562	0.181	2	2.50	0.066	2.01	0.1	<0.01	3.3	0.3	<0.05	7	<0.5	<0.2

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Project: Dillard
Report Date: October 29, 2011

QUALITY CONTROL REPORT VAN11005474.1

	Method	WGHT	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
	Analyte	Wgt	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P
	Unit	kg	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppb	ppm	ppm	ppm	ppm	ppm	ppm	%	%
	MDL	0.01	0.1	0.1	0.1	1	0.1	0.1	0.1	1	0.01	0.5	0.5	0.1	1	0.1	0.1	0.1	2	0.01	0.001
Pulp Duplicates																					
DRS47	Rock	1.93	<0.1	1179	6.7	88	0.3	7.6	21.0	1071	3.19	2.2	17.1	1.2	60	<0.1	0.3	<0.1	112	1.11	0.143
REP DRS47	QC		<0.1	1171	6.5	85	0.4	7.6	20.7	1044	3.06	2.4	14.0	1.1	55	0.1	0.3	<0.1	110	1.06	0.134
DRS55	Rock	1.78	0.5	39.5	9.2	59	0.2	9.6	12.1	430	2.98	3.7	4.8	0.8	82	0.1	0.5	0.3	69	0.87	0.131
REP DRS55	QC		0.4	39.8	9.0	59	0.2	9.8	12.0	414	2.94	3.8	6.7	0.8	84	0.4	0.5	0.2	68	0.91	0.134
Core Reject Duplicates																					
DRS57	Rock	1.41	0.7	379.0	17.6	98	0.5	8.2	8.9	964	3.94	2.7	3.2	0.7	86	0.3	0.2	<0.1	88	1.36	0.154
DUP DRS57	QC		0.7	411.8	18.5	100	0.4	8.8	9.1	962	3.98	2.4	4.7	0.7	74	0.3	0.1	<0.1	84	1.22	0.160
Reference Materials																					
STD DS8	Standard		12.4	108.8	122.2	301	1.8	38.1	7.2	573	2.36	24.1	120.4	7.1	66	2.2	5.4	7.2	40	0.67	0.073
STD DS8	Standard		13.0	105.8	125.2	302	1.7	36.0	7.1	604	2.46	24.7	117.9	6.7	75	2.4	6.4	7.7	41	0.73	0.072
STD DS8 Expected			13.44	110	123	312	1.69	38.1	7.5	615	2.46	26	107	6.89	67.7	2.38	5.7	6.67	41.1	0.7	0.08
BLK	Blank		<0.1	<0.1	<0.1	<1	<0.1	<0.1	<0.1	<1	<0.01	<0.5	<0.5	<0.1	<1	<0.1	<0.1	<0.1	<2	<0.01	<0.001
BLK	Blank		<0.1	<0.1	<0.1	<1	<0.1	<0.1	<0.1	<1	<0.01	<0.5	<0.5	<0.1	<1	<0.1	<0.1	<0.1	<2	<0.01	<0.001
Prep Wash																					
G1	Prep Blank	<0.01	<0.1	2.1	2.8	45	<0.1	2.2	3.9	532	1.83	<0.5	<0.5	5.7	64	<0.1	<0.1	<0.1	35	0.45	0.074
G1	Prep Blank	<0.01	<0.1	2.0	4.0	44	<0.1	2.3	3.8	530	1.87	<0.5	1.1	5.9	66	<0.1	<0.1	<0.1	36	0.46	0.074

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Project: Dillard
Report Date: October 29, 2011

QUALITY CONTROL REPORT VAN11005474.1

	Method	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
	Analyte	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Hg	Sc	Tl	S	Ga	Se	Te
	Unit	ppm	ppm	%	ppm	%	ppm	%	%	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm
	MDL	1	1	0.01	1	0.001	1	0.01	0.001	0.01	0.1	0.01	0.1	0.1	0.05	1	0.5	0.2
Pulp Duplicates																		
DRS47	Rock	6	12	1.63	53	0.136	3	1.73	0.045	0.52	0.1	<0.01	3.3	<0.1	0.06	7	1.4	<0.2
REP DRS47	QC	6	11	1.62	55	0.113	1	1.72	0.042	0.51	0.1	<0.01	3.1	<0.1	0.06	6	0.9	<0.2
DRS55	Rock	5	14	0.91	68	0.134	1	1.20	0.055	0.10	0.3	<0.01	1.5	<0.1	0.13	5	0.7	<0.2
REP DRS55	QC	5	14	0.91	66	0.136	6	1.16	0.055	0.10	0.3	<0.01	1.6	<0.1	0.13	5	0.7	<0.2
Core Reject Duplicates																		
DRS57	Rock	7	18	1.08	61	0.139	4	1.58	0.053	0.19	0.2	<0.01	1.5	<0.1	<0.05	5	<0.5	<0.2
DUP DRS57	QC	7	19	1.09	64	0.105	4	1.49	0.044	0.19	0.1	<0.01	1.4	<0.1	<0.05	5	<0.5	<0.2
Reference Materials																		
STD DS8	Standard	15	112	0.59	267	0.115	<1	0.92	0.094	0.41	2.8	0.20	2.0	5.4	0.15	4	5.7	4.8
STD DS8	Standard	17	114	0.63	270	0.121	5	0.97	0.090	0.42	3.0	0.21	1.7	5.2	0.16	5	5.0	4.8
STD DS8 Expected		14.6	115	0.6045	279	0.113	2.6	0.93	0.0883	0.41	3	0.192	2.3	5.4	0.1679	4.7	5.23	5
BLK	Blank	<1	<1	<0.01	<1	<0.001	<1	<0.01	<0.001	<0.01	<0.1	<0.01	<0.1	<0.1	<0.05	<1	<0.5	<0.2
BLK	Blank	<1	<1	<0.01	<1	<0.001	<1	<0.01	<0.001	<0.01	<0.1	<0.01	<0.1	<0.1	<0.05	<1	<0.5	<0.2
Prep Wash																		
G1	Prep Blank	13	5	0.50	154	0.116	<1	0.93	0.099	0.48	<0.1	<0.01	2.0	0.3	<0.05	5	<0.5	<0.2
G1	Prep Blank	12	5	0.50	155	0.118	<1	0.97	0.104	0.48	<0.1	<0.01	2.0	0.3	<0.05	4	<0.5	<0.2

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Submitted By: John Peters
Receiving Lab: Canada-Vancouver
Received: September 30, 2011
Report Date: October 22, 2011
Page: 1 of 2

CERTIFICATE OF ANALYSIS VAN11005157.1

CLIENT JOB INFORMATION

Project: Dillard
Shipment ID:
P.O. Number
Number of Samples: 30

SAMPLE PREPARATION AND ANALYTICAL PROCEDURES

Method Code	Number of Samples	Code Description	Test Wgt (g)	Report Status	Lab
R200-250	30	Crush, split and pulverize 250 g rock to 200 mesh			VAN
1DX2	30	1:1:1 Aqua Regia digestion ICP-MS analysis	15	Completed	VAN

SAMPLE DISPOSAL

STOR-PLP	Store After 90 days Invoice for Storage
DISP-RJT	Dispose of Reject After 90 days

Acme does not accept responsibility for samples left at the laboratory after 90 days without prior written instructions for sample storage or return.

ADDITIONAL COMMENTS

Invoice To: Fjordland Exploration Inc.
11th Floor, 1111 Melville Street
Vancouver BC V6E 3V6
Canada

CC: Tom Schroeter

BRITISH COLUMBIA CERTIFIED ASSAYER
CLARENCE LEONG
GENERAL MANAGER

This report supersedes all previous preliminary and final reports with this file number dated prior to the date on this certificate. Signature indicates final approval; preliminary reports are unsigned and should be used for reference only.
All results are considered the confidential property of the client. Acme assumes the liabilities for actual cost of analysis only. Results apply to samples as submitted.
"*" asterisk indicates that an analytical result could not be provided due to unusually high levels of interference from other elements.

