DIAMOND DRILLING ASSESSMENT REPORT

ON THE

DORADO ZONE

Lat. 50° 38' N, Long. 120° 31' 30" W
UTM Zone 10, 5612000 N, 674500 E

KAMLOOPS MINING DIVISION

For

Lakewood Mining Company Limited

And

Green Valley Mines Inc.

Operators

By

Joseph E.L. Lindinger, P.Geo.

April 18, 2008

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Summary

From late October to early December 2007 Lakewood Mining Ltd. under the direction of Mr., Charles R. Boitard attempted 5 and completed 2 diamond drill holes on the Dorado Target area 15 kilometers southwest of Kamloops, British Columbia.

The claims upon which the 2007 work program has been completed has been held by Mr. Charles Boitard for more than the last decade are and are under option to Lakewood Mining Ltd. 50% and Green Valley Mines Inc. (50%). Lakewood Mining Co. Ltd. and Green Valley Mines Inc. have completed many different geophysical and geochemical surveys as well as drilling programs on various areas of the claims.

The claims lie within the eastern alkali volcanic and sedimentary assemblage of the Nicola Group subterrane of the Upper Triassic to lower Jurassic volcanic island arc Quesnel Terrane. This arc and isolated segments of the intervening oceanic rocks and associated sediments were obducted onto the western edge of the north American continent during the Mid Jurassic. The eastern Nicola Group volcanics were intruded by numerous coeval generally alkalic stocks plugs and small batholiths that often host copper+/-gold porphyry and related skarn and replacement deposits. These deposits also can host appreciable silver, magnetite, palladium and platinum. The nearest such intrusive body on the claims partially underlies the eastern portions of the claims is the Iron Mask Batholith. Smaller dioritic dykes and sills also occur on the claims. The Iron Mask Batholith hosts several economic and numerous subeconomic copper-gold porphyry deposits, prospects and showings. Such deposits are the primary exploration targets on the claims.

Overlying the Nicola group rocks are early Tertiary Kamloops Group lacustrine and fluvial sediments which in turn are intruded and partially overlain along with Nicola Group rock by subareal intermediate to felsic Kamloops Group Volcanics. Numerous leucocratic felsic plugs, dykes, sills and possibly flows occur on the claims. These appear to have generated small or large very low grade epithermal and shear zone associated gold-+/-copper+/-arsenic+/-antimony deposits. These zones have possibly generated the large gold in soil MMI anomalies on the property that are a current focus of exploration.

Hole D07-04 was collared to test a buried IP anomaly. A short zone of disseminated pyrite mineralization with more extensive shear associated very weak native copper was intersected at the target depth. The pyrite appears to be associated with Tertiary intrusive and associated alteration. Holes D07-05, 6, 7 and 8 were collared 1500 meters north of hole 4 and attempted to test a large MMI gold anomaly overlying and subparalleling the Cherry Creek fault. All but hole 8 were abandoned in clay altered rock. All holes intersected red and maroon basaltic and andesitic volcanic breccias and lesser lapilli tuffs, ash tuffs and flows. The Nicola volcanics and intrusives have been intruded by Eocene Kamloops Group quartz feldspar porphyry rhyolite dykes and sills. The Kamloops group rocks appear to have accompanying silicic, potassic, albitic ankeritic and argillic clay alteration often accompanied by up to 10% fine grained pyrite with arsenopyrite. The alteration and mineralization appear to have been deposited by weak to moderate epithermal and shear zone associated hydrothermal cells driven by the Tertiary intrusives. Samples taken of mineralized material returned locally moderately anomalous arsenic with sporadic very weakly anomalous gold values. No alteration or mineralization associated

with an alkalic porphyry copper deposit was observed, or if so was overprinted by the strong Tertiary alteration.

Some documented gold showings associated with Kamloops groups extrusive rocks occur near the 2007 exploration areas. It is likely that the MMI anomalies are generated from hydrothermal remobilization along deep seated structures related to the Cherry Creek fault resulting in weak concentrations of metals sourced from the mafic metal rich Nicola volcanics.

To test for additional Tertiary volcanic associated gold mineralization, a preliminary $400,000 surface exploration program consisting of geological mapping and drilling is proposed. Further expenditures would be contingent on positive exploration results. Additional studies should continue to test for possible Afton style copper-gold mineralization.

Introduction

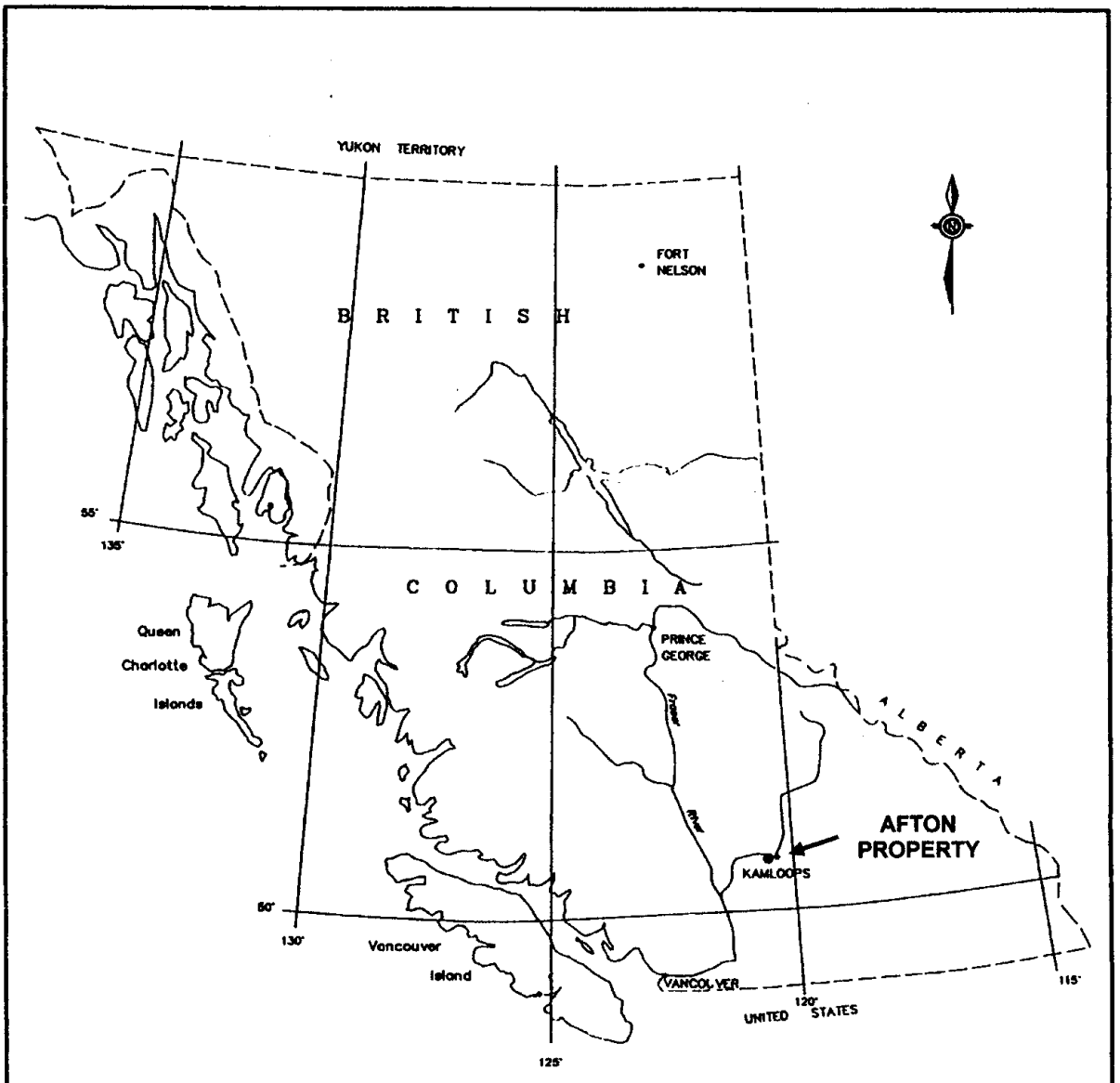
This report has been written at the request of Mr. Charles Boitard, President of Lakewood Mining Ltd and Green Valley Mines Inc. to discuss exploration findings from a diamond drill program completed in late 2007. The drill program was designed to test an area hosting an induced polarization anomaly and a second area hosting a strong and large gold-mobile metal ion (MMI) anomaly that could be a signature to one or more epithermal or shear hosted Tertiary aged gold bearing zones that is also associated with small erratically located IP anomalies.

Location and Access

The location of the Claims is centered on latitude Lat. 50^{O} 38' N, Long. 120^{O} 31' 30" W UTM 56 100 030 N, 678 000 E (Figures 1 and 2), in the Kamloops Mining Division, British Columbia, Canada. The claims making up the group are centered approximately 15 kilometers south southwest of Kamloops, B.C (map sheet 92-1/9 and 92-1/10), and south of the Afton Mine. Access to the northeast part of the claims is by range roads originating from the Lac Le Jeune Road near the Iron Mask industrial park. Access to the western parts of the claims is best via the Greenstone Mountain road and various range roads that transect various areas of the claims.

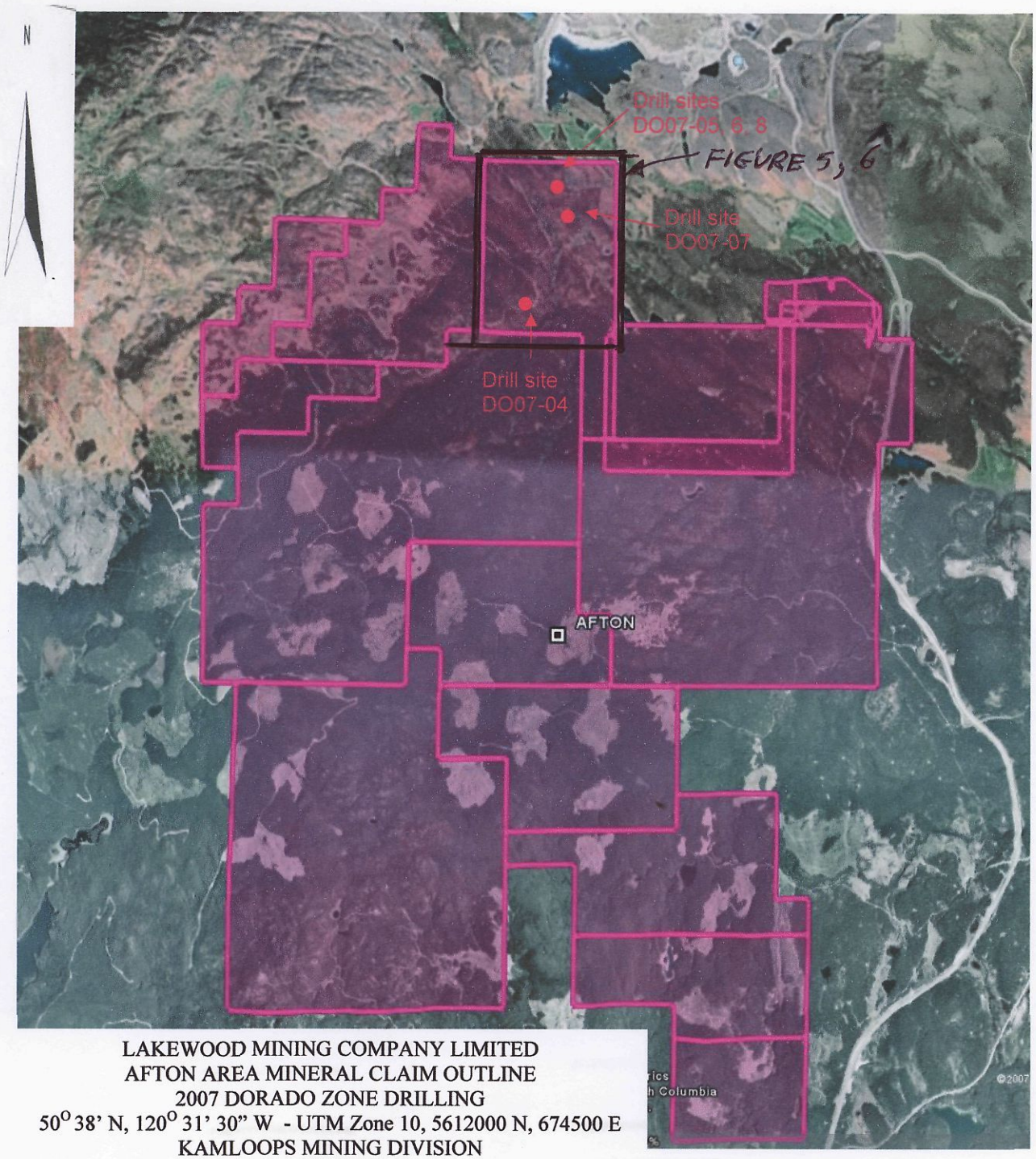
Physiography

The claims covers the north east slope with moderate to steep north sloping hills of forested and grassed open rangeland of Greenstone Mountain and parts of upper Cherry Creek and the adjoining lands towards the northeast towards the Coquihalla highway #5. The claims extend 11 to 20 kilometers southwest of Kamloops (Figure 2). The lower areas are underlain by deep and extensive glacial till, subsequently much of the claim area has very poor outcrop exposure. The northeast portions on the slopes of Greenstone Mtn. have sporadic outcrop totalling about 5% and much thinner till cover. The local elevations range from 750 metres to 1525 metres above sea level and the vegetation consists of previously logged dense forests of fir and (dying) pine interspersed with open grassland areas. There are many small creeks and drainage systems across the claims.

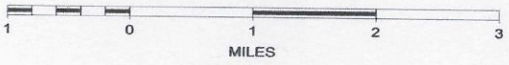


LAKEWOOD MINING COMPANY LIMITED
GREEN VALLEY MINES INC.
AFTON AREA MINERAL CLAIMS

Lat. 50° 38' N, Long. 120° 31' 30" W
UTM Zone 10, 5612000 N, 674500 E
KAMLOOPS MINING DIVISION
FIGURE 1 - LOCATION MAP



LAKEWOOD MINING COMPANY LIMITED
AFTON AREA MINERAL CLAIM OUTLINE
2007 DORADO ZONE DRILLING
50° 38' N, 120° 31' 30" W - UTM Zone 10, 5612000 N, 674500 E
KAMLOOPS MINING DIVISION
Figure 2 –
Data overlain on Google Earth background



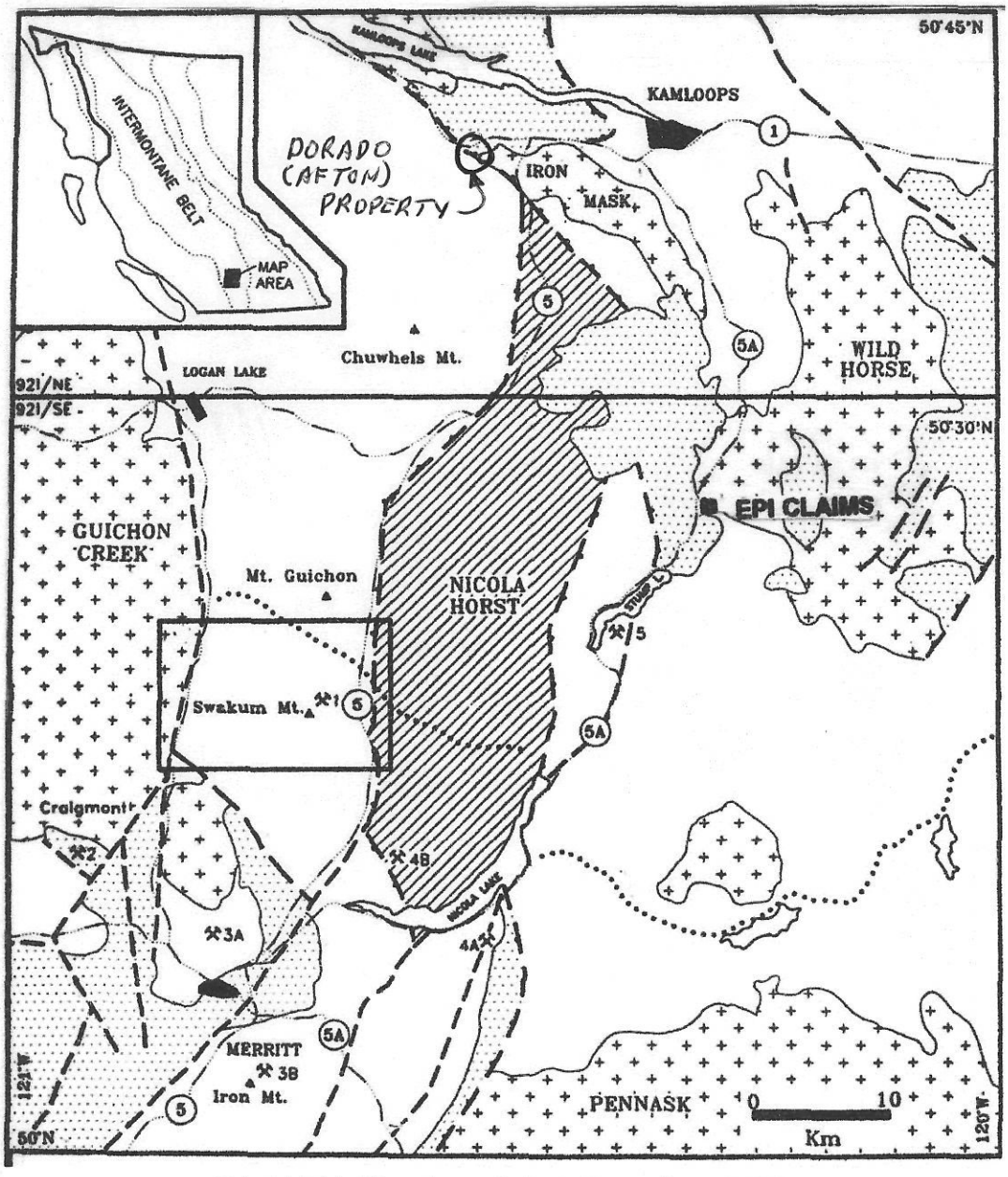
Mineral Tenure

The mineral claims are held 100% by Charles Roger Boitard of Vancouver under option to Lakewood Mining Company Limited (50%) and Green Valley Mine Incorporated (50%). The Claims for which work has been applied on for assessment purposes of the work that this report documents are in bold with the expiry date asterisked in Table 1 and were filed by Mr. Boitard with the Ministry of Energy and Mines on under Event Numbers 4191039, 4191041, 4191042 filed January 19, 2008.