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Project: Dillard
Report Date: October 22, 2011

CERTIFICATE OF ANALYSIS VAN11005157.1

Method		WGHT	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
Analyte		Wgt	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P
Unit		kg	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppb	ppm	ppm	ppm	ppm	ppm	ppm	%	%
MDL		0.01	0.1	0.1	0.1	1	0.1	0.1	0.1	1	0.01	0.5	0.5	0.1	1	0.1	0.1	0.1	2	0.01	0.001
DRS-01	Rock	1.71	0.2	754.2	4.5	41	0.6	14.4	22.0	251	3.08	2.2	23.4	0.1	33	0.3	0.2	0.1	102	1.22	0.115
DRS-02	Rock	1.24	0.2	93.3	2.6	70	0.1	7.8	11.8	548	3.04	4.0	3.6	0.5	70	0.3	0.2	<0.1	91	1.23	0.146
DRS-03	Rock	1.27	0.4	123.1	5.3	47	0.2	13.5	16.0	363	2.26	11.3	5.1	0.7	40	0.3	0.5	0.1	75	1.47	0.130
DRS-04	Rock	1.07	0.3	495.9	2.5	16	0.5	33.4	54.7	141	3.54	19.0	18.9	0.2	30	<0.1	0.2	0.7	22	0.87	0.046
DRS-05	Rock	1.36	0.1	3.8	0.2	1	<0.1	0.6	0.4	45	0.24	<0.5	12.2	<0.1	3	<0.1	0.1	<0.1	<2	<0.01	<0.001
DRS-06	Rock	1.37	0.8	394.5	0.8	55	0.3	13.6	24.1	657	4.05	2.2	15.9	0.5	39	<0.1	0.2	<0.1	163	0.92	0.129
DRS-07	Rock	1.27	12.8	183.2	8.2	7	3.8	144.1	269.6	68	31.77	26.4	175.9	<0.1	2	<0.1	2.5	1.9	16	0.02	0.011
DRS-08	Rock	1.56	2.2	31.9	2.0	2	<0.1	2.6	4.2	51	0.81	0.9	8.8	<0.1	2	<0.1	<0.1	<0.1	4	0.05	0.002
DRS-09	Rock	1.39	0.9	53.2	1.6	52	<0.1	26.5	24.0	564	3.53	3.6	9.4	0.5	34	<0.1	0.3	<0.1	121	0.91	0.110
DRS-10	Rock	1.17	2.3	65.2	0.9	3	0.2	1.2	3.7	55	0.74	2.8	16.4	<0.1	3	<0.1	0.2	<0.1	4	0.03	0.006
DRS-11	Rock	1.17	0.8	17.8	2.9	14	<0.1	3.1	6.6	221	2.04	1.5	4.8	10.9	26	<0.1	0.1	<0.1	74	0.63	0.021
DRS-12	Rock	1.50	0.3	497.0	1249	2526	3.9	41.0	33.4	4657	7.19	16.3	5.4	0.1	93	7.9	0.7	0.1	84	16.02	0.030
DRS-13	Rock	1.19	8.7	2.9	4.4	28	<0.1	1.9	3.9	295	1.67	0.6	1.8	2.8	39	<0.1	<0.1	<0.1	52	0.60	0.075
DRS-14	Rock	1.43	2.1	44.5	3.3	15	0.7	7.1	6.2	153	0.93	18.4	17.3	0.2	5	0.1	1.9	0.6	12	0.07	0.011
DRS-15	Rock	1.13	0.4	1173	13.9	107	1.1	9.0	18.6	1069	3.73	2.9	12.8	1.2	46	0.6	<0.1	<0.1	135	1.42	0.155
DRS-16	Rock	1.09	0.4	177.2	8.4	96	0.8	6.0	19.6	1856	3.94	35.4	6.5	0.8	143	0.9	0.9	<0.1	93	1.15	0.062
DRS-17	Rock	1.40	0.7	602.0	4.5	25	0.4	16.1	18.2	331	1.89	1.8	6.4	0.3	60	0.2	0.2	<0.1	80	1.14	0.124
DRS-18	Rock	1.26	0.5	59.7	22.3	97	0.2	5.4	6.8	395	1.51	2.8	1.6	0.9	22	1.2	0.1	<0.1	54	1.13	0.086
DRS-19	Rock	1.31	0.4	162.8	2.2	69	0.3	11.4	19.0	712	3.74	2.4	5.7	0.7	48	<0.1	<0.1	<0.1	133	0.92	0.143
DRS-20	Rock	1.54	5.3	187.4	4.0	28	0.3	9.7	11.9	251	2.64	2.6	19.7	0.8	68	0.1	0.2	<0.1	82	1.12	0.163
DRS-21	Rock	1.17	0.1	65.4	1.1	54	0.1	52.1	25.6	607	4.35	1.2	3.2	0.3	39	<0.1	<0.1	<0.1	127	1.20	0.119
DRS-22	Rock	1.14	0.1	983.6	295.0	71	0.8	33.8	18.5	675	4.06	5.9	10.8	0.2	87	0.3	0.4	0.3	144	2.03	0.108
DRS-23	Rock	1.25	0.3	465.4	13.5	115	0.7	9.3	36.9	1016	3.38	4.1	7.3	0.7	65	0.8	0.2	0.2	82	1.28	0.147
DRS-24	Rock	1.42	0.1	56.8	1.6	41	<0.1	64.5	7.5	1533	1.37	2.0	3.8	0.1	19	0.2	0.2	<0.1	32	1.63	0.031
DRS-25	Rock	1.47	0.5	814.4	3.7	30	0.6	45.6	42.8	436	2.39	5.4	33.8	0.2	61	0.2	0.5	0.3	50	1.68	0.136
DRS-65	Rock	1.39	0.3	558.6	2.2	73	0.2	42.8	24.1	673	2.66	4.9	10.9	<0.1	61	0.2	0.3	<0.1	92	1.55	0.119
DRS-66	Rock	1.23	0.4	491.5	2.0	28	0.4	40.0	12.7	446	1.97	4.2	21.3	0.3	50	0.1	0.3	0.1	47	1.45	0.139
DRS-67	Rock	1.40	0.8	182.3	0.7	32	<0.1	10.2	2.5	3862	9.49	8.5	3.2	0.5	5	0.2	0.5	<0.1	60	12.11	0.066
DRS-68	Rock	1.36	0.3	2770	3.2	16	1.8	3.7	1.8	2184	6.42	6.6	88.7	0.3	26	0.1	0.3	<0.1	85	11.76	0.081
DRS-69	Rock	1.84	1.0	51.2	0.6	17	<0.1	3.0	3.4	2319	10.31	8.3	6.3	0.2	8	<0.1	0.1	<0.1	44	15.00	0.038

Acme Analytical Laboratories (Vancouver) Ltd.
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11th Floor, 1111 Melville Street
Vancouver BC V6E 3V6 Canada