TABLE 1 Mineral Tenure

Tenure Number	Tenure Type	Claim Name	Owner	Map Number	Good To Date*	Area	Tag Number
218587	Mineral	CAMP	102688 (100%)	092I068	2011/jun/13	500.0	39360
396557	Mineral	MONARCH	102688 (100%)	092I068	2011/sep/24	150.0	235034
515335	Mineral		102688 (100%)	092I	2011/mar/08	1416.58	
515339	Mineral		102688 (100%)	092I	2011/jan/21	430.883	
516119	Mineral		102688 (100%)	092I	2012/aug/01	471.51	
570405	Mineral	VIC 2	102688 (100%)	092I	2012/nov/21	492.601	
570406	Mineral	VIC 3	102688 (100%)	092I	2012/nov/21	492.745	
570407	Mineral	VIC 4	102688 (100%)	092I	2012/nov/21	369.659	
571139	Mineral	VIC 5	102688 (100%)	092I	2012/dec/01	246.507	

* With application and acceptance for assessment credits of the exploration work this report documents.



LAKEWOOD MINING COMPANY LIMITED
Lat. 50° 38'N, Long. 120° 31' 20"W
UTM Zone 10, 5612000 N, 674500 E
KAMLOOPS MINING DIVISION
Figure 3 – REGIONAL GEOLOGY

Map from Moore and Meyers, 1990

Regional Geology

The most important lithology regionally is the Quesnel Terrane (Quesnellia), an extinct volcanic arc obducted onto the west coast of ancestral North America during the Jurassic. The Quesnel Terrane (Figure 4) extends from north of the Toodogone area to south of the United States border. The southern part of Quesnellia is called the Nicola belt.

Kwong, Page 3, Summarizes the regional geology of the area (Figure 3).

> "The Nicola belt extends from south of Kamloops Lake 200 kilometres to the International Boundary. The most important pre- Tertiary rocks in this belt are Late Triassic volcanic and sedimentary rocks of the Nicola Group. The Nicola belt is divided into a series of narrow northerly trending blocks by several large, high-angle, northerly trending faults These faults are interpreted to be basement structures which controlled the distribution of volcanic centres and flanking sedimentary basins (Preto, 1977). Preto et at. (1979) identified four groups of major plutonic events in the belt. They are characterized by the ages of 200 million years (Ma), 160 Ma, 100 Ma, and 50-70 Ma respectively. The Iron Mask batholith is one of the larger alkaline plutons of the 200-Ma age group. It is situated along the southwest side of a regional northwest-trending fracture zone and is itself cut by numerous northwesterly faults. Northcote (1976) and Preto (1977) suggested that the batholith and other alkaline plutons in the same group are likely centres of Nicola volcanism."

Local Geology

Kwong, Page 3, discusses the general local geology (Figure 4).

> "On the southwestern flank of the Iron Mask Pluton, well indurated, massive and bedded tuff, breccia, and interbedded flows and flow breccia are prominent. All of these rocks are weakly metamorphosed and most of them show a fairly uniform green-grey colour. ..."

Intruding the Nicola Volcanics north east of the Wood group is the coeval multiphased alkalic Iron Mask Pluton.

The period to the early Tertiary was primarily one of gradual erosion and several changes in regional tectonics including the docking of the Quesnel Terrane onto ancestral North America. Kwong, Page 5 describes the Eocene Kamloops Group.

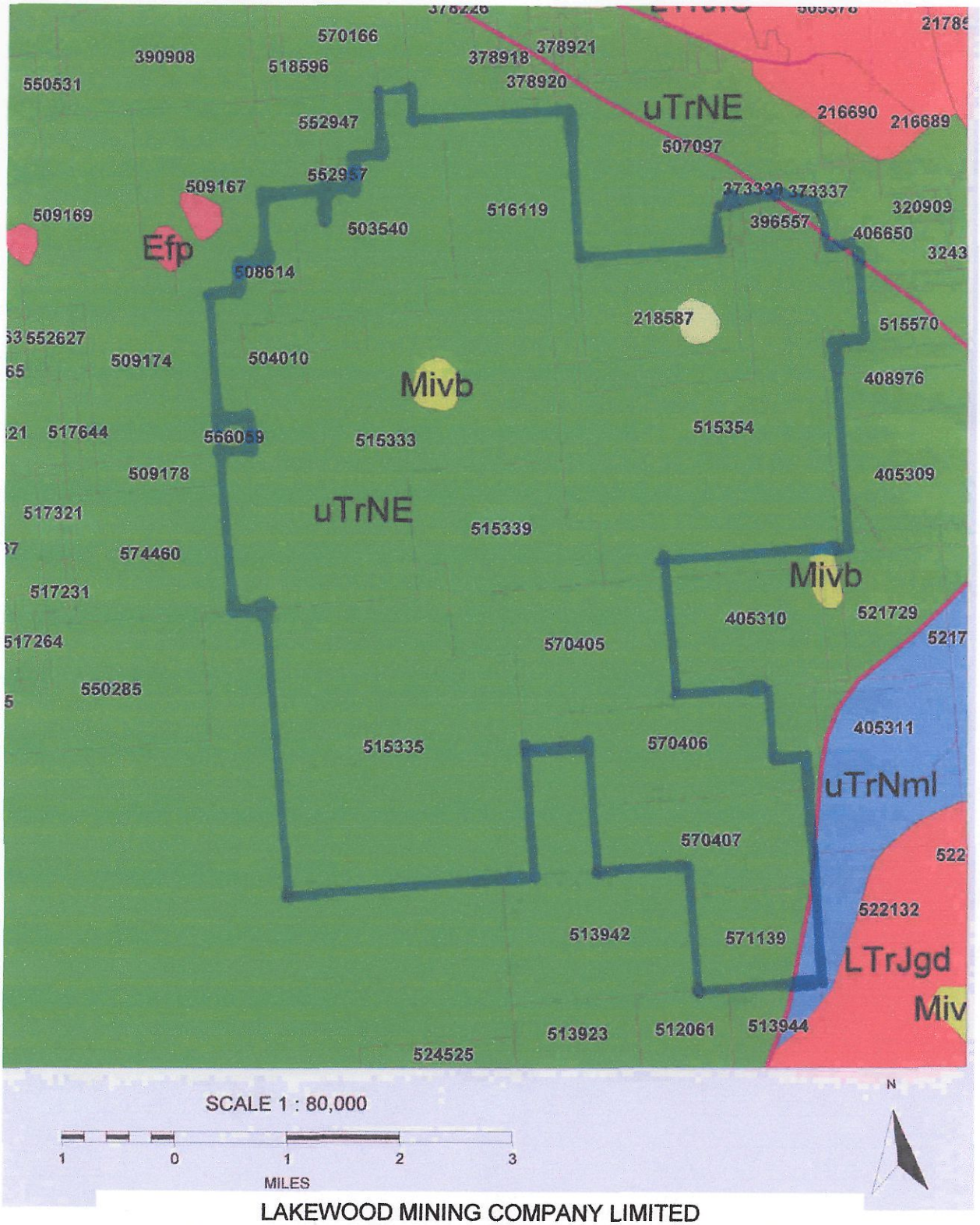
> "Early Tertiary sedimentary and volcanic rocks of the Kamloops Group unconformably overlie the Nicola rocks and the Iron Mask batholith. These include tuffaceous sandstone, siltstone, and shale with minor conglomerate, as well as basaltic to andesitic flows and agglomerates with minor dacite, latite, and trachyte. The Iron Mask pluton and the Cherry Creek pluton are separated by a thick sequence of Kamloops Group rocks occupying what appears to be a graben

structure resulting from renewed fault movement around the margins of the plutons during Paleocene or Early Eocene time. The geology of these rocks has been described in detail by Ewing (1982)."

BEDROCK GEOLOGICAL LEGEND for Figure 4

Tertiary		
Miocene		
	Mivb	Vesicular Basalt
Eocene		
	Efp	Feldspar Porphyry
Mesozoic		
Late Triassic-Jurassic		
	LTrJgd	Granodiorite (Iron Mask Batholith)
Upper Triassic		
	uTrNE	Eastern Group Nicola Volcanics undivided
	uTrNml	Nicola Volcanics metamorphic

FIGURE 4 – LOCAL GEOLOGY AND CLAIMS



LAKEWOOD MINING COMPANY LIMITED
AFTON AREA LOCAL GEOLOGY AND MINERAL CLAIMS
Lat. 50° 38'N, Long. 120° 31' 20"E
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KAMLOOPS MINING DIVISION

The area of the late 2007 drilling straddles the subregional northwest trending Cherry Creek fault, a deep seated long lived structure that has controlled or influenced the emplacement of several post Mesozoic intrusive bodies. These include Eocene Kamloops group volcanics and possibly Miocene basalts. The Kamloops group intrusives (Unit Efp) in particular occur as several feldspar +/- quartz phyric felsic dykes, plugs and possibly extrusive flow domes along the south side of the structure northwest on and southeast of the Claims. In several areas these appear to be associated with weak epithermal style gold mineralization such as at Ned Roberts Hill several kilometers immediately northwest of the claims. There are also unsubstantiated gold occurrences in rhyolite flow domes? immediately south of the exploration area 1 kilometer south of the Inks Lake interchange (Lindinger 1995)

2007 Drill Program

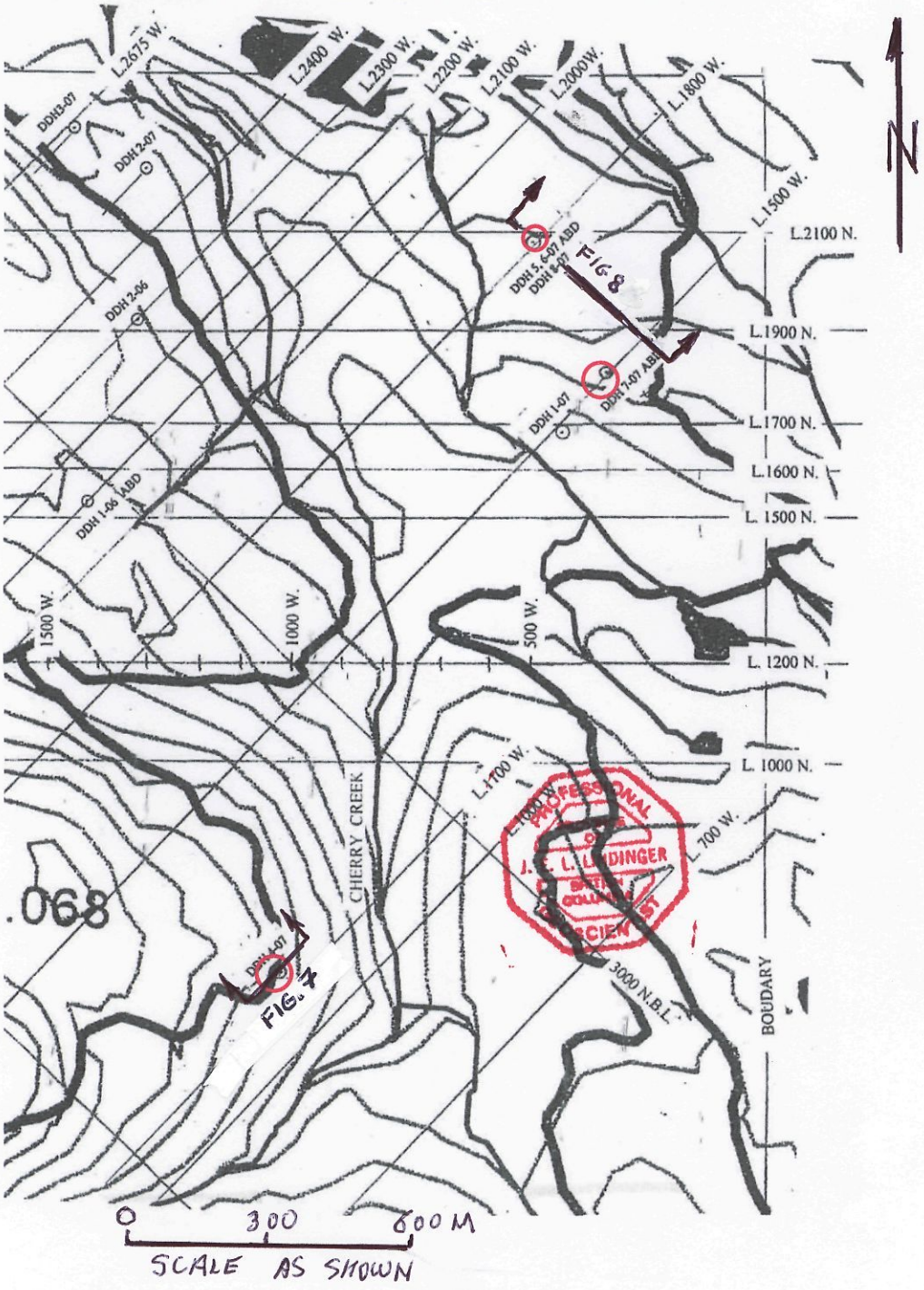
All of the following drill holes were collared on claim 516119 at the north central part of the claims.

Hole D07-04 was collared at UTM location ZONE10, 5612500 N, 674200 E. The hole was collared about 1 kilometer along a range road bearing south from the 7.4 km point on the Greenstone Mountain Road. The hole targeted an IP anomaly at least 80 meters deep. The hole intersected massive and sheared Nicola volcanics and small erratically located Tertiary felsic dykes. Most core contacts were about 35 degree suggesting moderate dip to faults and assumedly northwest striking northeast dipping. At the IP target weak erratically disseminated and fracture associated pyrite was intersected. The mineralization was associated with Tertiary Kamloops Group felsic dykes. Trace native copper was intersected in many shears. No significant mineralization was noted and no samples were taken.

Drill hole D07-05 was collared 1.5 kilometers north of hole D07-04 into an large Mobile Metal Ion (MMI) gold anomaly and is at the northeast end of a shallow apparently southwest dipping IP anomaly near the north central portion of claim 516119 . The hole was abandoned in strongly clay altered rock at a depth of 45 meters.

Hole D07-06 was collared about 5 meters east of hole 5 and was drilled to 50.6 meters and abandoned in similar rock as hole 5. 37.5 meters of casing were driven. The entire hole was sampled.