Project: Dillard
Report Date: October 22, 2011

CERTIFICATE OF ANALYSIS VAN11005157.1

	Method	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
	Analyte	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Hg	Sc	Tl	S	Ga	Se	Te
	Unit	ppm	ppm	%	ppm	%	ppm	%	%	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm
	MDL	1	1	0.01	1	0.001	1	0.01	0.001	0.01	0.1	0.01	0.1	0.1	0.05	1	0.5	0.2
DRS-01	Rock	1	16	0.67	58	0.132	2	1.47	0.117	0.28	0.1	<0.01	4.7	<0.1	0.89	5	3.6	0.2
DRS-02	Rock	3	13	0.83	39	0.107	2	1.20	0.122	0.22	0.1	<0.01	3.9	<0.1	<0.05	5	<0.5	<0.2
DRS-03	Rock	3	10	0.53	57	0.097	2	1.10	0.168	0.17	0.2	<0.01	5.1	0.1	0.26	4	0.6	<0.2
DRS-04	Rock	<1	4	0.12	46	0.058	2	0.83	0.133	0.11	0.3	<0.01	1.4	<0.1	2.06	2	4.7	0.5
DRS-05	Rock	<1	1	<0.01	3	<0.001	<1	0.02	0.004	<0.01	<0.1	<0.01	0.1	<0.1	<0.05	<1	<0.5	<0.2
DRS-06	Rock	4	10	1.74	254	0.221	2	2.02	0.093	1.39	0.1	0.02	4.1	0.2	0.26	6	<0.5	<0.2
DRS-07	Rock	<1	3	0.02	5	0.014	1	0.18	0.010	0.04	0.6	0.07	0.6	0.7	>10	<1	8.9	9.4
DRS-08	Rock	<1	1	0.02	7	0.002	<1	0.04	0.005	0.01	4.0	0.02	0.3	<0.1	0.41	<1	<0.5	<0.2
DRS-09	Rock	3	45	1.42	190	0.173	<1	1.62	0.085	0.72	0.1	0.05	4.0	0.2	0.12	7	<0.5	<0.2
DRS-10	Rock	<1	1	0.02	11	0.005	1	0.05	0.016	0.02	1.2	<0.01	0.3	<0.1	<0.05	<1	<0.5	<0.2
DRS-11	Rock	7	5	0.36	63	0.090	1	1.16	0.132	0.17	0.6	<0.01	1.8	0.2	<0.05	4	<0.5	<0.2
DRS-12	Rock	4	6	4.11	51	0.003	1	0.28	0.004	0.16	1.4	0.25	11.6	0.2	0.13	1	<0.5	<0.2
DRS-13	Rock	5	5	0.28	556	0.077	2	0.54	0.064	0.10	<0.1	<0.01	1.1	<0.1	<0.05	3	<0.5	<0.2
DRS-14	Rock	1	13	0.03	29	0.002	<1	0.09	0.016	0.04	0.2	0.04	1.2	<0.1	<0.05	<1	<0.5	0.8
DRS-15	Rock	6	20	1.80	254	0.161	3	1.88	0.107	0.96	0.1	<0.01	2.9	<0.1	0.09	7	<0.5	<0.2
DRS-16	Rock	6	4	0.51	2528	0.007	7	0.84	0.008	0.44	<0.1	0.33	6.8	0.1	0.07	2	<0.5	<0.2
DRS-17	Rock	3	25	0.78	30	0.140	5	1.07	0.097	0.18	<0.1	0.01	5.3	<0.1	0.24	3	1.0	<0.2
DRS-18	Rock	4	11	0.47	65	0.111	3	0.86	0.079	0.12	0.5	<0.01	2.0	<0.1	0.10	5	<0.5	<0.2
DRS-19	Rock	6	25	1.49	271	0.193	2	1.81	0.126	0.99	0.1	0.02	3.4	0.1	0.19	6	<0.5	<0.2
DRS-20	Rock	7	14	0.78	72	0.142	5	1.20	0.078	0.21	0.2	<0.01	2.4	<0.1	0.18	4	0.7	<0.2
DRS-21	Rock	3	84	2.33	342	0.182	2	1.86	0.111	1.30	<0.1	<0.01	3.9	<0.1	<0.05	6	<0.5	<0.2
DRS-22	Rock	1	58	1.48	141	0.184	4	2.04	0.304	0.40	<0.1	<0.01	6.9	<0.1	<0.05	6	<0.5	<0.2
DRS-23	Rock	5	12	1.33	122	0.157	2	1.56	0.072	0.40	0.2	<0.01	2.8	<0.1	0.18	4	<0.5	<0.2
DRS-24	Rock	1	56	0.67	46	0.073	<1	0.45	0.010	0.06	3.6	<0.01	2.7	<0.1	<0.05	1	<0.5	<0.2
DRS-25	Rock	2	104	0.95	35	0.127	1	1.03	0.053	0.08	0.2	<0.01	3.2	<0.1	0.74	3	1.4	0.6
DRS-65	Rock	2	58	1.72	229	0.183	2	1.84	0.046	0.57	0.3	<0.01	3.1	0.1	0.24	4	<0.5	<0.2
DRS-66	Rock	2	97	0.74	63	0.110	2	0.75	0.050	0.08	0.4	<0.01	3.7	<0.1	0.42	2	0.7	0.3
DRS-67	Rock	2	15	0.06	14	0.039	2	1.33	<0.001	<0.01	1.5	<0.01	3.5	<0.1	<0.05	7	<0.5	<0.2
DRS-68	Rock	2	17	0.23	176	0.079	<1	1.29	<0.001	<0.01	8.7	<0.01	2.8	<0.1	0.17	9	0.9	<0.2
DRS-69	Rock	1	7	0.13	9	0.031	1	0.85	0.001	<0.01	2.6	<0.01	1.9	<0.1	<0.05	8	<0.5	<0.2

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Vancouver BC V6E 3V6 Canada

Project: Dillard
Report Date: October 22, 2011

QUALITY CONTROL REPORT VAN11005157.1

	Method	WGHT	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
	Analyte	Wgt	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P
	Unit	kg	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppb	ppm	ppm	ppm	ppm	ppm	ppm	%	%
	MDL	0.01	0.1	0.1	0.1	1	0.1	0.1	0.1	1	0.01	0.5	0.5	0.1	1	0.1	0.1	0.1	2	0.01	0.001
Pulp Duplicates																					
DRS-16	Rock	1.09	0.4	177.2	8.4	96	0.8	6.0	19.6	1856	3.94	35.4	6.5	0.8	143	0.9	0.9	<0.1	93	1.15	0.062
REP DRS-16	QC		0.4	173.5	8.4	95	0.8	5.8	19.2	1849	3.79	34.4	3.9	0.8	133	0.9	1.0	<0.1	89	1.14	0.064
Core Reject Duplicates																					
DRS-06	Rock	1.37	0.8	394.5	0.8	55	0.3	13.6	24.1	657	4.05	2.2	15.9	0.5	39	<0.1	0.2	<0.1	163	0.92	0.129
DUP DRS-06	QC		0.7	379.3	0.8	52	0.3	12.7	23.6	609	3.82	2.0	13.8	0.4	34	<0.1	0.2	<0.1	149	0.84	0.121
Reference Materials																					
STD DS8	Standard		13.6	107.6	127.8	311	1.8	39.5	8.0	621	2.48	25.9	120.6	6.6	61	2.2	5.4	6.5	42	0.73	0.088
STD DS8	Standard		13.5	109.3	126.2	315	1.8	39.3	7.8	577	2.42	26.7	115.1	6.9	65	2.3	5.6	6.8	43	0.73	0.079
STD DS8 Expected			13.44	110	123	312	1.69	38.1	7.5	615	2.46	26	107	6.89	67.7	2.38	5.7	6.67	41.1	0.7	0.08
BLK	Blank		<0.1	<0.1	<0.1	<1	<0.1	<0.1	<0.1	<1	<0.01	<0.5	<0.5	<0.1	<1	<0.1	<0.1	<0.1	<2	<0.01	<0.001
BLK	Blank		<0.1	<0.1	<0.1	<1	<0.1	<0.1	<0.1	<1	<0.01	<0.5	<0.5	<0.1	<1	<0.1	<0.1	<0.1	<2	<0.01	<0.001
Prep Wash																					
G1	Prep Blank	<0.01	<0.1	3.4	2.6	44	<0.1	2.4	3.9	517	1.80	<0.5	2.8	4.9	53	<0.1	<0.1	<0.1	35	0.39	0.069
G1	Prep Blank	<0.01	<0.1	4.1	2.7	46	<0.1	2.5	3.9	508	1.82	<0.5	1.3	5.1	49	<0.1	<0.1	<0.1	36	0.39	0.077

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Project: Dillard
Report Date: October 22, 2011

QUALITY CONTROL REPORT VAN11005157.1

	Method	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15	1DX15
	Analyte	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Hg	Sc	Tl	S	Ga	Se	Te
	Unit	ppm	ppm	%	ppm	%	ppm	%	%	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm
	MDL	1	1	0.01	1	0.001	1	0.01	0.001	0.01	0.1	0.01	0.1	0.1	0.05	1	0.5	0.2
Pulp Duplicates																		
DRS-16	Rock	6	4	0.51	2528	0.007	7	0.84	0.008	0.44	<0.1	0.33	6.8	0.1	0.07	2	<0.5	<0.2
REP DRS-16	QC	5	5	0.50	2416	0.006	8	0.84	0.008	0.43	<0.1	0.35	6.4	0.1	0.07	2	<0.5	<0.2
Core Reject Duplicates																		
DRS-06	Rock	4	10	1.74	254	0.221	2	2.02	0.093	1.39	0.1	0.02	4.1	0.2	0.26	6	<0.5	<0.2
DUP DRS-06	QC	3	10	1.63	223	0.186	2	1.88	0.086	1.32	0.1	0.02	3.9	0.2	0.24	6	0.6	<0.2
Reference Materials																		
STD DS8	Standard	16	118	0.64	290	0.113	2	0.96	0.089	0.42	3.0	0.20	2.1	5.7	0.16	5	5.4	5.7
STD DS8	Standard	15	114	0.60	272	0.110	2	0.91	0.084	0.42	3.0	0.20	2.0	5.5	0.17	5	5.3	5.2
STD DS8 Expected		14.6	115	0.6045	279	0.113	2.6	0.93	0.0883	0.41	3	0.192	2.3	5.4	0.1679	4.7	5.23	5
BLK	Blank	<1	<1	<0.01	<1	<0.001	<1	<0.01	<0.001	<0.01	<0.1	<0.01	<0.1	<0.1	<0.05	<1	<0.5	<0.2
BLK	Blank	<1	<1	<0.01	<1	<0.001	<1	<0.01	<0.001	<0.01	<0.1	<0.01	<0.1	<0.1	<0.05	<1	<0.5	<0.2
Prep Wash																		
G1	Prep Blank	10	4	0.47	162	0.095	1	0.80	0.072	0.45	<0.1	<0.01	1.7	0.3	<0.05	4	<0.5	<0.2
G1	Prep Blank	10	4	0.47	161	0.092	1	0.83	0.069	0.45	<0.1	<0.01	1.7	0.3	<0.05	4	<0.5	<0.2

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CLIENT NAME: FJORDLAND EXPLORATIONS
11TH FLOOR-1111 MELVILLE ST
VANCOUVER, BC V6E3V6

ATTENTION TO: Tom Schroeter

PROJECT NO:

AGAT WORK ORDER: 11V516263

SOLID ANALYSIS REVIEWED BY: Ron Cardinall, Certified Assayer - Director - Technical Services (Mining)

DATE REPORTED: Aug 11, 2011

PAGES (INCLUDING COVER): 11

Should you require any information regarding this analysis please contact your client services representative at (905) 501 9998, or at 1-800-856-6261

*NOTES

All samples are stored at no charge for 90 days. Please contact the lab if you require additional sample storage time.

AGAT Laboratories

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Certificate of Analysis

AGAT WORK ORDER: 11V516263

PROJECT NO:

CLIENT NAME: FJORDLAND EXPLORATIONS

ATTENTION TO: Tom Schroeter

Aqua Regia Digest - Metals Package, ICP-OES finish (201073)

DATE SAMPLED: Aug 04, 2011 | DATE RECEIVED: Aug 04, 2011 | DATE REPORTED: Aug 11, 2011 | SAMPLE TYPE: Rock

Analyte:	Ag	Al	As	B	Ba	Be	Bi	Ca	Cd	Ce	Co	Cr	Cu	Fe
Unit:	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%
Sample Description RDL:	0.2	0.01	1	5	1	0.5	1	0.01	0.5	1	0.5	0.5	0.5	0.01
DILL 2	<0.2	1.55	12	7	65	<0.5	<1	1.61	<0.5	7	16.6	40.9	64.8	2.52
DILL 3	<0.2	2.05	6	<5	87	<0.5	<1	2.12	<0.5	28	15.0	17.7	700	3.22
DILL 4	<0.2	1.66	4	<5	180	<0.5	<1	1.43	<0.5	6	16.4	107	353	2.60
DILL 5	0.8	2.08	12	12	60	<0.5	<1	2.42	<0.5	10	18.5	23.7	215	2.78
DILL 7	0.2	1.54	4	14	67	<0.5	<1	1.20	<0.5	20	22.8	39.6	1400	3.85
DILL 8a	7.4	1.33	4	<5	55	<0.5	<1	1.08	<0.5	<1	12.8	31.6	>10000	3.06
DILL 8b	0.8	1.68	4	6	58	<0.5	<1	1.08	<0.5	22	11.3	30.7	2070	3.40
DILL 9	<0.2	1.99	5	7	94	<0.5	<1	1.01	<0.5	18	21.5	37.6	464	4.09
DILL 10	<0.2	1.99	5	8	229	<0.5	<1	2.57	<0.5	18	9.9	29.2	208	3.81
DILL 12	<0.2	1.01	7	10	30	<0.5	<1	5.36	<0.5	13	13.6	176	75.4	4.00
DILL 13a	11.4	0.89	9	8	68	<0.5	<1	0.41	<0.5	9	7.1	123	235	6.06
DILL 14	2.3	1.17	5	7	47	<0.5	<1	1.37	<0.5	8	23.2	104	1250	3.98
DILL 16	0.2	1.21	6	5	44	<0.5	<1	2.26	<0.5	10	13.8	157	59.0	3.76
DILL 20	<0.2	1.00	10	10	123	<0.5	1	0.98	<0.5	16	22.2	37.2	202	4.53
DILL 21	<0.2	0.75	18	8	19	0.6	<1	3.95	<0.5	12	4.6	21.5	37.7	2.80
DILL 22	<0.2	0.77	7	7	49	<0.5	<1	1.05	<0.5	14	7.8	27.2	20.5	1.78
DILL 24	0.4	0.70	5	<5	37	<0.5	<1	1.07	<0.5	14	4.6	44.4	1130	1.81
DILL 25	<0.2	2.02	4	7	118	<0.5	<1	1.50	<0.5	12	20.9	44.8	628	3.52
DILL 26	1.7	1.01	6	6	29	<0.5	<1	1.29	<0.5	10	18.3	38.2	2110	1.94
DILL 27	0.6	1.53	4	6	23	<0.5	<1	1.31	<0.5	19	22.0	24.2	1680	3.86
DILL 28	0.6	0.93	6	6	30	<0.5	<1	1.30	<0.5	13	9.3	41.1	174	1.98
WN-Tis-1	0.3	2.08	19	27	70	<0.5	<1	1.93	<0.5	13	56.2	553	2000	9.05

Certified By: Ron Cardinall

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Certificate of Analysis

AGAT WORK ORDER: 11V516263
PROJECT NO:

CLIENT NAME: FJORDLAND EXPLORATIONS

ATTENTION TO: Tom Schroeter

Aqua Regia Digest - Metals Package, ICP-OES finish (201073)