FIGURE 5 – DORADO ZONE 2007 DRILLING PLAN



Hole D07-07 was collared 310 meters east northeast of Hole 6. The hole was targeted to intersect a strong IP anomaly beginning at about 75 meters depth. After 6 meters of overburden strongly propylitic and clay altered sheared Nicola volcanics were intersected. This hole was drilled to 34.1 meters and abandoned. The entire hole was very strongly ankerite-clay altered with two "felsic or siliceous vein-dykes intruding sheared Nicola Breccias. The entire hole was sampled.

Hole D07-08 was collared at the same location as D07-06 and was drilled vertically to 331.3 meters. The entire hole encountered very strongly propylitically altered and sheared Nicola volcanics and possibly intrusives that have been intruded by several coeval phases of felsic Tertiary Kamloops Group intrusives. There does not appear to be a noticeable zoning trend to alteration or mineralization related to the felsic intrusives from shallow to deep dykes encountered.

All core is stored at 680 Dairy Road, Kamloops, B.C. V2B-8N5

Core Recovery, Sampling and Analytical Procedures

The core from all holes except hole D07-05 were washed, and the footage blocks were converted to metric. The core was examined for lost or ground segments and misplaced pieces. Sections of ground core and lost segments were noted in a geotech log. Also measured and noted were overall block to block recovery and RQD. Then to core was imaged. The core was geologically logged and sections selected for analyses were chosen. The sections of core selected for sampling were done so by the writer in the case of Holes D07-04, 5, 6, and 7 and in the case of hole D07-08 by Michael Cathro, P.Geo. The selected samples were split by experienced Geotechs Ken Dillabough and Brandon Barker. A conventional manual core splitter was used. Field standards and blanks were inserted approximately every 25 samples. The core samples for hole 6 and 7 were delivered via Greyhound Courier to Acme Analytical Laboratories in Vancouver. The core samples for hole 8 was delivered directly to Ecotech Laboratories Ltd. in Kamloops. All core samples were dried, then crushed to -6 mm from which a 250 gram portion was pulverized to 80% minus 200 mesh. Sub samples were taken from the pulp, a 15 to 30 gram subsample for gold analyses with AA finish, and a 5 gram subsample for conventional 28 or 34 element ICP multielement analyses. Analytical results are detailed in Appendix 1

Results

Holes 4 and 5 were not sampled. Hole D07-06 did not return any anomalous economic elements.

Hole D07-07 returned a 1.53 meter interval of anomalous arsenic (538 ppm) associated with an ivory ankerite or intensely ankeritized cryptocrystalline dacite vein/dyke that was at about 30 deg. to C.A.. The interval had a strong ankerite and minor silica overprint. Silica also occurs as minute tensional veinlet swarm in fragments. The interval was not anomalous for any other economic elements. The IP anomaly at depth remains untested.

Hole C07-08 returned several individual intervals reporting anomalous arsenic from 97.6 to 188.1 meters and one 36.6 meter zone from 197.21 to 233.78 m that averaged 394 ppm arsenic. The 36.6 meter zone was also reported weakly to very weakly anomalous antimony, copper and

gold. Several other weakly anomalous gold intervals also occur near to but not directly, associated with the arsenic. The arsenic values are directly associated with ragged aggregates and disseminations of fine grained pyrite-arsenopyrite mineralization. The gold mineralization appears to be associated with weak silicification and shear associated tension quartz veining often with trace very fine grained pyrite mineralization. The mineralization encountered was below the IP anomaly.

Conclusions

The immediate area surrounding hole D07-04 is not considered prospective for hosting an Afton style copper-gold deposit. The rocks encountered in hole 05 to 08 were sheared and altered Nicola volcanics within or near to the subregional northwest striking apparently steeply northeast dipping Cherry Creek fault. Weak gold mineralization associated with silicification and quartz-veining with weak pyrite mineralization is spatially associated with moderately anomalous arsenic. The alteration appears to be associated with Tertiary Kamloops Group felsic intrusives. The intrusives, alteration and mineralization are at least partially coeval with shearing. To date the mineralization encountered in holes D07- 06 to 08 partially explains the strong gold MMI anomaly.

Discussion

The Cherry Creek Fault Zone is known to host gold bearing structures and veins especially in the Ned Roberts Lake area about 1 kilometer NW of hole D07-05, 6, and 8, although traces of gold mineralization occur over a much larger area along the fault zone. The mineralization is hosted by silicified zones and shears and small quartz veins within larger carbonate and clay alteration zones. Gold mineralized veins often host very fine grained pyrite. Stronger pyrite mineralization within intense ankerite? altered zones report moderately anomalous arsenic, and weakly anomalous copper and antimony. The alteration and related mineralization appear to be associated with Tertiary Kamloops Group feldspar phyric felsic dykes and small plugs. Hole D06-02 was collared 900 meters due west of Hole D07-05 intersected a narrow 4 meter drill width intersection grading 0.55% copper with accompanying anomalous gold and bismuth in a style similar shear and alteration zone. The higher copper and bismuth here may be due to hole intersecting a deeper part of a similar mineralized system than that intersected in further east in Holes 5 to 8 with reported lower copper and gold values but higher arsenic and antimony.

TABLE 2

2007 - 2008 DORADO EXPENSE SUMMARY

DATES	EXPENSE ITEM	DAYS or HOURS	RATE		COST
Sep 14-Dec 28, 2007	Lindinger P.Geo. Geological supervision. core logging, drill site supervision. Hours	60.0	$ 100.00	$	6,000.00
Jan 1 - Feb 6, 2008	Lindinger P.Geo. Geological supervision. core logging, drill site supervision. Hours	3.0	$ 100.00	$	300.00
Sept 14-Oct 1, 07	Ken Dillabough - geoteching core (hours)	29.0	$ 30.00	$	870.00
Jan 1 - Mar 31, 2008	B. Barker - geoteching core (hours)	96.0	$ 30.00	$	2,880.00
Sep 14-Dec 28, 2007	Core processing facility (days)	80.0	$ 10.00	$	800.00
Jan 1 - Mar 31, 2008	Core processing facility (heated) (days)	70.0	$ 25.00	$	1,750.00
4X4 pickup	Usage days	14.8	$ 75.00	$	1,109.40
Mike Cathro, P. Geo	Core Logging (hours)	23.0	$ 100.00	$	2,300.00
Mike Cathro, P. Geo	Vehicle rental			$	36.75
Sept 14-Oct 1, 07	LNB Construction (bulldozer) (hours)	62.4	$ 150.00	$	9,360.50
Sept 14-Oct 1, 07	Full Force Drilling Ltd.	Mobilization		$	5,000.00
	Full Force Drilling Ltd.	341.2	meters	$	37,691.72
	Full Force Drilling Ltd.	hole stabilization		$	53,886.42
	Full Force Drilling Ltd.	consumables		$	32,124.51
Nov 27 - Dec 05, 2007	Frontier Diamond Drilling Ltd.	Mobilization		$	2,000.00
	Frontier Diamond Drilling Ltd.	329.5	meters	$	29,637.00
	Frontier Diamond Drilling Ltd.	hole stabilization		$	6,820.00
	Frontier Diamond Drilling Ltd.	consumables		$	6,775.00
	Frontier Diamond Drilling Ltd.	water truck		$	15,470.00
13-Sep-07	Caribou Chilcotin helicopters. Mobe in water pump to Cherry Creek)	0.5	$ 1,882.56	$	941.28
16-Sep-07	Caribou Chilcotin helicopters. Demobe in water pump from Cherry Creek)	0.5	$ 1,882.56	$	941.28
	Acme labs analyses			$	639.10
April1-10, 2005	Report			$	3,500.00
TOTAL APPLIED FOR ASSESSMENT				$	220,832.96

Recommendations

The target of Hole D07-07 should be re attempted to get to the target depth. The anomalous arsenic within strongly ankeritized and carbonate veined-flooded rock some 50 to 100 meters above a strong IP anomaly is an encouraging sign indicating possible upper reaches of a possible precious metal hydrothermal system.

The presence in Hole 8 of a pyrite-arsenopyrite mineralized zone from 197.2 to 233.8 M that returned anomalous arsenic with weaker anomalous antimony, copper and gold with associated silicification that overprints extensive carbonate alteration may suggest that the hole intersected the upper levels of a gold+/- copper mineralized system similar to and hopefully larger than that intersected in Hole 06-02, 900 meters to the west. Erratically anomalous gold and arsenic in the rock overlying the larger zone are also suggestive of the upper levels of a system.

The deep overburden and wide spaced IP and MMI surveys in the area makes structural interpretation difficult. The dominant structural trend is NW-SE with and inferred steep NE dip for the Cherry Creek fault zone. However Whittles 1990, 3 and 5 indicates that in the Ned Roberts Lake area 2 km NW of the 2007 drilling area the north trending faults may be also control alteration and possible mineralization.

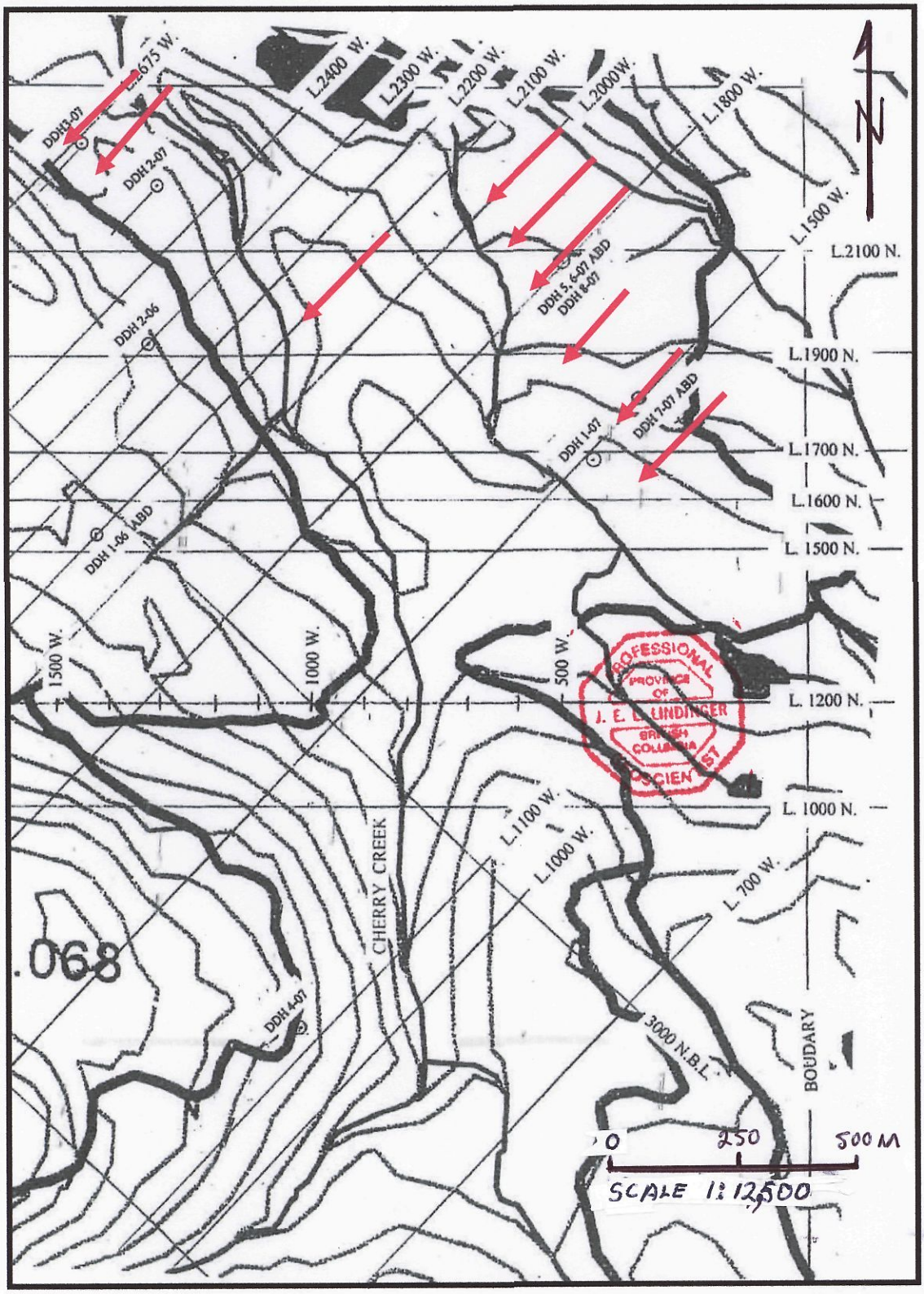
The author recommends a north west trending fence of 5 or 6 drill holes collared at 150 meter intervals beginning 100 meters north east of hole D07-07 and 150 meters northeast of D07-08 to the north property boundary. Each hole would be 300 meters long drill and drilled to the southwest at 50 degrees cross cutting the inferred NW strike and NE dip of the target zone, as well as intersecting possible north trending zones at a more favourable angle. Another hole should be collared 75 meters NW of hole D07-08 to intersect the IP anomaly that may host auriferous sulphide mineralization. At least 2 holes should be collared north and northwest of hole D06-02 to undercut the strong MMI gold in soil anomaly that extends north and northwest of hole D06-02 and hopefully stronger gold-copper mineralization that that intersected in hole 2.

Additional exploration expenditures in the MMI gold anomalous zone would be contingent on the success of this program. If the holes near the northwest corner of the claims encounters mineralization optioning the adjoining claim covering Ned Roberts Lake should be considered.

TABLE 3

RECOMMENDED EXPENDITURES				
EXPENSE ITEM	COST PER UNIT			
GEOTECH AND CORE SAMPLING				$ 20,000
GEOLOGICAL MAPPING SUPERVISION AND CORE LOGGING				$ 40,000
DIAMOND DRILLING 8 315 M HOLES 2500 M	2,500	$	150	$375,000
SAMPLING AVERAGE 2.5 M PER SAMPLE 1000 SAMPLES	1,000	$	35	$ 35,000
REPORT				$ 15,000
CONTINGENCY 10%				$ 55,000
TOTAL BUDGET				$540,000

FIGURE 6 - RECOMMENDED DRILLING PLAN



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Certificate of Independent Qualified Person:

I, Joseph Eugene Leopold Lindinger, P.Geo. am a consulting geoscientist residing at 680 dairy road, Kamloops, B.C. V2B-8N5

2. I am Registered member as a Professional Geoscientist of the Association of Professional Engineers and Geoscientists of the Province of British Columbia (1992), member #19155.

3. I am a graduate of the University of the University of Waterloo, Ontario with a Bachelor of Sciences (BSc) in Honours Earth Sciences, and have practiced my profession continuously since that time.

4. Since 1975, I have been involved in mineral exploration for gold, copper, zinc, lead and silver, and Uranium, in British Columbia, Ontario, Labrador, Nunavut, Northwest Territory, Yukon Territory, Nevada (USA) and Mexico.

5. As a result of my education, professional experience and professional qualifications, I am a qualified person as defined in National Instrument 43-101 for the mineral deposits being explored for on the mineral property that is the subject of this report.

6. Since 1992 I have been a Professional Geoscientist operating a geoscience consulting practice based in Kamloops, British Columbia.

7. I first visited the property on July 8, 2004, on behalf of Lakewood Mining Company Limited, and have revisited the property later several times to site drill holes, monitor drill progress, and retrieve or log and sample the drill core on site including the drilling that this report documents

8. I prepared this report based on historical and new exploration data generated by the 2007exploration programs.

9. In the disclosure of information relating to permitting, legal title, action, and related issues, I have relied on information from the Ministry of Energy Mines and Petroleum Resources, Mineral Titles Division and the I disclaim responsibility for the accuracy of such information.

10. I am not aware of any material fact or material change with respect to the subject matter of this report that is not reflected in this report, the omission to disclose which would make this report misleading.

11. I am independent of Lakewood Mining Company Limited and Green Valley Mines Incorporated and have no interest material or otherwise in the claims comprising the Wood Group.

12. I consent to the filing of this report with any stock exchange and other regulatory authority and any publication by them, including electronic publication in the public company

files on their websites accessible to the public, of the technical report so long as these publications do not provide conclusions different than this report documents.

Dated at Kamloops, British Columbia, this 18 day of April, 2008.

Signed "*Joseph E.L. Lindinger, P.Geo*"
Joseph E.L. Lindinger, P.Geo.
Consulting Geoscientist

APPENDIX 1
ANALYTICAL RESULTS

ACME ANALYTICAL LABORATORIES LTD. Final Report
Client: Lakewood Mining Co. Ltd.
File Create 1-Feb-08
Job Numb VAN07003127
Number of 18
Project: DORADA
Shipment I 1-Jul
P.O. Number:
Received: 9-Oct-07

	Method	WGHT	3A	1D	1D	1D	1D	1D	1D	1D	1D	1D	1D	1D	1D	1D	1D	1D	1D
	Analyte	WT	Au	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr		
	Unit	KG	PPB	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM		
	MDL	0.01	0.5	1	2	3	1	0.3	1	1	2	0.01	2	8	2	2	1		
Sample	Type																		
132001	Drill Core	7.8	1.5	<1	73	<3	47	0.4	3	10	991	2.85	8	<8	<2	3	147		
132002	Drill Core	1.5	1.3	<1	44	<3	39	0.3	3	12	1632	2.68	9	<8	<2	<2	165		
132003	Drill Core	4.9	0.8	<1	44	<3	50	0.4	3	13	1620	3.12	6	<8	<2	<2	162		
132004	Drill Core	3.6	4.6	<1	40	<3	64	<0.3	3	14	1355	3.36	6	<8	<2	3	154		
132005	Drill Core	7.2	2.5	<1	42	3	62	0.4	3	14	1093	3.78	5	<8	<2	2	128		
132006	Drill Core	6.5	3.5	<1	57	<3	45	0.4	2	14	1689	3.14	9	<8	<2	<2	172		
132007	Drill Core	5.5	2.6	<1	67	4	49	<0.3	5	14	1376	3.03	10	<8	<2	2	190		
132008	Drill Core	3.6	3.7	<1	110	<3	57	0.4	4	15	1015	3.8	6	<8	<2	2	217		
132009	Drill Core	7.1	1.3	<1	59	<3	53	0.5	3	17	1636	3.77	26	<8	<2	<2	217		
132010	Drill Core	6.4	2.7	<1	65	<3	71	0.4	3	18	1183	4.41	9	<8	<2	2	169		
132011	Drill Core	4.5	1.8	<1	91	<3	56	0.5	3	15	1128	3.42	15	<8	<2	3	161		
132012	Drill Core	4.3	1.8	<1	73	4	47	0.3	12	41	1968	4.41	538	<8	<2	<2	186		
132013	Drill Core	4.9	1.9	<1	80	<3	55	0.5	7	26	1420	3.3	86	<8	<2	<2	204		
132014	Drill Core	4.7	2.1	<1	65	<3	53	0.4	4	19	1338	3.23	28	<8	<2	<2	179		
132015	Drill Core	2.8	2.6	<1	63	10	45	0.3	3	14	1927	3.75	10	<8	<2	<2	198		
132016	Drill Core	2.5	1.5	<1	68	<3	57	0.4	4	19	1939	4.09	15	<8	<2	<2	203		
132017	Drill Core	4.6	2.9	<1	90	<3	79	0.5	5	19	1402	3.44	42	<8	<2	2	198		
132018	Rock Pulp		843.5	735	4392	65	78	36.9	20	40	634	2.1	1244	<8	<2	<2	127		
Pulp Duplicates																			
132013	Drill Core	4.9	1.9	<1	80	<3	55	0.5	7	26	1420	3.3	86	<8	<2	<2	204		
132013	REP		2.9																
132014	Drill Core	4.7	2.1	<1	65	<3	53	0.4	4	19	1338	3.23	28	<8	<2	<2	179		
132014	REP			<1	65	24	51	0.3	4	18	1335	3.24	29	<8	<2	<2	177		
Reference Materials																			
STD OXD	STD		356.8																
STD OXD	STD		359.9																
STD DS7	STD			19	105	63	390	1.1	52	8	582	2.28	48	<8	<2	5	65		
STD DS7	STD			23	101	73	422	1.1	58	8	666	2.53	55	<8	<2	5	78		
BLK	BLK		<0.5																
BLK	BLK			<1	<2	<3	<1	<0.3	<1	<1	<2	<0.01	<2	<8	<2	<2	<1		
Prep Wash																			
G1	Prep Blank	<0.01	<0.5	<1	<2	<3	44	<0.3	5	3	542	1.86	<2	<8	<2	4	52		
G1	Prep Blank	<0.01	1.1	<1	<2	<3	44	<0.3	4	3	525	1.81	2	<8	<2	5	46		

1D Cd PPM	1D Sb PPM	1D Bi PPM	1D V PPM	1D Ca %	1D P %	1D La PPM	1D Cr PPM	1D Mg %	1D Ba PPM	1D Ti %	1D B PPM	1D Al %	1D Na %	1D K %	1D W PPM	1D Tl PPM	1D Hg PPM
0.5	3	3	1	0.01	0.001	1	1	0.01	1	0.01	20	0.01	0.01	0.01	2	5	1
<0.5	<3	<3	82	3.91	0.132	12	3	1.08	28	<0.01	<20	1.57	0.23	0.09	<2	<5	<1
<0.5	<3	<3	76	7.24	0.113	11	2	2.09	18	<0.01	<20	0.95	0.07	0.06	<2	<5	<1
<0.5	<3	<3	93	8.16	0.115	11	1	2.73	85	<0.01	<20	0.94	0.07	0.1	<2	<5	<1
<0.5	<3	<3	84	4.83	0.159	14	2	1.15	279	<0.01	<20	1.31	0.1	0.11	<2	<5	<1
<0.5	<3	<3	96	4.01	0.16	14	2	1.04	25	0.02	<20	1.34	0.1	0.11	<2	7	<1
<0.5	<3	<3	66	9.43	0.119	11	<1	1.96	215	<0.01	<20	1.02	0.11	0.09	<2	<5	<1
<0.5	<3	<3	67	5.49	0.14	12	4	1.42	24	<0.01	<20	1.06	0.15	0.1	<2	<5	<1
<0.5	<3	<3	111	4.58	0.157	13	2	1.05	112	0.03	<20	2.19	0.19	0.15	2	<5	<1
<0.5	<3	<3	88	7.85	0.126	12	2	2.22	62	<0.01	<20	1.17	0.12	0.1	<2	<5	<1
<0.5	<3	<3	102	4.64	0.149	13	2	1.21	117	0.01	<20	2.21	0.17	0.13	<2	5	<1
<0.5	<3	<3	85	5.18	0.147	13	1	1.02	50	<0.01	<20	1.44	0.14	0.11	<2	<5	<1
<0.5	<3	<3	64	10.57	0.095	11	<1	2.79	84	<0.01	<20	0.8	0.11	0.08	<2	<5	<1
<0.5	<3	<3	79	6.22	0.141	12	2	1.24	90	<0.01	<20	1.66	0.18	0.1	<2	<5	<1
<0.5	<3	<3	81	7.19	0.133	12	1	1.7	49	<0.01	<20	1.2	0.12	0.1	<2	<5	<1
<0.5	<3	<3	77	10.45	0.102	11	<1	1.57	59	<0.01	<20	1.1	0.12	0.12	<2	<5	<1
<0.5	<3	<3	67	9.91	0.103	12	1	2.35	38	<0.01	<20	1.11	0.14	0.09	<2	<5	<1
<0.5	<3	<3	80	7.02	0.136	13	3	1.44	35	<0.01	<20	1.48	0.13	0.12	<2	<5	<1
0.6	73	31	17	3.43	0.051	7	175	0.2	64	0.02	22	0.71	0.04	0.12	15	<5	<1
<0.5	<3	<3	79	6.22	0.141	12	2	1.24	90	<0.01	<20	1.66	0.18	0.1	<2	<5	<1
<0.5	<3	<3	81	7.19	0.133	12	1	1.7	49	<0.01	<20	1.2	0.12	0.1	<2	<5	<1
<0.5	<3	<3	80	7.18	0.133	12	2	1.71	50	<0.01	<20	1.2	0.12	0.1	<2	<5	<1
5.5	5	3	78	0.9	0.071	11	180	0.99	382	0.1	41	0.94	0.08	0.42	3	13	<1
6.3	3	4	93	1.01	0.076	14	216	1.12	409	0.12	48	1.07	0.1	0.46	4	14	<1
<0.5	<3	<3	<1	<0.01	<0.001	<1	<1	<0.01	<1	<0.01	<20	<0.01	<0.01	<0.01	<2	<5	<1
<0.5	<3	<3	34	0.46	0.07	6	8	0.6	219	0.12	<20	0.98	0.07	0.52	2	6	<1
<0.5	<3	<3	33	0.42	0.069	6	9	0.57	217	0.11	<20	0.95	0.07	0.51	<2	9	<1

20-Mar

ECO TECH LABORATORY LTD.
10041 Dallas Drive
KAMLOOPS, B.C.
V2C 6T4

Phone: 250-573-5700
Fax : 250-573-4557

ICP CERTIFICATE OF ANALYSIS AK 2008- 0139

Lakewood Mining
680 Dairy Road
Kamloops, BC
V2B 8N5

Attention: Brandon Barker

No. of samples received: 84
Sample Type: Core
Project: Dorado
Shipment #: 08-01
Submitted by: Brandon Barker

Values in ppm unless otherwise reported

Et #.	Tag #	Au(ppb)	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
1	786051	10	<0.2	1.82	20	55	<5	2.95	<1	29	119	176	6.24	<10	3.11	929	<1	0.19	59	3210	6	<5	<20	175	0.10	<10	191	<10	10	60
2	786052	<5	<0.2	2.24	130	990	<5	3.58	<1	36	175	35	3.21	<10	4.30	511	<1	0.80	362	380	6	<5	<20	1911	0.02	<10	35	<10	16	33
3	786053	<5	<0.2	1.73	15	105	<5	4.49	<1	55	353	53	4.36	<10	8.60	589	<1	0.37	449	540	2	<5	<20	1277	0.02	<10	73	<10	3	30
4	786054	5	<0.2	1.51	15	105	<5	3.87	<1	54	321	50	4.08	<10	8.36	548	<1	0.39	452	520	4	<5	<20	385	0.02	<10	61	<10	3	31
5	786055	25	0.2	0.83	10	115	<5	4.82	<1	50	208	48	3.87	<10	6.66	638	<1	0.32	443	490	6	<5	<20	423	<0.01	<10	47	<10	4	22
6	786056	<5	<0.2	0.65	15	245	<5	4.24	<1	57	207	92	4.58	<10	4.92	909	<1	0.35	463	700	6	<5	<20	342	<0.01	<10	54	<10	5	19
7	786057	<5	<0.2	0.43	10	105	<5	5.25	<1	59	236	53	4.81	<10	7.07	992	<1	0.32	480	210	6	<5	<20	336	<0.01	<10	37	<10	3	19
8	786058	<5	<0.2	1.24	10	80	<5	4.86	<1	44	316	45	3.83	<10	8.91	812	<1	0.35	447	530	4	<5	<20	426	<0.01	<10	41	<10	4	29
9	786059	5	<0.2	0.74	10	150	<5	4.98	<1	34	218	88	4.52	<10	3.67	786	<1	0.16	267	920	10	<5	<20	724	<0.01	<10	63	<10	7	27
10	786060	5	<0.2	0.54	70	50	<5	7.69	<1	46	337	34	3.75	<10	4.99	810	<1	0.09	424	190	10	5	<20	744	<0.01	<10	53	<10	3	29
11	786061	5	<0.2	0.58	5	60	<5	>10	<1	40	421	9	3.83	<10	7.47	655	<1	0.04	442	20	10	<5	<20	1056	<0.01	<10	61	<10	2	27
12	786062	30	<0.2	0.34	10	105	<5	8.07	<1	55	271	39	4.09	<10	6.14	786	<1	0.08	449	80	10	<5	<20	897	<0.01	<10	62	<10	3	36
13	786063	10	<0.2	0.20	<5	80	<5	8.95	<1	28	125	27	3.34	<10	5.32	685	<1	0.06	317	50	8	<5	<20	657	<0.01	<10	45	<10	3	27
14	786064	5	<0.2	0.15	<5	60	<5	8.96	<1	22	69	31	3.50	<10	5.05	710	<1	0.06	158	100	8	<5	<20	614	<0.01	<10	48	<10	4	22
15	786065	5	<0.2	0.26	25	85	<5	7.63	<1	31	118	45	3.53	<10	4.16	744	<1	0.07	373	170	8	<5	<20	544	<0.01	<10	43	<10	4	32
16	786066	10	<0.2	0.31	110	60	<5	7.36	<1	42	255	49	3.96	<10	4.82	980	<1	0.08	440	300	10	10	<20	538	<0.01	<10	45	<10	3	31
17	786067	25	<0.2	0.21	15	60	<5	9.12	<1	33	162	25	3.74	<10	5.07	851	<1	0.06	368	130	10	<5	<20	588	<0.01	<10	47	<10	4	29
18	786068	30	<0.2	0.17	<5	55	<5	9.87	<1	32	123	15	3.98	<10	5.80	798	<1	0.05	317	50	12	<5	<20	636	<0.01	<10	54	<10	4	32
19	786069	10	<0.2	0.20	<5	110	<5	6.99	<1	19	77	24	3.10	<10	3.69	656	2	0.04	176	200	8	<5	<20	532	<0.01	<10	38	<10	3	18
20	786070	10	<0.2	0.19	5	130	<5	4.74	<1	11	41	40	2.58	<10	2.07	519	5	0.04	63	510	6	<5	<20	469	<0.01	<10	27	<10	3	17
21	786071	10	<0.2	0.32	5	115	<5	7.31	<1	18	22	113	3.66	<10	3.21	904	<1	0.06	38	970	8	<5	<20	391	<0.01	<10	35	<10	5	31
22	786072	5	<0.2	0.54	10	180	<5	6.22	<1	20	44	96	4.37	<10	2.60	1131	<1	0.09	49	1600	8	<5	<20	253	<0.01	<10	30	<10	6	48
23	786073	25	<0.2	0.77	15	140	<5	4.87	<1	25	36	98	5.78	<10	1.95	888	<1	0.12	41	2260	10	<5	<20	177	<0.01	<10	53	<10	7	70
24	786074	25	<0.2	0.61	10	215	<5	4.83	<1	32	46	67	6.74	<10	1.71	975	<1	0.12	32	2700	10	<5	<20	152	<0.01	<10	60	<10	7	86
25	786075	540	0.6	0.72	2675	20	30	5.30	<1	91	10	116	3.39	<10	0.16	469	23	0.10	25	1420	18	5	<20	105	0.03	<10	26	<10	6	76