DATE SAMPLED: Aug 04, 2011 | DATE RECEIVED: Aug 04, 2011 | DATE REPORTED: Aug 11, 2011 | SAMPLE TYPE: Rock

Analyte:	Ga	Hg	In	K	La	Li	Mg	Mn	Mo	Na	Ni	P	Pb	Rb
Unit:	ppm	ppm	ppm	%	ppm	ppm	%	ppm	ppm	%	ppm	ppm	ppm	ppm
Sample Description RDL:	5	1	1	0.01	1	1	0.01	1	0.5	0.01	0.5	10	0.5	10
DILL 2	<5	<1	<1	0.22	<1	14	1.17	407	4.1	0.11	10.5	927	6.6	15
DILL 3	6	<1	<1	0.63	16	13	1.69	997	0.8	0.06	4.9	1410	4.1	59
DILL 4	<5	<1	<1	0.82	<1	8	1.60	434	<0.5	0.05	31.3	716	3.9	55
DILL 5	<5	<1	<1	0.17	2	20	0.99	684	1.0	0.14	6.7	1360	8.8	18
DILL 7	<5	<1	<1	0.86	6	7	1.18	599	<0.5	0.08	2.2	1180	5.0	71
DILL 8a	6	<1	<1	0.23	5	5	1.12	356	56.9	0.06	5.8	1240	27.4	24
DILL 8b	<5	<1	<1	0.44	8	8	1.52	587	3.8	0.07	4.1	1290	6.8	42
DILL 9	<5	<1	<1	1.08	5	14	1.91	1050	0.5	0.07	6.8	1220	5.4	97
DILL 10	<5	<1	<1	1.01	5	19	1.82	1290	0.9	0.08	3.3	1150	7.7	60
DILL 12	<5	<1	<1	0.33	3	5	1.50	1070	0.6	<0.01	66.1	734	1.6	52
DILL 13a	<5	6	<1	0.13	<1	8	0.42	117	3.0	0.09	6.3	321	14.9	<10
DILL 14	<5	<1	<1	0.53	<1	5	1.09	353	40.2	0.09	19.2	1220	3.6	51
DILL 16	<5	<1	<1	0.06	2	5	0.89	550	1.1	0.02	20.2	1080	2.0	<10
DILL 20	<5	<1	<1	0.15	4	9	0.60	209	0.6	0.08	27.8	1560	4.0	19
DILL 21	<5	<1	<1	0.22	3	5	0.74	598	<0.5	<0.01	8.7	1010	1.2	34
DILL 22	<5	<1	<1	0.17	4	6	0.29	179	<0.5	0.07	5.9	754	4.5	16
DILL 24	<5	<1	<1	0.09	5	3	0.34	230	7.2	0.06	3.7	978	3.5	<10
DILL 25	<5	<1	<1	1.09	3	11	1.59	505	0.9	0.11	10.8	880	4.7	128
DILL 26	<5	<1	<1	0.13	3	5	0.58	238	12.1	0.07	5.6	1080	6.8	13
DILL 27	<5	<1	<1	0.23	6	9	1.44	487	1.4	0.06	4.2	1220	4.9	24
DILL 28	<5	<1	<1	0.10	4	3	0.40	208	6.9	0.07	4.3	1170	2.9	<10
WN-Tis-1	<5	<1	<1	0.11	1	10	0.32	359	2.7	0.04	116	826	1.9	12

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Certificate of Analysis

AGAT WORK ORDER: 11V516263
PROJECT NO:

CLIENT NAME: FJORDLAND EXPLORATIONS

ATTENTION TO: Tom Schroeter

Aqua Regia Digest - Metals Package, ICP-OES finish (201073)

DATE SAMPLED: Aug 04, 2011 | DATE RECEIVED: Aug 04, 2011 | DATE REPORTED: Aug 11, 2011 | SAMPLE TYPE: Rock

Analyte:	S	Sb	Sc	Se	Sn	Sr	Ta	Te	Th	Ti	Tl	U	V	W
Unit:	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
Sample Description RDL:	0.005	1	0.5	10	5	0.5	10	10	5	0.01	5	5	0.5	1
DILL 2	0.998	<1	5.8	<10	<5	58.9	<10	<10	<5	0.21	7	<5	102	<1
DILL 3	0.079	<1	9.6	<10	<5	34.3	<10	<10	<5	0.08	<5	<5	180	<1
DILL 4	0.025	<1	4.5	<10	<5	85.7	<10	<10	<5	0.38	12	<5	106	<1
DILL 5	0.241	<1	6.5	<10	<5	47.6	<10	<10	<5	0.23	7	<5	134	<1
DILL 7	0.573	<1	5.0	<10	<5	66.2	<10	<10	<5	0.21	8	<5	143	<1
DILL 8a	1.70	<1	4.2	11	<5	106	<10	<10	<5	0.25	7	<5	110	<1
DILL 8b	0.173	<1	5.5	<10	<5	79.6	<10	<10	<5	0.22	8	<5	120	<1
DILL 9	0.033	<1	4.9	<10	<5	45.9	<10	<10	<5	0.27	10	<5	134	<1
DILL 10	0.093	<1	5.4	<10	<5	88.5	<10	<10	<5	0.33	11	<5	133	<1
DILL 12	0.099	3	18.2	<10	<5	103	<10	<10	<5	0.01	<5	<5	120	<1
DILL 13a	2.83	<1	3.7	<10	<5	41.5	<10	<10	<5	0.09	<5	<5	70.3	<1
DILL 14	0.594	<1	6.1	<10	<5	51.9	<10	<10	<5	0.31	9	<5	115	<1
DILL 16	0.546	1	5.0	<10	<5	101	<10	<10	<5	0.16	<5	<5	88.0	<1
DILL 20	3.43	2	4.7	<10	<5	24.8	<10	<10	<5	0.22	<5	<5	106	<1
DILL 21	0.383	3	16.1	<10	<5	38.2	<10	<10	<5	<0.01	<5	<5	113	<1
DILL 22	0.917	1	5.3	<10	<5	26.2	<10	<10	<5	0.21	<5	<5	60.9	<1
DILL 24	0.134	1	2.9	<10	<5	79.9	<10	<10	<5	0.18	5	<5	68.9	<1
DILL 25	0.438	<1	6.8	10	<5	104	<10	<10	<5	0.34	11	<5	146	<1
DILL 26	0.578	1	4.3	<10	<5	101	<10	<10	<5	0.23	6	<5	77.7	<1
DILL 27	1.40	<1	4.1	<10	<5	92.4	<10	<10	<5	0.29	9	<5	134	<1
DILL 28	0.302	<1	4.1	<10	<5	119	<10	<10	<5	0.23	5	<5	80.2	<1
WN-Tis-1	2.37	<1	2.3	<10	<5	53.7	<10	<10	<5	0.10	<5	<5	162	<1

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AGAT WORK ORDER: 11V516263
PROJECT NO:

CLIENT NAME: FJORDLAND EXPLORATIONS

ATTENTION TO: Tom Schroeter

Aqua Regia Digest - Metals Package, ICP-OES finish (201073)

DATE SAMPLED: Aug 04, 2011 | DATE RECEIVED: Aug 04, 2011 | DATE REPORTED: Aug 11, 2011 | SAMPLE TYPE: Rock

Analyte:	Y	Zn	Zr	Cu-OL
Unit:	ppm	ppm	ppm	%
Sample Description RDL:	1	0.5	5	0.02
DILL 2	5	51.8	<5	
DILL 3	21	56.4	<5	
DILL 4	5	65.5	<5	
DILL 5	9	92.6	<5	
DILL 7	11	58.1	<5	
DILL 8a	10	53.7	8	1.64
DILL 8b	13	81.0	6	
DILL 9	8	112	<5	
DILL 10	9	104	6	
DILL 12	9	80.0	<5	
DILL 13a	2	39.6	<5	
DILL 14	6	66.2	<5	
DILL 16	5	36.3	<5	
DILL 20	10	27.9	<5	
DILL 21	10	64.6	<5	
DILL 22	12	34.8	<5	
DILL 24	7	23.2	<5	
DILL 25	8	47.8	<5	
DILL 26	8	28.5	6	
DILL 27	12	68.1	9	
DILL 28	8	18.1	7	
WN-Tis-1	4	64.9	<5	

Comments: RDL - Reported Detection Limit

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AGAT WORK ORDER: 11V516263
PROJECT NO:

CLIENT NAME: FJORDLAND EXPLORATIONS

ATTENTION TO: Tom Schroeter

Fire Assay - Trace Au, ICP-OES finish (202052)

DATE SAMPLED: Aug 04, 2011 | DATE RECEIVED: Aug 04, 2011 | DATE REPORTED: Aug 11, 2011 | SAMPLE TYPE: Rock

Analyte:	Au	Sample Login Weight	Au-Grav
Unit:	ppm	kg	g/t
Sample Description RDL:	0.001	0.01	0.05
DILL 2	0.010	0.73	-
DILL 3	0.008	1.31	-
DILL 4	0.004	0.53	-
DILL 5	0.010	1.00	-
DILL 7	0.020	1.05	-
DILL 8a	0.025	0.75	-
DILL 8b	0.004	0.55	-
DILL 9	0.011	1.15	-
DILL 10	0.003	0.56	-
DILL 12	0.008	1.05	-
DILL 13a	>10	0.44	28.9
DILL 14	0.219	0.33	-
DILL 16	0.038	0.56	-
DILL 20	0.039	0.81	-
DILL 21	0.021	0.66	-
DILL 22	0.021	0.51	-
DILL 24	0.039	0.60	-
DILL 25	0.024	0.50	-
DILL 26	0.081	0.52	-
DILL 27	0.010	0.30	-
DILL 28	0.068	1.11	-
WN-Tis-1	0.025	0.59	-

Comments: RDL - Reported Detection Limit

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Quality Assurance

CLIENT NAME: FJORDLAND EXPLORATIONS

AGAT WORK ORDER: 11V516263

PROJECT NO:

ATTENTION TO: Tom Schroeter

Solid Analysis

RPT Date: Aug 11, 2011		REPLICATE				Method Blank	REFERENCE MATERIAL				
PARAMETER	Batch	Sample Id	Original	Rep #1	RPD		Result Value	Expect Value	Recovery	Acceptable Limits	
										Lower	Upper
Fire Assay - Trace Au, ICP-OES finish (202052)											
Au	1	2591115	0.010	0.010	0.0%	< 0.001	0.98	0.922	106%	80%	120%
Au-Grav	1	2591125	28.9	28.1	2.8%	< 0.05	5.98	5.909	101%	90%	110%
Fire Assay - Trace Au, ICP-OES finish (202052)											
Au	1	2591127	0.0383	0.0391	2.1%	< 0.001	0.0855	0.0849	100%	80%	120%
Fire Assay - Trace Au, ICP-OES finish (202052)											
Au	1	2591185	0.025	0.024	4.1%	< 0.001	0.205	0.203	100%	80%	120%
Aqua Regia Digest - Metals Package, ICP-OES finish (201073)											
Ag	1	2591115	< 0.2	< 0.2	0.0%	< 0.2				80%	120%
Al	1	2591115	1.55	1.55	0.0%	< 0.01				80%	120%
As	1	2591115	12	12	0.0%	< 1				80%	120%
B	1	2591115	7	6	15.4%	< 5				80%	120%
Ba	1	2591115	65	63	3.1%	< 1				80%	120%
Be	1	2591115	< 0.5	< 0.5	0.0%	< 0.5				80%	120%
Bi	1	2591115	< 1	< 1	0.0%	< 1				80%	120%
Ca	1	2591115	1.61	1.60	0.6%	< 0.01				80%	120%
Cd	1	2591115	< 0.5	< 0.5	0.0%	< 0.5				80%	120%
Ce	1	2591115	7	6	15.4%	< 1				80%	120%
Co	1	2591115	16.6	17.4	4.7%	< 0.5				80%	120%
Cr	1	2591115	40.9	42.2	3.1%	< 0.5				80%	120%
Cu	1	2591115	64.8	67.1	3.5%	< 0.5				80%	120%
Fe	1	2591115	2.52	2.50	0.8%	< 0.01				80%	120%
Ga	1	2591115	< 5	< 5	0.0%	< 5				80%	120%
Hg	1	2591115	< 1	< 1	0.0%	< 1				80%	120%
In	1	2591115	< 1	< 1	0.0%	< 1				80%	120%
K	1	2591115	0.22	0.22	0.0%	< 0.01				80%	120%
La	1	2591115	< 1	< 1	0.0%	< 1				80%	120%
Li	1	2591115	14	14	0.0%	< 1				80%	120%
Mg	1	2591115	1.17	1.16	0.9%	< 0.01				80%	120%
Mn	1	2591115	407	419	2.9%	< 1				80%	120%
Mo	1	2591115	4.1	3.6	13.0%	< 0.5				80%	120%
Na	1	2591115	0.11	0.11	0.0%	< 0.01				80%	120%
Ni	1	2591115	10.5	10.6	0.9%	< 0.5				80%	120%
P	1	2591115	927	934	0.8%	< 10				80%	120%
Pb	1	2591115	6.6	6.9	4.4%	< 0.5				80%	120%
Rb	1	2591115	15	16	6.5%	< 10				80%	120%
S	1	2591115	0.998	0.975	2.3%	< 0.005				80%	120%
Sb	1	2591115	< 1	< 1	0.0%	< 1				80%	120%
Sc	1	2591115	5.8	5.8	0.0%	< 0.5				80%	120%
Se	1	2591115	< 10	< 10	0.0%	< 10				80%	120%
Sn	1	2591115	< 5	< 5	0.0%	< 5				80%	120%
Sr	1	2591115	58.9	57.6	2.2%	2.1				80%	120%
Ta	1	2591115	< 10	< 10	0.0%	< 10				80%	120%

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Quality Assurance

CLIENT NAME: FJORDLAND EXPLORATIONS
AGAT WORK ORDER: 11V516263
PROJECT NO:
ATTENTION TO: Tom Schroeter

Solid Analysis (Continued)