Et #.	Tag #	Au(ppb)	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
26	786076	<5	<0.2	0.06	<5	5	<5	0.03	<1	<1	<1	<1	0.21	<10	0.01	23	<1	<0.01	2	110	2	<5	<20	3	<0.01	<10	2	<10	<1	3
27	786077	60	<0.2	0.69	20	150	<5	6.44	<1	25	27	132	4.91	<10	1.44	766	<1	0.10	24	2690	8	5	<20	173	<0.01	<10	41	<10	6	62
28	786078	20	<0.2	0.64	5	230	<5	5.98	<1	21	30	113	4.34	<10	2.37	1127	<1	0.07	20	2070	10	<5	<20	218	<0.01	<10	36	<10	6	53
29	786079	5	<0.2	1.08	15	75	<5	5.37	<1	23	36	179	4.03	<10	1.25	630	<1	0.09	35	1300	6	5	<20	138	<0.01	<10	41	<10	4	54
30	786080	10	<0.2	1.14	20	70	<5	5.65	<1	24	38	62	4.17	<10	1.50	660	<1	0.10	27	1310	6	<5	<20	159	<0.01	<10	49	<10	6	46
31	786081	20	<0.2	1.51	20	65	<5	6.14	<1	24	50	81	5.24	<10	2.06	739	<1	0.10	26	1390	6	<5	<20	169	<0.01	<10	63	<10	6	51
32	786082	15	<0.2	1.99	20	80	<5	5.71	<1	30	69	110	4.56	<10	2.69	1058	<1	0.10	33	1310	4	<5	<20	187	0.02	<10	79	<10	7	52
33	786083	5	<0.2	1.70	190	65	<5	4.11	<1	41	54	159	5.30	<10	1.99	1032	<1	0.13	43	1470	6	20	<20	145	<0.01	<10	72	<10	7	55
34	786084	20	<0.2	2.04	50	50	<5	4.78	<1	27	60	127	5.12	<10	2.27	926	<1	0.14	35	1560	4	<5	<20	190	<0.01	<10	93	<10	7	52
35	786085	40	<0.2	0.96	30	55	<5	4.51	<1	26	42	153	5.26	<10	0.89	807	<1	0.13	32	1530	8	10	<20	112	<0.01	<10	39	<10	8	46
36	786086	10	<0.2	1.65	95	50	<5	3.37	<1	31	65	130	5.29	<10	1.88	916	<1	0.15	38	1590	12	5	<20	170	<0.01	<10	85	<10	7	66
37	786087	10	<0.2	1.67	375	80	<5	3.39	<1	38	53	134	5.33	<10	1.97	850	<1	0.13	39	1480	6	20	<20	148	<0.01	<10	79	<10	6	56
38	786088	5	<0.2	2.12	135	40	<5	5.21	<1	27	96	118	4.55	<10	2.32	714	<1	0.11	32	1210	4	5	<20	178	<0.01	<10	104	<10	4	51
39	786089	30	<0.2	1.54	60	50	<5	3.18	<1	32	61	131	5.51	<10	2.22	1106	<1	0.12	42	1520	6	<5	<20	147	<0.01	<10	74	<10	7	56
40	786090	30	0.2	1.87	335	40	<5	1.40	<1	35	40	182	6.05	<10	1.71	908	<1	0.17	30	3010	8	15	<20	115	<0.01	<10	123	<10	8	67
41	786091	15	<0.2	2.47	235	45	<5	1.31	<1	36	40	197	6.32	<10	1.89	1065	<1	0.22	31	3670	6	15	<20	171	<0.01	<10	150	<10	9	77
42	786092	10	<0.2	1.86	970	45	<5	1.34	<1	33	59	161	5.90	<10	1.81	846	<1	0.21	33	2280	6	75	<20	161	<0.01	<10	89	<10	5	56
43	786093	10	0.2	1.82	850	45	<5	1.31	<1	32	28	183	6.25	<10	1.54	699	<1	0.18	24	3430	8	55	<20	128	<0.01	<10	117	<10	5	64
44	786094	25	0.2	2.04	270	45	<5	1.50	<1	27	23	219	5.86	<10	1.51	825	<1	0.22	17	3880	6	15	<20	176	<0.01	<10	134	<10	7	71
45	786095	25	<0.2	1.65	330	45	<5	1.61	<1	29	18	199	6.27	<10	1.40	1043	<1	0.20	17	3860	6	15	<20	164	<0.01	<10	107	<10	10	72
46	786096	15	<0.2	1.79	510	40	<5	1.77	<1	28	30	179	5.63	<10	1.45	873	<1	0.22	19	3410	6	20	<20	189	<0.01	<10	100	<10	8	63
47	786097	5	<0.2	1.33	500	40	<5	1.30	<1	39	54	130	5.31	<10	1.32	925	<1	0.22	39	1520	6	20	<20	178	<0.01	<10	64	<10	6	51
48	786098	10	<0.2	0.88	155	45	<5	2.85	<1	46	49	100	5.46	<10	1.69	1571	<1	0.14	50	1290	8	5	<20	137	<0.01	<10	49	<10	9	48
49	786099	10	<0.2	0.86	45	40	<5	4.27	<1	29	30	110	4.52	<10	1.08	864	<1	0.15	31	1440	10	5	<20	139	<0.01	<10	50	<10	7	43
50	786100	565	0.6	0.71	2705	20	30	5.43	<1	93	10	117	3.45	<10	0.17	466	24	0.10	25	1430	16	5	<20	106	0.03	<10	25	<10	6	78
51	786101	5	<0.2	0.05	<5	10	<5	0.02	<1	<1	<1	<1	0.16	<10	0.01	24	<1	<0.01	2	80	2	<5	<20	4	<0.01	<10	2	<10	<1	1
52	786102	50	<0.2	0.88	90	50	<5	6.22	<1	23	46	85	4.94	<10	1.02	818	<1	0.12	27	1370	8	5	<20	138	<0.01	<10	38	<10	8	45
53	786103	15	<0.2	1.08	20	45	<5	7.94	<1	21	49	189	5.28	<10	0.98	673	<1	0.12	22	2190	8	<5	<20	139	<0.01	<10	50	<10	9	51
54	786104	15	<0.2	1.15	15	45	<5	6.94	<1	20	47	110	5.38	<10	1.15	758	<1	0.15	16	1870	6	<5	<20	172	<0.01	<10	60	<10	8	48
55	786105	10	<0.2	1.15	15	45	<5	6.24	<1	12	48	121	4.26	<10	0.90	690	<1	0.17	11	1860	10	<5	<20	191	<0.01	<10	62	<10	7	34
56	786106	5	<0.2	0.47	25	30	<5	5.70	<1	41	26	133	2.46	<10	0.75	494	<1	0.10	38	830	14	<5	<20	115	<0.01	<10	21	<10	15	51
57	786107	15	0.2	1.14	20	75	<5	8.51	<1	18	72	64	5.86	<10	1.72	930	<1	0.11	18	1600	8	<5	<20	173	<0.01	<10	60	<10	12	45
58	786108	35	<0.2	0.78	30	30	<5	4.94	<1	17	36	85	3.68	<10	0.79	530	<1	0.09	18	1040	8	<5	<20	110	<0.01	<10	33	<10	13	38
59	786109	5	<0.2	0.23	15	115	<5	0.18	<1	5	26	26	0.11	<10	0.09	23	<1	0.08	9	40	6	<5	<20	55	<0.01	<10	<1	<10	18	12
60	786110	<5	<0.2	0.21	10	160	<5	1.02	<1	8	18	7	0.44	<10	0.36	166	<1	0.07	9	110	14	<5	<20	61	<0.01	<10	2	<10	16	10
61	786111	10	<0.2	0.28	15	80	<5	5.73	<1	16	28	89	3.53	<10	2.15	943	<1	0.06	8	860	12	<5	<20	238	<0.01	<10	13	<10	10	26
62	786112	10	<0.2	0.69	25	150	<5	5.54	<1	15	35	52	4.53	<10	1.81	806	<1	0.08	12	1310	12	<5	<20	157	<0.01	<10	28	<10	9	38
63	786113	15	<0.2	1.22	15	410	<5	4.90	<1	23	90	133	5.06	<10	2.49	1036	<1	0.13	15	1390	12	<5	<20	188	<0.01	<10	109	<10	9	58
64	786114	20	<0.2	1.92	20	290	<5	1.23	<1	27	75	135	5.80	<10	1.73	736	<1	0.14	18	2350	8	<5	<20	108	<0.01	<10	92	<10	5	67
65	786115	20	<0.2	0.77	20	130	<5	6.82	<1	27	49	138	5.50	<10	3.04	1267	<1	0.12	14	1440	8	<5	<20	193	<0.01	<10	49	<10	9	54

Et #.	Tag #	Au(ppb)	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
66	786116	15	<0.2	0.54	15	55	<5	6.96	<1	18	60	37	5.10	<10	2.68	1070	<1	0.12	9	1450	8	<5	<20	221	<0.01	<10	64	<10	12	45
67	786117	10	<0.2	0.55	20	55	<5	5.79	<1	30	43	137	5.51	<10	2.20	991	<1	0.13	18	1640	10	<5	<20	187	<0.01	<10	44	<10	12	50
68	786118	25	<0.2	0.73	15	85	<5	4.01	<1	33	73	123	4.92	<10	1.80	1054	<1	0.11	21	1640	6	<5	<20	160	<0.01	<10	77	<10	7	58
69	786119	10	<0.2	0.53	30	170	<5	6.41	<1	17	28	99	5.13	<10	2.48	822	<1	0.11	9	1800	8	<5	<20	205	<0.01	<10	29	<10	7	40
70	786120	15	<0.2	0.48	35	525	<5	5.30	<1	19	47	103	5.21	<10	1.90	795	<1	0.10	11	2010	8	<5	<20	192	0.01	<10	42	<10	7	47
71	786121	20	<0.2	0.49	35	190	<5	3.92	<1	18	46	98	5.29	<10	1.42	689	<1	0.10	12	2000	14	<5	<20	140	0.02	<10	49	<10	7	55
72	786122	10	<0.2	0.47	35	150	<5	6.02	<1	18	36	109	4.44	<10	2.11	769	<1	0.10	12	1790	10	<5	<20	243	<0.01	<10	46	<10	10	46
73	786123	10	<0.2	0.85	35	135	<5	5.67	<1	29	114	100	4.43	<10	1.27	803	<1	0.12	22	1680	8	<5	<20	224	0.01	<10	102	<10	9	53
74	786124	10	<0.2	0.95	20	735	<5	8.34	<1	27	95	72	5.16	<10	1.32	1041	<1	0.11	21	1710	8	<5	<20	231	0.01	<10	108	<10	9	56
75	786125	550	0.7	0.72	2700	25	30	5.35	<1	94	10	120	3.43	<10	0.17	488	23	0.10	26	1450	16	10	<20	107	0.03	<10	26	<10	6	78
76	786126	5	<0.2	0.05	<5	5	<5	0.02	<1	<1	<1	<1	0.15	<10	<0.01	18	<1	<0.01	2	70	2	<5	<20	2	<0.01	<10	2	<10	<1	2
77	786127	5	<0.2	2.01	20	120	<5	5.94	<1	30	81	71	5.43	<10	2.11	1140	<1	0.14	26	1420	6	<5	<20	227	0.01	<10	152	<10	9	67
78	786128	10	<0.2	2.06	25	205	<5	6.52	<1	30	75	78	5.58	<10	2.13	1364	2	0.18	22	1290	6	<5	<20	312	0.02	<10	149	<10	11	75
79	786129	30	<0.2	1.02	15	140	<5	2.70	<1	14	16	62	2.54	<10	1.08	731	1	0.10	10	940	16	<5	<20	128	0.02	<10	41	<10	18	42
80	786130	20	0.2	0.65	30	30	<5	5.45	<1	12	36	94	3.83	<10	0.59	506	<1	0.09	13	1360	14	<5	<20	133	<0.01	<10	52	<10	14	31
81	786131	10	<0.2	0.56	25	45	<5	6.39	<1	18	54	86	5.21	<10	0.82	611	<1	0.08	12	1700	12	<5	<20	160	0.01	<10	69	<10	9	46
82	786132	5	<0.2	0.88	15	120	<5	6.73	<1	24	71	67	5.48	<10	1.78	1036	<1	0.10	19	1540	10	<5	<20	247	0.01	<10	98	<10	9	58
83	786133	5	<0.2	0.82	10	80	<5	7.22	<1	30	63	219	4.91	<10	3.12	1651	1	0.08	19	1380	18	<5	<20	297	<0.01	<10	96	<10	9	49
84	786134	10	<0.2	0.59	10	80	<5	4.43	<1	20	50	88	3.30	<10	1.77	898	<1	0.10	23	1100	12	<5	<20	227	<0.01	<10	67	<10	12	46
QC DATA:																														
Repeat:																														
1	786051	5	0.2	2.01	20	60	<5	3.15	<1	31	134	191	6.32	<10	3.39	1022	<1	0.21	65	3390	6	<5	<20	192	0.12	<10	209	<10	11	66
10	786060	15	<0.2	0.57	70	50	<5	8.01	<1	45	354	33	3.78	<10	5.02	810	<1	0.10	432	180	10	<5	<20	751	<0.01	<10	55	<10	3	30
19	786069	5	0.2	0.21	<5	115	<5	7.10	<1	20	80	25	3.16	<10	3.78	664	2	0.05	179	210	8	<5	<20	550	<0.01	<10	38	<10	3	19
27	786077	35																												
36	786086	10	<0.2	1.64	95	50	<5	3.35	<1	30	65	133	5.28	<10	1.89	899	<1	0.15	37	1580	6	<5	<20	172	<0.01	<10	86	<10	7	58
45	786095	20	<0.2	1.69	340	45	<5	1.72	<1	29	18	203	6.30	<10	1.40	1027	<1	0.21	17	3870	10	15	<20	164	<0.01	<10	108	<10	10	71
52	786102	20																												
54	786104	15	<0.2	1.14	15	50	<5	6.83	<1	20	48	106	5.44	<10	1.16	764	<1	0.15	16	1870	8	<5	<20	174	<0.01	<10	61	<10	8	49
71	786121	15	<0.2	0.51	35	195	<5	4.01	<1	18	47	103	5.41	<10	1.43	716	<1	0.10	12	2060	10	<5	<20	143	0.02	<10	50	<10	7	51
Resplit:																														
1	786051	15	0.2	2.10	20	70	<5	3.18	<1	32	133	188	6.52	<10	3.30	1071	<1	0.21	69	3360	6	<5	<20	203	0.10	<10	210	<10	12	70
36	786086	15	<0.2	1.76	105	50	<5	3.41	<1	32	74	129	5.52	<10	1.86	938	<1	0.16	40	1630	6	5	<20	184	<0.01	<10	93	<10	7	60
71	786121	15	<0.2	0.53	40	170	<5	3.86	<1	19	46	105	5.72	<10	1.43	729	<1	0.10	13	2110	8	<5	<20	142	0.02	<10	50	<10	7	53
Standard:																														
Pb129a			11.8	0.84	10	75	<5	0.51	55	5	3	1394	1.60	<10	0.65	364	5	0.04	5	440	6182	10	<20	37	0.05	<10	16	<10	1	9901
Pb129a			12.2	0.83	10	70	<5	0.48	55	5	3	1392	1.56	<10	0.65	366	5	0.04	4	440	6172	10	<20	41	0.05	<10	15	<10	1	9973
Pb129a			11.6	0.84	10	65	<5	0.50	56	6	3	1423	1.55	<10	0.63	373	5	0.04	5	450	6210	10	<20	41	0.05	<10	14	<10	1	9969
SE29		595																												
SE29		600																												
SE29		610																												

JJ/ap
df:p139s

ECO TECH LABORATORY LTD.
Jutta Jealouse

APPENDIX II
DIAMOND DRILL LOGS

LAKEWOOD MINING LTD. 2007 DRILLING PROGRAM DORADO ZONE												
HOLE No.			**D07-04**	**DORADO GRID**		**1500N 1145W**						
FROM	**TO**		**Azimuth NA, Dip -90.**							**ASSAYS**		
meters	**meters**	**GEOCODE**	**GEOLOGICAL DESCRIPTION**	**ALTERATION AND VEINING**	**MINERALIZATION**	**SAMP#**	**FROM**	**TO**	**Au ppm**	**Cu ppm**	**As ppm**	
0.0	13.0		CASING boulders and till ~25% recovery 95% fragments Nicola volcanics, 5% Kamloops intrusives.									
13.0	23.1	QTIL	heterolithic basal till.									
23.1	59.6	NEBA	Melanocratic dark green Nicola augite porphyry basalt flows and agglomerate. Locally vesicular with vugs lined with carbonate. Local fine grained tuff and epiclastic to argillaceous intervals.	Pervasive moderate to very strong to locally weak propylitic alteration overprinted by weak to locally strong white calcite stockwork and shear veins.								
		FALZ	26-29.75 shear zone both competent and incompetent shears 40-70 deg to C.A. moderate chloritic alteration throughout. Destruction of vesicles and preservation of augite.									
			26.67 - 27.15 chloritic shear zone. Fabric 45 deg to C.A.									
			28.05 - 28.65 chloritic shear zone -rotational fabrics with brecciated white quartz, carbonate veining comprising 8% of interval and syn-shear crytocrystalliine "rhyolite" comprising 5% of interval.									
				29.7- 30.7 moderate pervasive but mottled epidote alteration with weak chloritic stockwork fractures.								
				30.7-32.2 brecciated quartz carbonate veined shear and wrench zone. 3% of interval is open vugs. zone.								
			32.2 - rock fairly heterogeneous with zones appearing to be flow and other with heterolithic by compositionally similar angular fragments locally common. Fragments-zones?? With vesicles aligned ~75 deg. To C.A.	32.2-40.5 moderate pervasive but mottled epidote alteration with weak chloritic stockwork fractures. Decreasing overall epidote alteration down hole with increasingly selective epidote alteration of matrix. Plages sauseritized. Augite weakly chloritized with possible gypsum coatings. Weak to moderate planar tan ankerite and ragged white calcite veining.	very weak erratically occurring hematite in basaltic fragments.								
				37 - possible slight increase in clay alteration but may also be decreasing weathering down hole.								
			59.6 Flow laminated contact 43 deg. To C.A.									