RPT Date: Aug 11, 2011		REPLICATE				Method Blank	REFERENCE MATERIAL				
PARAMETER	Batch	Sample Id	Original	Rep #1	RPD		Result Value	Expect Value	Recovery	Acceptable Limits Lower	Acceptable Limits Upper
Te	1	2591115	< 10	< 10	0.0%	< 10				80%	120%
Th	1	2591115	< 5	< 5	0.0%	< 5				80%	120%
Ti	1	2591115	0.205	0.204	0.5%	< 0.01				80%	120%
Tl	1	2591115	7	7	0.0%	< 5				80%	120%
U	1	2591115	< 5	< 5	0.0%	< 5				80%	120%
V	1	2591115	102	105	2.9%	< 0.5				80%	120%
W	1	2591115	< 1	< 1	0.0%	< 1				80%	120%
Y	1	2591115	5	5	0.0%	< 1				80%	120%
Zn	1	2591115	51.8	54.0	4.2%	6.6				80%	120%
Zr	1	2591115	< 5	< 5	0.0%	< 5				80%	120%
Aqua Regia Digest - Metals Package, ICP-OES finish (201073)											
Ag	1	2591185	0.27	0.20	29.8%	< 0.2				80%	120%
Al	1	2591185	2.08	2.29	9.6%	< 0.01				80%	120%
As	1	2591185	19	17	11.1%	< 1				80%	120%
B	1	2591185	27	28	3.6%	< 5				80%	120%
Ba	1	2591185	70	81	14.6%	< 1				80%	120%
Be	1	2591185	< 0.5	< 0.5	0.0%	< 0.5				80%	120%
Bi	1	2591185	< 1	< 1	0.0%	< 1				80%	120%
Ca	1	2591185	1.93	2.16	11.2%	< 0.01				80%	120%
Cd	1	2591185	< 0.5	< 0.5	0.0%	< 0.5				80%	120%
Ce	1	2591185	13	14	7.4%	< 1				80%	120%
Co	1	2591185	56.2	58.0	3.2%	< 0.5				80%	120%
Cr	1	2591185	553	560	1.3%	< 0.5				80%	120%
Cu	1	2591185	2000	2030	1.5%	< 0.5				80%	120%
Fe	1	2591185	9.05	9.55	5.4%	< 0.01				80%	120%
Ga	1	2591185	< 5	< 5	0.0%	< 5				80%	120%
Hg	1	2591185	< 1	1		< 1				80%	120%
In	1	2591185	< 1	< 1	0.0%	< 1				80%	120%
K	1	2591185	0.11	0.12	8.7%	< 0.01				80%	120%
La	1	2591185	1	1	0.0%	< 1				80%	120%
Li	1	2591185	10	11	9.5%	< 1				80%	120%
Mg	1	2591185	0.32	0.34	6.1%	< 0.01				80%	120%
Mn	1	2591185	359	382	6.2%	< 1				80%	120%
Mo	1	2591185	2.7	2.1	25.0%	< 0.5				80%	120%
Na	1	2591185	0.04	0.04	0.0%	< 0.01				80%	120%
Ni	1	2591185	116	118	1.7%	< 0.5				80%	120%
P	1	2591185	826	768	7.3%	< 10				80%	120%
Pb	1	2591185	1.9	1.6	17.1%	< 0.5				80%	120%
Rb	1	2591185	12	14	15.4%	< 10				80%	120%
S	1	2591185	2.37	2.45	3.3%	< 0.005				80%	120%
Sb	1	2591185	< 1	< 1	0.0%	< 1				80%	120%
Sc	1	2591185	2.3	2.4	4.3%	< 0.5				80%	120%
Se	1	2591185	< 10	< 10	0.0%	< 10				80%	120%
Sn	1	2591185	< 5	< 5	0.0%	< 5				80%	120%
Sr	1	2591185	53.7	61.6	13.7%	< 0.5				80%	120%

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CLIENT NAME: FJORDLAND EXPLORATIONS

AGAT WORK ORDER: 11V516263

PROJECT NO:

ATTENTION TO: Tom Schroeter

Solid Analysis (Continued)

RPT Date: Aug 11, 2011		REPLICATE				Method Blank	REFERENCE MATERIAL				
PARAMETER	Batch	Sample Id	Original	Rep #1	RPD		Result Value	Expect Value	Recovery	Acceptable Limits Lower	Acceptable Limits Upper
Ta	1	2591185	< 10	< 10	0.0%	< 10				80%	120%
Te	1	2591185	< 10	< 10	0.0%	< 10				80%	120%
Th	1	2591185	< 5	< 5	0.0%	< 5				80%	120%
Ti	1	2591185	0.10	0.11	9.5%	< 0.01				80%	120%
Tl	1	2591185	< 5	< 5	0.0%	< 5				80%	120%
U	1	2591185	< 5	< 5	0.0%	< 5				80%	120%
V	1	2591185	162	167	3.0%	< 0.5				80%	120%
W	1	2591185	< 1	< 1	0.0%	< 1				80%	120%
Y	1	2591185	4	4	0.0%	< 1				80%	120%
Zn	1	2591185	64.9	63.2	2.7%	< 0.5				80%	120%
Zr	1	2591185	< 5	< 5	0.0%	< 5				80%	120%
Aqua Regia Digest - Metals Package, ICP-OES finish (201073)											
Cu-OL	1					< 0.02	0.52	0.50	104%	90%	110%

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Method Summary

CLIENT NAME: FJORDLAND EXPLORATIONS

AGAT WORK ORDER: 11V516263

PROJECT NO:

ATTENTION TO: Tom Schroeter

PARAMETER	AGAT S.O.P	LITERATURE REFERENCE	ANALYTICAL TECHNIQUE
Solid Analysis			
Ag	MIN-200-12020		ICP/OES
Al	MIN-200-12020		ICP/OES
As	MIN-200-12020		ICP/OES
B	MIN-200-12020		ICP/OES
Ba	MIN-200-12020		ICP/OES
Be	MIN-200-12020		ICP/OES
Bi	MIN-200-12020		ICP/OES
Ca	MIN-200-12020		ICP/OES
Cd	MIN-200-12020		ICP/OES
Ce	MIN-200-12020		ICP/OES
Co	MIN-200-12020		ICP/OES
Cr	MIN-200-12020		ICP/OES
Cu	MIN-200-12020		ICP/OES
Fe	MIN-200-12020		ICP/OES
Ga	MIN-200-12020		ICP/OES
Hg	MIN-200-12020		ICP/OES
In	MIN-200-12020		ICP/OES
K	MIN-200-12020		ICP/OES
La	MIN-200-12020		ICP/OES
Li	MIN-200-12020		ICP/OES
Mg	MIN-200-12020		ICP/OES
Mn	MIN-200-12020		ICP/OES
Mo	MIN-200-12020		ICP/OES
Na	MIN-200-12020		ICP/OES
Ni	MIN-200-12020		ICP/OES
P	MIN-200-12020		ICP/OES
Pb	MIN-200-12020		ICP/OES
Rb	MIN-200-12020		ICP/OES
S	MIN-200-12020		ICP/OES
Sb	MIN-200-12020		ICP/OES
Sc	MIN-200-12020		ICP/OES
Se	MIN-200-12020		ICP/OES
Sn	MIN-200-12020		ICP/OES
Sr	MIN-200-12020		ICP/OES
Ta	MIN-200-12020		ICP/OES
Te	MIN-200-12020		ICP/OES
Th	MIN-200-12020		ICP/OES
Ti	MIN-200-12020		ICP/OES
Tl	MIN-200-12020		ICP/OES
U	MIN-200-12020		ICP/OES
V	MIN-200-12020		ICP/OES
W	MIN-200-12020		ICP/OES
Y	MIN-200-12020		ICP/OES
Zn	MIN-200-12020		ICP/OES
Zr	MIN-200-12020		ICP/OES
Cu-OL			AA
Au	MIN-200-12006	BUGBEE, E: A Textbook of Fire Assaying	ICP-OES
Sample Login Weight	MIN-12009		BALANCE

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Method Summary

CLIENT NAME: FJORDLAND EXPLORATIONS

AGAT WORK ORDER: 11V516263

PROJECT NO:

ATTENTION TO: Tom Schroeter

PARAMETER	AGAT S.O.P	LITERATURE REFERENCE	ANALYTICAL TECHNIQUE
Au-Grav			GRAVIMETRIC

Appendix C:
Laboratory Procedures

Fjordland TAK Soils Project Sample Preparation Methodology Summary

DRYING OF MINERAL TESTING SAMPLES – MINING BRANCH OFFICES OVERVIEW: MIN-200-12008

INTRODUCTION AND SCOPE

This procedure describes the process for drying samples that will undergo analysis in the Mining Geochemistry Assay Division. Most samples contain certain amount of water as a hydrate or as occluded or surface absorbed water. There are several factors affecting moisture content including atmospheric humidity and particle size. Drying is the first step for sample preparation and is required to ensure that a homogeneous sample can be obtained. This will reduce error and bias in the analyses. Upon arrival the samples may appear dry, wet or excessively wet, however most samples require drying, as a pretreatment for the assigned tests such as sieving, fusions, digestions, etc. The types of samples include rocks, core and other drill samples, minerals, concentrates, tills, sands, soils, stream sediments, and dump and grab samples.

PRINCIPLE OF THE METHOD

The purpose of drying is usually to make the sample anhydrous or to remove absorbed moisture but retain chemically combined water. Drying temperatures above 100°C result in the loss of the water of hydration of some minerals, which affects the mass balance of whole rock analysis. It is preferred to dry samples at lower temperatures for extended periods of time (12 – 24 hours). Once the samples are received, they are placed into trays that will go in the oven at 60 ± 10°C for a period of time depending on the sample. Afterwards, the samples will be ready for the next step of analysis.

SAMPLE REQUIREMENTS

The whole amount of sample received should be dried. The temperature of the drying oven should be set at 60 ± 10°C.

SCREEN ANALYSIS AND PARTICLE SIZE DISTRIBUTION OF MINERALOGICAL SAMPLES OVERVIEW: MIN-200-12007

INTRODUCTION AND SCOPE

Many natural and manufactured materials occur in a disperse form, which means that they consist of differently shaped and sized particles. Sieving is used to isolate a particular particle size or to determinate the particle size distribution of the samples (i.e. the number of particles of different sizes), which can be related to important physical and chemical properties of solids, such as mechanical bulk behavior, surface reaction, miscibility, filtration properties, conductivity, etc. The types of samples include rocks, core and other drill samples, minerals, concentrates, tills, sands, soils, stream sediments, and dump and grab samples.

This overview focuses on one of two types of sieve analyses described in this procedure: Screen Analysis, where the sample is passed through a single sieve.

PRINCIPLE OF THE METHOD

Screen Analysis is used to determine the retained and passing fraction through a specific sieve. For the majority of client soils projects 80 mesh (180 μm) sieves are used. The retained portion is also referred as plus (+) portion and the passing is called minus (-) portion. The results are reported as percentage of the passing fraction relative to the total mass of sample.

During sieving the sample is subjected to horizontal and vertical movement. This causes a relative movement between the particles and the sieve; depending on their size the individual particles either pass through the sieve mesh or are retained on the sieve surface. The likelihood of a particle passing through the sieve mesh is determined by the ratio of the particle size to the sieve openings, the orientation of the particle and the number of encounters between the particle and the mesh openings.

SAMPLE REQUIREMENTS

The samples received may need preparation, or may be prepared by the customer (ready as received), or prepared by a different company. Thus, unless the sample is specifically defined as dry, the sample needs to be dried at 60 + 10°C as described in the SOP for drying. For samples with high clay content (particles under 75μm are classified as clay particles) some clumping could be present. In this case the clumps must be broken up with (gloved) fingers or mortar and pestle, and returned to the oven for further drying. The minimum amount of sample required is 100g.

DETERMINATION OF GOLD, PLATINUM AND PALLADIUM IN GEOLOGICAL SAMPLES BY LEAD FUSION FIRE ASSAY WITH INDUCTIVELY COUPLED PLASMA – OPTICAL EMISSION SPECTROSCOPY (ICP-OES) FINISH OVERVIEW: MIN-200-12006

INTRODUCTION AND SCOPE

This method determines the concentration of gold, platinum and palladium in many types of solid matrices by Inductively Coupled Plasma - Optical Emission Spectroscopy (ICP-OES) following fire assay and aqua regia digestion of the raw material. The types of samples include rocks, core and other drill samples, minerals, concentrates, tills, sands, soils, stream sediments, slurries, and dump and grab samples.

PRINCIPLE OF THE METHOD

Once the samples have undergone Fire Assay treatment, the resultant doré bead is attacked by wet chemical digestion (aqua regia) and then the instrumental finish is carried out using ICP-OES.

Inductively Coupled Plasma – Optical Emission Spectroscopy is an analytical technique used for the detection of trace metals. It is a type of emission spectroscopy that uses the inductively coupled plasma to produce excited atoms and ions that emit electromagnetic radiation at wavelengths characteristic of a particular element. The intensity of this emission is indicative of the concentration of the element within the sample.

SAMPLE REQUIREMENTS

The samples received may need preparation, or may be prepared by the client (ready as received), or prepared by a different company. Thus, unless the sample is specifically defined as dry, the sample needs to be dried at 60°C. Some samples may also require crushing, splitting and/or milling depending on the package selected by the client and the type of material to be analyzed. The samples are treated to fire assay and then the bead doré is submitted to digestion.

Quality Control

Reagent Blank: is run every 20 samples or once per fire assay set.

QC Solutions: are run at the beginning and end of the instrument data acquisition and also run every 20 samples for Calibration Verification.

Certified Reference Materials (CRM): a reference materials is used to verify calibration and fire assay conditions. A certified reference material must be weighed at least every 20 samples or once per fire assay set.

Replicates: every 20 samples or once per fire assay set a sample is chosen at random and weighed and fused in replicate.

Method Blank: every 40 samples or once per fire assay set a blank is fused (containing no sample).

DETERMINATION OF METALS IN GEOLOGICAL SAMPLES USING AN AQUA REGIA (NITRIC AND HYDROCHLORIC ACID) DIGESTION AND A COMBINATION OF INDUCTIVELY COUPLED PLASMA – OPTICAL EMISSION SPECTROSCOPY (ICP-OES) AND INDUCTIVELY COUPLED PLASMA MASS SPECTROSCOPY (ICP-MS) OVERVIEW: MIN-200-12018

INTRODUCTION AND SCOPE

This method describes the digestion with four acids in many types of solid matrices prior to instrumental determination by Inductively Coupled Plasma - Optical Emission Spectroscopy (ICP-OES) and Inductively Coupled Plasma – Mass Spectrometry (ICP-MS). The types of samples include metal bearing ores and related materials, rocks, core and other drill samples, minerals, concentrates, tills, sands, soils, stream sediments, and dump and grab samples.

PRINCIPLE OF THE METHOD

Aqua Regia digestions are used in the digestion of certain geological samples and are effective for most base metal sulphates, sulphides, oxides and carbonates. It is noted that aqua regia only provides a partial digestion for most rock forming elements and elements of a refractory nature. Each sample of ~ 1.0 g is digested with a 3:1 hot mixture of hydrochloric and nitric acids for one hour. The resultant product is dissolved and diluted to 50 mL with deionized water. An aliquot is measured by a suitable spectrometry instrument.

SAMPLE REQUIREMENTS

The samples received may need preparation, or may be prepared by the client (ready as received), or prepared by a different company. Thus, unless the sample is specifically defined as dry, the sample needs to be dried at 60°C. Some samples may also require crushing, splitting and/or milling depending on the package selected by the client and the type of material to be analyzed.

There are no holding times; however there is the possibility of sulfide oxidation (sample has been received already prepared but the sample is hard). The minimum amount of sample required is 0.5g.

QUALITY CONTROL

Reagent Blank: is run randomly once in every group of up to 30 samples.

QC Solutions: are run at the beginning and end of the instrument data acquisition and also run every 20 samples for Calibration Verification.

Certified Reference Materials (CRM): a reference materials is used to verify digestion conditions. A certified reference material must be weighed at least every 20 samples or once per digestion set.

Replicates: every 20 samples or once per digestion set a sample is chosen at random and weighed and digested in replicate.

REPORTING

The analyst reviews the results ensuring the blanks, certified reference materials, QC and replicates satisfy acceptance criteria. Data is transferred into the LIMS system by the analyst and the Lab Supervisor or General Manager authorizes the release to the customer. The results are reported in either weight % or mg/L, with a maximum of six significant figures (3 or 4 decimal places depending on the element). All data is kept with each file folder containing the COC and all relevant documentation.

51 Elements

Ag	Ni
Al	P
As	Pb
Au*	Rb
B	Re
Ba	S
Be	Sb
Bi	Sc
Ca	Se
Cd	Sn
Ce	Sr
Co	Ta
Cr	Te

Cs	Th
Cu	Ti
Fe	Tl
Ga	U
Ge	V
Hf	W
Hg	Y
In	Zn
K	Zr
La	
Li	
Mg	
Mn	
Mo	
Na	
Nb	

*** Please note Gold detection is only suitable for exploration purposes**