HOLE No.			D07-04	DORADO GRID	1500N 1145W						
FROM	TO		Azimuth NA, Dip -90.						ASSAYS		
meters	meters	GEOCODE	GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	FROM	TO	Au ppm	Cu ppm	As ppm
59.6	61.8	NEAF	Dark green fine grained feldspar porphyry flow.	Pervasive chlorite alteration with sausseritized plagioclase (6%)'. Random subparrallel banded quartz +/- carbonate vines ~70 deg. to C.A.	strong trace erratically occurring disseminated hematite.						
			Broken core at contact.								
61.8	91.2	NEBA	Melanocratic dark green Nicola augite porphyry basalt agglomerate. Breccia and lapilli tuff. Locally vesicular with vugs lined with carbonate. Local fine grained tuff and epiclastic to argillaceous intervals.	Pervasive moderate to very strong to locally weak propyllitic alteration overprinted by weak to locally strong white calcite stockwork and shear veins. Rock mass has a subtle but pervasive weak to locally moderate shear associated (~70 deg to C.A. "grey" clay alteration that overprints earlier green chlorite+/- epidote alteration.	very rae trace malachite						
				76.9 - 86.4 strong light green pervasive epidote or clinozoisite alteration that is overprinted by "invisible" clay alteration.	73.5 - 75 3% dark red hematite in later curviplanar slips. 60-75 deg. To C.A.						
			86.4 fault zone 25 deg. To C.A.	intense chlorite-clay alteration.							
				86.4-89.5 Strong light green pervasive epidote or clinozoisite alteration that is weakly overprinted by "invisible" clay alteration. Epidote alteration decreasing down hole.							
			faulted contact - 28 deg. To C.A.								
91.2	91.5	FALT	Fault Zone clay gouge with ankeritic vein or rhyolitic fragments. Locally strongly hematitic.	intense chlorite-clay alteration.	25% hematite in lower 10 cm of interval. ~35 deg. to C.A.						
			gradational contact decreasing clay alteration.								

HOLE No.			D07-04	DORADO GRID	1500N 1145W							
FROM	TO		Azimuth NA, Dip -90.						ASSAYS			
meters	meters	GEOCODE	GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	FROM	TO	Au ppm	Cu ppm	As ppm	
91.5	109.1	NEBA	Melanocratic dark green Nicola augite porphyry basalt agglomerate. Breccia and lapilli tuff. Locally vesicular with vugs lined with carbonate. Local fine grained tuff and epiclastic to argillaceous intervals. The sequence is intruded by minute to 5 cm thick laminated "rhyolite" dykelets at all orientations with most displaying moderate deformation.	Pervasive moderate to very strong to locally weak propyllitic alteration overprinted by weak to locally strong white calcite +/- quartz stockwork and shear veins. Rock mass has pervasive moderate stockwork clay alteration that overprints earlier green chlorite+/- epidote alteration. This alteration style makes the rock very incompetent.								
				97 5 cm carbonate-quartz vein ragged 20 deg. To C.A.								
				100.0 109 very weak clay stockwork. And increased pale epidote-clinozoisite pervasive alteration.								
109.1	109.5	FALZ	Fault Zone clay gouge with ankeritic vein or rhyolitic fragments. Locally strongly hematitic.	intense chlorite-clay alteration.	very weak hematite as clots in chloritic gouge.							
109.5	122.2	NEBA	Medium to pale green Nicola augite porphyry basalt agglomerate.	Continuation of unit above fault but with similar alteration as from 100 m.	Weak hematite in late epidoteized and veined curviplanar fractures.							
					116-118 translucent quartz veins (4% of rock) with (5% in vein) bright native copper. Finely disseminated native copper in wallrock at vein margins. Trace overall.							
122.2	123.7	FALZ	Shear zone - Nicola basalt. Small ptygmatically folded rhyolite dykes with increasing shearing down hole. Dykes originally subparrallel to C.A.	strong chloritic and clay alteration.								
			Fault contact 20 deg. To C.A.									
123.7	124.1	FALZ	Shear Zone - alaskite host rock. Identical alteration and intrusions as above.									

HOLE No.			D07-04	DORADO GRID	1500N 1145W						
FROM	TO		Azimuth NA, Dip -90.						ASSAYS		
meters	meters	GEOCODE	GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	FROM	TO	Au ppm	Cu ppm	As ppm
124.1	138.4	KFPA	Leucocratic grey fine grained feldspar porphyry alaskite dyke. Wide chilled margins with a combination of tectonic and flow sheared contacts. Composition may be variable but possibly due to grain size. Coarser grained center of dyke had distinct ~10% quartz phenocrysts.	Unit moderately to strongly clay altered. With sausseritized plagioclase. Kspr appears unaltered. Crosscut by white bull quartz followed by small 2 to 20 mm pegmatitic grain quartz feldspar porphyry rhyolite dykelets that also cross cut early bull quartz veins and are in turn cross cut and incorporated into shear and raged tensional fine grained calcite-dolomite veins.	Dark brown oxidized hematite with productive zones associated with late carbonate stockwork and veining. Unit hosts rare trace to locally 1% extremely fine to medium grained euhedral pyrite.						
			134.3 30 cm rubble from pulling rods.								
			138.1 - 138.4 sheared but annealed (flow laminated?) alaskite.								
138.4	139.0	FALZ	Shear zone - Nicola basalt. Small ptygmatically folded rhyolite dykes with increasing shearing down hole. Dykes originally subparrallel to C.A.	strong chloritic and clay alteration.							
139.0	148.4	NEBA	Medium to pale green Nicola augite porphyry basalt agglomerate. Continuation of sequence above alaskite dyke. No "rhyolite' dykelets below 141 m.	Continuation of unit above fault but with similar alteration as from 100 m.	Weak hematite in late epidotized and veined curviplanar fractures.						
					139-143.5 Fine grained disseminated native copper mostly associated with strongly chloritic shears and wallrock linings small quartz-calcite veinlets.						
			Sheared but annealed contact 35 deg. To C.A.								
148.4	150.8	KFPA	Alaskite dyke. Same sequence with identical alteration and veining as at 124.1 m.								
			planar contact 20 deg. To C.A.								
150.8	179.8	NEBA	Medium to pale green Nicola augite porphyry basalt agglomerate. Continuation of sequence above alaskite dyke. No "rhyolite' dykelets below 141 m.	Continuation of unit above dyke but with similar alteration. Moderate white calcite veining and weak stockwork.	Weak hematite in late epidotized and calcite veined curviplanar fractures.						
				Very strong chlorite alteration to 153.7							

HOLE No.			D07-04	DORADO GRID	1500N 1145W						
FROM meters	TO meters	GEOCODE	Azimuth NA, Dip -90. GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	ASSAYS FROM	TO	Au ppm	Cu ppm	As ppm
				169.7 - 174.9 - wrenched and slightly sheared very strongly chlorite altered.. Locally gougy, several quartz-tan to red garnet veins ~70 deg. to C.A.	very weak hematite.						
			curviplanar Contact ~50 deg to C.A.								
179.8	183.1	KFHP	Grey fine grained Flow Laminated Dacite dyke. Similar in appearance to overlying "Alaskite" except for the presence of 6% fine grained flow aligned hornblende. Sequence has been tectonically brecciated with rotate fragments in gougy clay altered dacite gouge. brecciation appears to be a shallow angles to core axis. rock contains very fine grained interstitial sericite masses that do not affect rock strength.	locally strongly clay altered.							
			Clay altered broken core at contact.								
				185.1 - 185.4 intense chlorite-clay alteration.	weak hematite in late planar fractures.						
183.1	185.4	NEBA	Medium to pale green Nicola augite porphyry basalt agglomerate. Continuation of sequence above alaskite dyke. Laminations ~70 deg. To C.A.	Continuation of unit above dyke but with similar alteration. Moderate white calcite veining and weak stockwork.	Weak hematite in late epidotized and calcite veined curviplanar fractures and shears.						
185.4	186.1	KFHP	Grey fine grained Flow Laminated Dacite dyke.	Intensely clay altered.							
			planar contact 20 deg. To C.A.								
186.1	213.7	NEBA	Medium to pale green Nicola augite porphyry basalt agglomerate. Continuation of sequence above alaskite dyke.	Continuation of unit above dyke but with stronger chlorite alteration. Moderate white calcite veining and weak stockwork. Chlorite may be partially sericitized.	Weak hematite in late epidotized and calcite veined curviplanar fractures and shears.						
			clay altered raggedly curviplanar contact 20 deg to C.A.		190 much weaker hematite.						
213.7	217.0	KFHP	Grey fine grained Flow Laminated Dacite dyke. Flow laminated or welded tuff texture 35 to 65 deg. To C.A.	5% fine grained interstitial sericite. Locally intensely clay altered, sericite preserved but plagioclase destructive..							
217.0		EOH	END OF HOLE.								

LAKEWOOD MINING LTD. 2007 DRILLING PROGRAM - MONARCH CLAIM - IRON MASK PROJECT

HOLE No. **D07-05** **DORADO GRID** **1800W** **4000N**

FROM meters	TO meters	Azimuth NA, Dip -90. GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	ASSAYS FROM	TO	Au ppm	Cu ppm	As ppm
		Not logged recovery so bad that cannot measure anything								
		Hole abandoned @ ~45 meters and less than 4 meters of core.								

LAKEWOOD MINING LTD. 2007 DORADO DRILLING PROGRAM

HOLE No.			D07-06	DORADO GRID	1795 W	4003 N						
FROM	TO		Azimuth NA, Dip -90.	re drill of hole 05					ASSAYS			
meters	meters	GEOCODE	GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	FROM	TO	Au ppm	Cu ppm	As ppm
0.0	35.7		OVERBURDEN NO CORE RECOVERED -overburden is primarily a brown clay matrix cobble till.								
35.7	40.0	NEBB	Maroon heterolithic volcanic breccia. Fragments are subrounded invariably basaltic and quite variable in grain size.	Local intense pervasive to white kaolinitic clay alteration overprinting white carbonate breccia veining. Carbonate occupies 3 to 7% of rock and occurs as fragments and often broken discreet planar and anastomozing stockwork zones with a preferred orientation of 0 deg. To C.A.		132001	35	38.41	1.5	73	8
40	40.5	FALZ	Tan-red ankeritic vein shear zone	intense carbonate and clay alteration.		132002	38.41	39.01	1.3	44	9
40.5	42	NEBB	Tan clay altered and ankeritic broken volcanic breccia.	strong ankerite and clay alteration		132003	39.01	41.45	0.8	44	6
42	46.75	NEBB	Green heterolithic volcanic breccia. Fragments are subrounded invariably basaltic and quite variable in grain size.	strongly chlorite-clay altered rock is overprinted by intense cryptic to white kaolinitic clay alteration overprinting white carbonate breccia veining. Carbonate occupies 3 to 7% of rock and occurs as fragments and often broken discreet planar and anastomozing stockwork zones with a preferred orientation of 0 deg. To C.A.		132004	41.45	42.98	4.6	40	6
46.75	47.25	TANK	Ivory ankerite? Or intensely ankeritized cryptocrystalline dacite vein/dyke. Dominant orientation ~15 deg. to C.A. Matrix comprises about 30% or rock. Fragments are reddish stained NICOLA? Volcanic with strong ankerite and minor silica overprint. Silica also occurs as minute tensional veinlet swarm in fragments.			132005	42.98	46.03	2.5	42	5
47.25	50.6	NEBB	Green and dark red heterolithic volcanic breccia. Fragments are subrounded invariably basaltic and quite variable in grain size.	Intense pervasive to white kaolinitic clay alteration overprinting white carbonate breccia veining. Carbonate occupies 10-15% of rock and occurs as fragments and often broken discreet planar and anastomozing stockwork zones with a preferred orientation of 70 deg. To C.A.		132006	46.03	47.85	3.5	57	9
50.6		EOH	Hole abandoned. EOH			132007	47.85	50.6	2.6	67	10

LAKEWOOD MINING LTD. 2007 DRILLING PROGRAM - DORADO GRID												
HOLE No.			D07-07	DORADO GRID	1500 W	3950 N						
FROM	TO		Azimuth NA, Dip -90.							ASSAYS		
meters	meters		GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	FROM	TO	Au ppm	Cu ppm	As ppm	
0.0	6.0		OVERBURDEN NO CORE RECOVERED -overburden is primarily a brown clay matrix cobble till.									
6.0	18.1	NEBB	Green and dark red heterolithic volcanic breccia. Fragments are subrounded invariably basaltic and quite variable in grain size.	Intense pervasive to white kaolinitic clay alteration overprinting white carbonate breccia veining. Carbonate occupies 3 to 7% of rock and occurs as fragments and often broken discreet planar and anastomozing stockwork zones with a preferred orientation of 0 deg. To C.A.		132008	16.13	17.68	3.7	110	6	
18.1	18.85	FALZ	GREEN AND WHITE clay altered shear zone with numerous rounded red milled wallrock fragments.	intense clay alteration.		132009	17.68	20.42	1.3	59	26	
18.85	19.6	TANK	Ivory ankerite? Or intensely ankeritized cryptocrystalline dacite vein/dyke. Dominant orientation ~15 deg. to C.A. Matrix comprises about 30% or rock. Fragments are reddish stained NICOLA? Volcanic with strong ankerite and minor silica overprint. Silica also occurs as minute tensional veinlet swarm in fragments.			132010	20.42	23.17	2.7	65	9	
19.6	20.3	FALZ	GREEN AND WHITE clay altered shear zone with numerous rounded red milled wallrock fragments.	intense clay alteration.		132011	23.17	23.17	1.8	91	15	
20.3	24.3	NEBB	Green and dark red heterolithic volcanic breccia. Fragments are subrounded invariably basaltic and quite variable in grain size.	Intense pervasive to white kaolinitic clay alteration overprinting white carbonate breccia veining. Carbonate occupies 3 to 7% of rock and occurs as fragments and often broken discreet planar and anastomozing stockwork zones with a preferred orientation of 0 deg. To C.A.		132012	24.99	26.52	1.8	73		
24.3	24.95	FALZ	GREEN AND WHITE clay altered shear zone with numerous rounded red milled wallrock fragments.	intense clay alteration.		132013	26.52	28.35	1.9	80	86	

HOLE No.			D07-07	DORADO GRID	1500 W	3950 N						
FROM	TO		Azimuth NA, Dip -90.							ASSAYS		
meters	meters		GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION	SAMP#	FROM	TO	Au ppm	Cu ppm	As ppm	
24.95	26	TANK	Ivory ankerite? Or intensely ankeritized cryptocrystalline dacite vein/dyke. Dominant orientation ~30 deg. to C.A. Matrix comprises about 30% or rock. Fragments are reddish stained NICOLA? Volcanic with strong ankerite and minor silica overprint. Silica also occurs as minute tensional veinlet swarm in fragments.	SEE GEOLOGICAL DESCRIPTION NOTE silicification and silica tension veinlets		132014	28.35	30.18	2.1	65	28	
26	30.18	NEBB	Dark red heterolithic volcanic breccia. Fragments are subrounded invariably basaltic and quite variable in grain size.	Intense pervasive to white kaolinitic clay alteration overprinting white carbonate breccia veining. Carbonate occupies 3 to 7% of rock and occurs as fragments and often broken discreet planar and anastomozing stockwork zones with a preferred orientation of 0 deg. To C.A. also greenish clay alteration stain of various intensities alters rock. green rock is coherent but soft with 0 harness but harder than clay.		132015	30.18	31.39	2.6	63	10	
30.18	32.4	TANK	Ivory ankerite breccia vein and beige-green felsic volcanic? shear zone. Dominant orientation ~0-30 deg. to C.A. Greenish sandy-clay matrix comprises about 30% or rock. Fragments are reddish stained NICOLA? Volcanic with strong ankerite and minor silica overprint. Silica also occurs as minute tensional veinlet swarm in fragments. Greenish clay may be altered pyrite.	SEE GEOLOGICAL DESCRIPTION NOTE silicification and silica tension veinlets		132016	31.39	32.61	1.5	68	15	
32.4	34.14	FALZ	Grey clay alteration-shear zone. Rock appears to have ~20% rounded sand sized to 20 mm dia. wallrock fragments in it.			132017	32.61	34.14	2.9	90	42	
34.1		EOH	Hole abandoned-EOH			132018	Cu130 std		843.5	4392	1244	

Hole # D-07-08, Dorado Grid Area **Grid Location 1800W, 3997 N (Re-drill of Holes 2007-05 and 06)**
Date Hole Started: Novbember 26, 2007 **Total Depth: 331.32 m**
Date Completed: December 3, 2007 **Azimuth: NA Dip -90**
Driller: Frontier Drilling
Drilling Supervised by J.E.L. Lindinger, P.Geo
Logged by M.S. Cathro, P.Geo, January 22-24, 2008

FROM	TO					SAMPLE #	FROM	TO	INTERVAL	ASSAYS				
meters	meters	GEOCODE	GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION		Metres	Metres	Metres	Au ppb	Ag ppm	As ppm	Cu ppm	Sb ppm
0.0	45.7		OVERBURDEN - NO CORE RECOVERED											
45.7	58.6	NEBB	Broken rock and gouge. Fragments of andesitic tuff and breccia, Dark green and maroon.	1-3% carbonate veins and vein fragments										
58.6	61.0	NEBB	Heterolithic breccia. Dark green and maroon. Fragments are andesitic plagioclase crystal tuff.	59.8 to 60.2 - 50% qtz/carb veins		786051	96.7	97.6	0.90	10	<0.2	20	176	<5
61.0	67.2	FALZ	Broken rock and gouge. Andesitic tuff and breccia, Dark green and maroon.	1-2% carb veins and fragments		786052	97.6	98.62	1.02	<5	<0.2	130	35	<5
67.2	79.0	NEBB	Andesitic tuff and breccia. Dark green and maroon. Plagioclase phyric in places. Becomes weakly magnetic at about 76 m.	1-3% carb veins		786053	98.62	102.72	4.10	<5	<0.2	15	53	<5
79.0	91.5	FALZ	Gouge and broken rock. Fragmental volcanic with clasts to 2 cm. Pale pinkish-red. Soft, muddy, altered in places. Core appears lumpy and slightly washed out.	Weak to moderate hematite		786054	102.72	105.77	3.05	5	<0.2	15	50	<5
91.5	97.6	NEBB	Andesitic breccia, lapilli tuff and medium grained tuff.	Weak propylitic (carb, epidote) and		786055	105.77	108.81	3.04	25	0.2	10	48	<5
			Dark greenish grey with wispy red hematitic patches. Rare dark grey-black grains (altered augite?)	hematite alteration. 1-3% carb veins at 20-40 degrees to CA		786056	108.81	111.86	3.05	<5	<0.2	15	92	<5
97.6	98.6	TANK	Sheared intermediate tuff with orange stained carbonate veining.	20% carb veins at 20-50 degrees to CA. Moderate ankerite staining.		786057	111.86	114.91	3.05	<5	<0.2	10	53	<5
98.6	117.9	NEBB	Sheared andesitic volcanic breccia. Massive to poorly laminated (beds to 5 cm wide at 10-40 degrees to CA).	Weak propylitic alteration. Sericite?		786058	114.91	117.96	3.05	<5	<0.2	10	45	<5
			Pale to medium green. Core becomes quite broken up	Approx 20% carbonate veins as a		786059	117.96	121.01	3.05	5	<0.2	10	88	<5
			into 2-10 cm pieces; rarely 20 cm. Breccia fragments are	network ranging from 1-2 mm wide to		786060	121.01	124.06	3.05	5	<0.2	70	34	5
			pale, fine-medium grained tuff? Rare fragments (~5%)	1 cm wide. Larger veins to 5 cm show		786061	124.06	127.10	3.04	5	<0.2	5	9	<5
			with porphyritic mafic grains, possibly picrite(?). Overall	banding. Veins have CA 10-40 and		786062	127.10	130.15	3.05	30	<0.2	10	39	<5
				occasionally 70-90 degrees.		786063	130.15	133.20	3.05	10	<0.2	<5	27	<5
117.9	122.4	FALZ	Fault gouge and broken rock. Soft, muddy, pale to	Carbonate veins and fragments to 1		786064	133.20	136.25	3.05	5	<0.2	<5	31	<5
			medium green. Non magnetic. Core looks washed out in	cm (e.g. 120.8 m)		786065	136.25	139.30	3.05	5	<0.2	25	45	<5
122.4	130.8	NEBB	Altered volcanic breccia. Clasts of medium grained tuff	Moderate propylitic (epidote, chlorite,		786066	139.30	142.34	3.04	10	<0.2	110	49	10
			and rare grey argillite. Some clasts silicified. Pale	carbonate) and sericite alteration.		786067	142.34	145.39	3.05	25	<0.2	15	25	<5
			greenish-grey. Non magnetic. Lower contact gradational.	Patchy silica. Calcite-quartz veins to 1		786068	145.39	148.44	3.05	30	<0.2	<5	15	<5
130.8	143.0	NESS	Intermixed greyish fine-grained siltstone/argillite and	Moderate to heavy (10-40%)	142.5 - trace pyrite	786069	148.44	151.49	3.05	10	<0.2	<5	24	<5
			greenish andesitic fine-medium grained tuff.	carbonate veining and carbonate		786070	151.49	154.54	3.05	10	<0.2	5	40	<5
			Siltstone/argillite bands are generally less than 1 m thick.	alteration. Carb veins generally <2 cm		786071	154.54	157.58	3.04	10	<0.2	5	113	<5
143.0	150.0	KVBX	Breccia with fragments of intermediate tuff/siltstone and	Carbonate and sericite altered? 10-		786072	157.58	160.63	3.05	5	<0.2	10	96	<5
			lapilli tuff. Pale grey. Becoming massive grey-buff	40% calcite-quartz veins. 1-3% bright		786073	160.63	163.68	3.05	25	<0.2	15	98	<5
150.0	159.9	KVTF	Massive tuff(?). Pale grey-buff. Vague breccia textures	Moderate to intense carbonate-sericite		786074	163.68	166.73	3.05	25	<0.2	10	67	<5
			locally. Non-magnetic,	(?) alteration. Wispy ankeritic shears,		786075	STD			540	0.6	2675	116	5
				veins and patches.		786076	BLK			<5	<0.2	<5	<1	<5
159.9	161.4	KVTF	Tuff or siltstone. Buff-orange. Local breccia textures. Locally banded at 45 degrees to CA. Core vuggy and washed out in places. 15 cm gouge at 161.25 to 161.4.	Moderate ankerite/hematite staining.		786077	166.73	169.78	3.05		<0.2	20	132	5
161.4	169.1	KVTF	Tuff/siltstone. Buff-grey. Grey wispy laminations in	Weak to moderate ankeritic alteration.		786078	169.78	172.82	3.04	20	<0.2	5	113	<5
			places. CBA 45-75 degrees. Non magnetic. Lower	1-3 % carb veins to 1 cm wide.		786079	172.82	175.87	3.05	5	<0.2	15	179	5

FROM	TO					SAMPLE #	FROM	TO	INTERVAL	ASSAYS				
meters	meters	GEOCODE	GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION		Metres	Metres	Metres	Au ppb	Ag ppm	As ppm	Cu ppm	Sb ppm
			contact gradational. CBA 45 degrees at 165 m, and 75			786080	175.87	178.92	3.05	10	<0.2	20	62	<5
169.1	181.1	KVLT	Fine grained tuff/siltstone and local lapilli tuff. Locally finely laminated. CBA 45 degrees at 169.6 m, 45 degrees at 170.8, 60 degrees at 176.3 m.	Weak propylitic alteration. Local weak ankeritic alteration. 1-3% carb veins to 1 cm wide.		786081	178.92	181.97	3.05	20	<0.2	20	81	<5
						786082	181.97	185.02	3.05	15	<0.2	20	110	<5
						786083	185.02	188.01	2.99	5	<0.2	190	159	20
						786084	188.01	191.11	3.10	20	<0.2	50	127	<5
181.1	190.5	NEAT	Andesitic tuff. Fine-grained. Rarely augite phyric. Green to medium green-grey. CA vague 50-70 degrees.	Weak propylitic (cal, chl) alteration. 2-3% calcite veins to 1 cm (20-60 degrees to CA)		786085	191.11	194.16	3.05	40	<0.2	30	153	10
						786086	194.16	197.21	3.05	10	<0.2	95	130	5
						786087	197.21	200.26	3.05	10	<0.2		134	20
190.5	195.6	KVLT	Fine-grained siltstone/tuff. Orange, grey and green. Locally finely laminated grey siltstone/argillite. Contorted bedding in places. CBA 40 degrees at 191.4 m; 50 degrees at 192.6 m.	Weak to moderate ankeritic alteration. Irregular carb veins to 1 cm wide that appear to crosscut the ankerite.		786088	200.26	203.30	3.04	5	<0.2	135	118	5
195.6	233.8	NEAT	Massive to laminated fine-grained andesitic tuff. Greenish grey. Locally plagioclase phyric. Non magnetic. Local finely laminated argillite. Lower contact gradational. CBA 50@199.8; 80@203.3; 40@206.3; 45@212; 60@223.	Weak carbonate-sericite alteration. 1-3% carb vns.	200.7-201.6 m- Quartz-calcite vein with trace py. 206.8-234 m - minor py (1-2% overall) in veins to 1 cm thick and irregular patches.	786089	203.30	206.35	3.05	30	<0.2	60	131	<5
						786090	206.35	209.40	3.05	30	0.2		182	15
						786091	209.40	212.45	3.05	15	<0.2		197	15
						786092	212.45	215.50	3.05	10	<0.2	970	161	
						786093	215.50	218.54	3.04	10	0.2	850	183	
						786094	218.54	221.59	3.05	25	0.2		219	15
						786095	221.59	224.64	3.05	25	<0.2		199	15
						786096	224.64	227.69	3.05	15	<0.2	510	179	20
						786097	227.69	230.74	3.05	5	<0.2		130	20
						786098	230.74	233.78	3.04	10	<0.2	155	100	5
						786099	233.78	236.83	3.05	10	<0.2	45	110	5
						786100	STD			565	0.6	2705	117	5
						786101	BLANK			5	<0.2	<5	<1	<5
233.8	245.5	NEAT	Laminated tuff and argillite. Grey and buff. Non magnetic. Crushed zones towards bottom. CBA 35@233.9; 60@235; 60@ 238.5.	Bands of pervasive weak to moderate ankeritic alteration. 2% carb vns to 1 cm. Zones of up to 10% carb vein over 30 cm crosscutting ankerite		786102	236.83	239.88	3.05		<0.2	90	85	5
						786103	239.88	242.93	3.05	15	<0.2	20	189	<5
						786104	242.93	245.98	3.05	15	<0.2	15	110	<5
						786105	245.98	249.02	3.04	10	<0.2	15	121	<5
245.5	248.4	FALZ	Fault gouge. Grey. Fragments of fine-grained tuff. Non-magnetic. Lowe contact gradational.	Ankeritic staining last 0.5 m.		786106	249.02	252.07	3.05	5	<0.2	25	133	<5
248.4	257.0	NEAT	Massive fine-grained tuff. Buff and pale grey. Locally brecciated with augite phyric fragments. Non-magnetic,	Weak to moderate ankeritic alteration. Rare carb veins.		786107	252.07	255.12	3.05	15	0.2	20	64	<5
						786108	255.12	258.17	3.05	35	<0.2	30	85	<5
						786109	258.17	261.22	3.05	5	<0.2	15	26	<5
257.0	266.7	KFPA	Massive, leucocratic (pale buff-grey) felsic dyke or crystal tuff(?). Vague flow(?) laminations at bottom. Irregular vague breccia textures at 265m.	266.8-becomes ankeritic. Rare carb veins last 3-4 m.		786110	261.22	264.26	3.04	<5	<0.2	10	7	<5
						786111	264.26	267.31	3.05	10	<0.2	15	89	<5
						786112	267.31	270.36	3.05	10	<0.2	25	52	<5
266.7	269.6	NEAT	Massive fine-grained tuff. Buff and pale grey. Locally brecciated with augite phyric fragments. Non-magnetic			786113	270.36	273.41	3.05	15	<0.2	15	133	<5
269.6	270.5	FALZ	Fault gouge. Pale pinkish-red. Fragments of massive, fine-grained felsic dyke or tuff present at 257 m. Non-magnetic.	Hematitic.		786114	273.41	276.46	3.05	20	<0.2	20	135	<5
270.5	273.5	NELT	Lapilli tuff. Mottled. Grey and buff. Massive to poorly laminated. Non magnetic. CBA 60 degrees at 273 m.			786115	276.46	279.50	3.04	20	<0.2	20	138	<5
		NEAT	Fine-grained tuff. Massive to poorly laminated. Buff with local grey patches. Non-magnetic.	Weak to moderate ankerite. Rare carb veins.		786116	279.50	282.55	3.05	15	<0.2	15	37	<5
						786117	282.55	285.50	2.95	10	<0.2	20	137	<5
						786118	285.50	288.65	3.15	25	<0.2	15	123	<5
						786119	288.65	291.70	3.05	10	<0.2	30	99	<5

FROM	TO					SAMPLE #	FROM	TO	INTERVAL	ASSAYS				
meters	meters	GEOCODE	GEOLOGICAL DESCRIPTION	ALTERATION AND VEINING	MINERALIZATION		Metres	Metres	Metres	Au ppb	Ag ppm	As ppm	Cu ppm	Sb ppm
273.5	306.0					786120	291.70	294.75	3.05	15	<0.2	35	103	<5
						786121	294.75	297.79	3.04	20	<0.2	35	98	<5
						786122	297.79	300.84	3.05	10	<0.2	35	109	<5
						786123	300.84	303.89	3.05	10	<0.2	35	100	<5
						786124	303.89	306.94	3.05	10	<0.2	20	72	<5
						786125	STD			550	0.7	2700	120	10
						786126	BLANK			5	<0.2	<5	<1	<5
306.0	313.9	NEAT	Fine-grained tuff or dyke(?) and lapilli tuff (at top).	1-2% carb veins.		786127	306.94	309.99	3.05	5	<0.2	20	71	<5
			Medium to dark grey and greyish green. Lower contact			786128	309.99	313.03	3.04	10	<0.2	25	78	<5
313.9	318.7	TANK	Massive altered tuff. Pale grey to buff. Non-magnetic.	Ankeritic patches. Sericite?		786129	313.03	316.09	3.06	30	<0.2	15	62	<5
			CBA 80 degrees @326.8m; 60@ 328.4 m.			786130	316.09	319.15	3.06	20	0.2	30	94	<5
318.7	325.0	TANK	Massive to poorly laminate tuff(?). Buff to orange. Non	Moderate ankerite alteration. 1-2%		786131	319.15	322.18	3.03	10	<0.2	25	86	<5
			magnetic.	carb veins.		786132	322.18	325.23	3.05	5	<0.2	15	67	<5
325.0	331.3	NEAT	Massive to poorly laminated tuff. Non-magnetic. EOH.	Carbonate alteration? 1-3% carb veins.		786133	325.23	328.27	3.04	5	<0.2	10	219	<5
						786134	328.27	331.32	3.05	10	<0.2	10	88	<5

APPENDIX III

GEOTECHNICAL LOGS

Renaissance Geoscience Services Inc- Geotechnical Logging Template

Hole:	D07-04			Logger:	DILLABOUGH					
Drill date:	10/1/2007				Core size:	NQ	Dip:	-90	Azimuth:	0
Log date:	10/1/2007		N:	2750	W	1145			Elev.	930
From	To	Length	Recovery	Percent	RQD	Percent	Fractures	Strength	Comments	Fracture
(m)	(m)	(m)	(m)	(%)	(m)	(%)	(per run)			Condition
6.40	6.71	0.31	0.15	48%		0%				
6.71	9.75	3.04	1.00	33%		0%				
9.75	10.06	0.31	0.34	110%		0%				
10.06	12.80	2.74	0.53	19%		0%				
12.80	14.30	1.50	0.54	36%		0%				
14.30	15.85	1.55	0.38	25%		0%				
15.85	17.37	1.52	0.39	26%		0%				
17.37	18.54	1.17	0.95	81%		0%				
18.54	18.90	0.36	0.31	86%		0%				
18.90	19.35	0.45	0.50	111%		0%				
19.35	19.66	0.31	0.30	97%		0%				
19.66	20.42	0.76	0.90	118%		0%				
20.42	21.64	1.22	0.30	25%		0%				
21.64	22.56	0.92	0.75	82%		0%				
22.56	23.17	0.61	0.35	57%		0%				
23.17	24.69	1.52	1.54	101%	0.87	57%			lots of jumbled core	chlorite
24.69	26.21	1.52	1.50	99%	1.26	83%			lots of jumbled core	chlorite
26.21	27.74	1.53	1.37	90%	0.41	27%			lots of jumbled core	chlorite
27.74	29.26	1.52	1.30	86%	0.21	14%			lots of jumbled core	chlorite
29.26	30.79	1.53	1.24	81%	0.46	30%			lots of jumbled core	chlorite
30.79	32.31	1.52	1.50	99%	0.57	37%			lots of jumbled core	chlorite
32.31	33.83	1.52	1.54	101%	0.81	53%			lots of jumbled core	chlorite
33.83	35.36	1.53	1.35	88%	1.24	81%			lots of jumbled core	chlorite
35.36	36.88	1.52	1.53	101%	1.53	101%			lots of jumbled core	chlorite
36.88	38.41	1.53	1.55	101%	1.48	97%			lots of jumbled core	chlorite
38.41	39.93	1.52	1.47	97%	1.39	91%			lots of jumbled core	chlorite
39.93	41.45	1.52	1.57	103%	1.05	69%			lots of jumbled core	chlorite
41.45	42.98	1.53	1.59	104%	1.38	90%			lots of jumbled core	chlorite
42.98	44.50	1.52	1.31	86%	0.61	40%			0.19 crumbly gouge	chlorite
44.50	46.03	1.53	1.46	95%	1.37	90%			lots of jumbled core	chlorite
46.03	47.55	1.52	1.56	103%	1.32	87%			lots of jumbled core	chlorite
47.55	49.22	1.67	1.57	94%	0.67	40%			lots of jumbled core	
49.22	50.75	1.53	1.61	105%	0.80	52%			lots of jumbled core	
50.75	52.43	1.68	1.20	71%	0.95	57%			lots of jumbled core	
52.43	53.65	1.22	1.53	125%	1.10	90%			lots of jumbled core	
53.65	55.17	1.52	1.51	99%	1.40	92%			lots of jumbled core	

Hole:	D07-04			Logger:	DILLABOUGH					
55.17	56.69	1.52	1.48	97%	1.36	89%			lots of jumbled core	
56.69	58.22	1.53	1.52	99%	1.06	69%			lots of jumbled core	
58.22	59.74	1.52	1.48	97%	1.18	78%			lots of jumbled core	
59.74	61.27	1.53	1.50	98%	0.51	33%			lots of jumbled core	
61.27	63.09	1.82	1.64	90%	0.57	31%			0.18 crumbly gouge	
63.09	64.62	1.53	1.54	101%	0.70	46%			lots of jumbled core	
64.62	66.14	1.52	1.54	101%	1.17	77%			lots of jumbled core	
66.14	67.67	1.53	1.48	97%	0.89	58%			lots of jumbled core	
67.67	69.19	1.52	1.53	101%	1.16	76%			lots of jumbled core	
69.19	70.71	1.52	1.53	101%	1.27	84%			lots of jumbled core	
70.71	72.24	1.53	1.48	97%	1.41	92%			core crumbled	
72.24	73.76	1.52	1.52	100%	1.32	87%			core shattered	
73.76	75.29	1.53	1.47	96%	1.12	73%				
75.29	76.81	1.52	1.53	101%	1.14	75%				
76.81	78.33	1.52	1.51	99%	1.19	78%				
78.33	79.86	1.53	1.51	99%	1.42	93%				
79.86	81.38	1.52	1.48	97%	1.42	93%				
81.38	82.91	1.53	1.50	98%	1.28	84%				
82.91	84.43	1.52	1.51	99%	0.94	62%			core soft crumbled	
84.43	85.95	1.52	1.50	99%	1.32	87%			core shattered	
85.95	87.48	1.53	1.50	98%	1.10	72%			core crumbled	
87.48	89.00	1.52	1.55	102%	1.32	87%			core soft and crumbled	
89.00	90.53	1.53	1.53	100%	1.19	78%			core soft And crumbled	
90.53	92.05	1.52	1.50	99%	1.13	74%			core shattered	
92.05	93.57	1.52	1.52	100%	0.68	45%			core shattered	
93.57	95.10	1.53	1.47	96%	0.38	25%			core crumbled	
95.10	96.63	1.53	1.42	93%	0.70	46%			core crumbled	
96.63	98.15	1.52	1.50	99%	1.32	87%				
98.15	99.67	1.52	1.54	101%	0.96	63%				
99.67	101.19	1.52	1.50	99%	0.26	17%				
101.19	102.72	1.53	1.42	93%	1.07	70%				
102.72	104.24	1.52	1.53	101%	1.00	66%				
104.24	105.27	1.03	1.52	148%	1.40	136%				
105.27	107.29	2.02	1.52	75%	1.29	64%				
107.29	108.81	1.52	1.49	98%	1.07	70%				
108.81	110.34	1.53	1.53	100%	0.96	63%				
110.34	111.26	0.92	1.34	146%	0.94	102%				
111.26	113.39	2.13	1.50	70%	0.35	16%				
113.39	114.91	1.52	1.53	101%	1.34	88%				
114.91	115.67	0.76	0.77	101%	0.68	89%				
115.67	117.96	2.29	2.28	100%	1.81	79%				
117.96	119.48	1.52	1.50	99%	0.92	61%				

Hole:	D07-04			Logger:	DILLABOUGH					
119.48	121.01	1.53	1.51	99%	1.27	83%				
121.01	122.58	1.57	1.48	94%	1.09	69%				
122.58	124.06	1.48	1.51	102%	0.54	36%				
124.06	125.58	1.52	1.40	92%	0.41	27%				
125.58	128.47	2.89	1.50	52%	0.13	4%				
128.47	129.54	1.07	1.10	103%	0.11	10%				
129.54	131.07	1.53	1.50	98%	0.11	7%				
131.07	132.28	1.21	1.20	99%	0.14	12%				
132.28	133.20	0.92	1.03	112%	0.36	39%				
133.20	134.72	1.52	1.35	89%	0.00	0%				
134.72	137.77	3.05	2.80	92%	0.78	26%				
137.77	140.82	3.05	2.97	97%	1.40	46%				
140.82	143.87	3.05	3.05	100%	2.13	70%				
143.87	146.92	3.05	3.06	100%	1.63	53%				
146.92	149.96	3.04	3.03	100%	1.28	42%				
149.96	151.49	1.53	1.53	100%	0.58	38%			soft core	
151.49	154.54	3.05	3.12	102%	0.93	30%			soft core	
154.54	155.45	0.91	0.71	78%	0.64	70%				
155.45	156.06	0.61	0.83	136%	0.74	121%				
156.06	159.11	3.05	3.03	99%	2.86	94%				
159.11	162.16	3.05	3.06	100%	2.11	69%				
162.16	165.20	3.04	2.93	96%	2.56	84%				
165.20	168.25	3.05	3.04	100%	2.88	94%				
168.25	171.30	3.05	3.06	100%	2.12	70%				
171.30	174.35	3.05	3.07	101%	2.00	66%				
174.35	177.40	3.05	3.03	99%	2.27	74%				
177.40	180.44	3.04	2.99	98%	2.19	72%				
180.44	183.49	3.05	3.05	100%	0.52	17%			**core shattered**	
183.49	186.54	3.05	3.05	100%	1.38	45%			**core shattered**	
186.54	189.59	3.05	2.58	85%	1.32	43%			**core shattered**	
189.59	191.11	1.52	1.77	116%	1.32	87%				
191.11	194.16	3.05	2.76	90%	1.26	41%				
194.16	195.68	1.52	1.90	125%	0.83	55%			**core shattered**	
195.68	198.73	3.05	2.55	84%	1.14	37%				
198.73	201.78	3.05	2.93	96%	0.79	26%			**core shattered**	
201.78	204.83	3.05	3.02	99%	1.29	42%				
204.83	207.88	3.05	3.02	99%	1.86	61%				
207.88	210.92	3.04	3.02	99%	2.26	74%				
210.92	213.97	3.05	3.05	100%	1.34	44%			**core shattered**	
213.97	217.02	3.05	3.05	100%	0.14	5%			**core shattered**	

Renaissance Geoscience Services Inc- Geotechnical Logging Template

Hole:	D07-06	N:	4000	Logger:	DILLABOUGH		Dip:	-90		
Drill date:	9/27/2007	W	1795	Core size:	NTK		Azimuth:	0	Page: 1 of 1	
Log date:	9/29/2007	Elev.								
From	To	Length	Recovery	Percent	RQD	Percent	Fractures	Strength	Comments	Fracture
(m)	(m)	(m)	(m)	(%)	(m)	(%)	(per run)			Condition
12.50	15.54	3.04	0.00	0%	0.00	0%				
15.54	18.59	3.05	0.00	0%	0.00	0%				
18.59	21.64	3.05	0.25	8%	0.00	0%				
21.64	35.36	13.72	0.35	3%	0.20	1%			core shattered	
35.36	38.41	3.05	3.00	98%	1.91	63%			core crumbled	
38.41	39.01	0.60	0.51	85%	0.34	57%			core soft crumbled	
39.01	41.45	2.44	1.91	78%	1.01	41%			core crumbled	
41.45	42.98	1.53	1.47	96%	0.92	60%			core soft crumbled	
42.98	46.03	3.05	2.80	92%	1.48	49%			core soft crumbled	
46.03	47.85	1.82	2.21	121%	1.25	69%			core soft crumbled	
47.85	50.60	2.75	1.97	72%	1.56	57%			core crumbled	

Renaissance Geoscience Services Inc- Geotechnical Logging Template

Hole:	D07-07	N:	3950	Logger:	DILLABOUGH		Dip:	-90		
Drill date:	9/29/2007	E:	1500	Core size:	NTK		Azimuth:	0	Page: 1 of 1	
Log date:	10/1/2007	Elev.								
From	To	Length	Recovery	Percent	RQD	Percent	Fractures	Strength	Comments	Fracture
(m)	(m)	(m)	(m)	(%)	(m)	(%)	(per run)			Condition
6.00	6.40	0.40	0.40	100%	0.11	28%				
6.40	9.45	3.05	3.00	98%	1.40	46%				
9.45	12.95	3.50	3.43	98%	1.02	29%				
12.95	15.54	2.59	2.54	98%	0.81	31%			core shattered	
15.54	15.85	0.31	0.50	161%	-	0%			core crumbled	
15.85	17.37	1.52	1.45	95%	-	0%			core soft crumbled	
17.37	17.68	0.31	0.30	97%	-	0%			core crumbled	
17.68	20.42	2.74	3.23	118%	0.40	15%			core soft crumbled	
20.42	23.17	2.75	2.63	96%	0.76	28%			core soft crumbled	
23.17	24.99	1.82	1.73	95%	0.67	37%			core soft crumbled	
24.99	26.52	1.53	1.80	118%	0.23	15%			core crumbled	
26.52	28.35	1.83	2.01	110%	0.29	16%			core shattered	
28.35	30.18	1.83	2.03	111%	0.32	17%			core crumbled	
30.18	31.39	1.21	1.13	93%	0.23	19%			core soft	
31.39	32.61	1.22	1.22	100%	-	0%			core crumbled	
32.61	34.14	1.53	1.55	101%	-	0%			sandy	

Renaissance Geoscience Services Inc- Geotechnical Logging Template

Hole:	D07-08			Logger:	Brandon Barker					
Drill date:	Nov 28/2007			Core size:	NTK		Azimuth:	0	Dip:	-90
Log date:	Jan 20/2008	Elev.		N:	3950	W	1795			
From	To	Length	Recovery	Percent	RQD	Percent	Fractures	Strength	Comments	Fracture
(m)	(m)	(m)	(m)	(%)	(m)	(%)	(per run)			Condition
45.72	47.85	2.13	0.33	15%	0.00					
47.85	50.90	3.05	1.96	64%	0.00					
50.90	53.95	3.05	2.69	88%	0.00					
53.95	57.00	3.05	2.30	75%	0.00					
57.00	60.05	3.05	2.88	94%	0.41				gouge to 58.55	clay films
60.05	63.09	3.04	2.26	74%	0.16				60.95 gouge	
63.09	66.14	3.05	2.83	93%	0.46				63.93 fault	
66.14	69.19	3.05	1.46	48%	0.00				broken core	
69.19	72.14	2.95	2.44	83%	0.57				gouge at 69.29	
72.14	75.29	3.15	2.76	88%	1.30				broken core	
75.29	78.33	3.04	2.28	75%	0.19					
78.33	81.38	3.05	2.49	82%	0.59					
81.38	84.43	3.05	2.96	97%	0.28			2.0		
84.43	87.48	3.05	2.69	88%	0.77			1.0	86.38=grinding	
87.48	90.53	3.05	2.89	95%	0.14			0.0	ground material from faulting	
90.53	93.57	3.04	2.49	82%	0.69			2.0		
93.57	96.62	3.05	1.49	49%	0.41			2.0	missing core at 96.4m	
									grinding at 93.8, 96.18 and 96.46	
96.62	99.67	3.05	1.87	61%	1.20			2.0	grinding at 96.82, 95.62 and 99.53	
									broken core at 98.43	
99.67	102.72	3.05	1.95	64%	1.02			2.0	grinding at 101.69 and 101.56	
102.72	105.77	3.05	1.88	62%	0.14			2.0	grinding at 103.61, 103.67 and 105.60	
									various areas of broken core	
105.77	108.81	3.04	1.73	57%	0.72			1.0	very broken up core	
									113.99 missong core	
									grindng at 113.45 and 113.12	
114.91	119.96	5.05	1.70	34%	0.00			2.0	broken core	
117.96	121.01	3.05	2.26	74%	0.11			1.0	clay rich and fractures	
121.01	124.06	3.05	2.19	72%	0.55			2.0	ground powder	
									121.01-122.02	
124.06	127.10	3.04	2.74	90%	2.46			2.0	pieces missing at 120.37	
127.10	130.15	3.05	3.01	99%	2.86			2.0	nice core	
130.15	133.20	3.05	2.76	90%	1.37			2.0	132.56-133.20 fractured	
133.20	136.25	3.05	2.94	96%	2.37			2.0	missing core at 139.92m	
136.25	139.30	3.05	2.89	95%	2.19			2.0	140.44-142.06-142.82-139.51 pyrite	

Hole:	D07-08			Logger:	Brandon Barker				
142.34	145.39	3.05	2.76	90%	0.91			2.0	fracture core with some broken
145.39	148.44	3.05	2.71	89%	0.93				
148.44	151.49	3.05	2.69	88%	1.51			2.0	151.26 grinding
151.49	154.54	3.05	2.10	69%	1.47			2.0	152.69 broken core
154.54	157.58	3.04	3.08	101%	2.20			2.0	fault at 156.36
157.58	160.63	3.05	2.99	98%	1.49			2.0	broken core at 160.52
160.63	163.68	3.05	2.86	94%	1.56			2.0	160.83 broken core
163.68	166.73	3.05	3.04	100%	1.49			1.0	nice core
166.73	169.78	3.05	2.94	96%	2.14			1.0	
169.78	172.82	3.04	2.96	97%	2.57			2.0	some fractures
172.82	175.87	3.05	3.00	98%	2.41			2.0	nice core
175.87	178.93	3.06	3.05	100%	2.25			2.0	
178.93	181.97	3.04	2.96	97%	2.40			2.0	
181.97	185.02	3.05	3.03	99%	2.27			3.0	
185.02	188.06	3.04	2.97	98%	2.00			2.0	broken at 185.02
188.06	191.11	3.05	2.92	96%	0.40			2.0	lots of cracking (fractures)
191.11	194.16	3.05	3.01	99%	1.23			3.0	silica at 192.6
194.16	197.21	3.05	3.01	99%	0.53			2.0	fracturd w/ some gouge
197.21	200.26	3.05	3.00	98%	2.01			2.0	197.31 broken
200.26	203.30	3.04	2.95	97%	1.90			3.0	200.79 quartz
203.30	206.35	3.05	3.00	98%	1.46			3.0	204.3 quartz
206.35	209.40	3.05	2.94	96%	1.18			2.0	gouge at 208.96
209.40	212.45	3.05	3.03	99%	1.21			2.0	
212.45	215.50	3.05	3.04	100%	0.58			2.0	
215.50	218.54	3.04	2.98	98%	1.73			2.0	
218.54	221.59	3.05	2.78	91%	1.84			2.0	grind at 240.78
221.59	224.64	3.05	3.01	99%	0.76			2.0	lots of fractures
224.64	227.69	3.05	2.93	96%	0.81			2.0	
227.69	230.74	3.05	2.90	95%	0.75			1.0	
230.74	233.78	3.04	2.98	98%	1.79			2.0	
233.78	236.83	3.05	2.94	96%	1.45			2.0	236.61 gouge
236.83	239.88	3.05	2.95	97%	2.32			2.0	
239.88	242.93	3.05	3.00	98%	2.79			2.0	nice
242.93	245.98	3.05	3.05	100%	1.10			2.0	gouge at 245.58
									fractured
245.98	249.02	3.04	3.03	100%	0.21			1.0	most gouge
249.02	252.07	3.05	2.74	90%	2.20				251.81 broken core
252.07	255.12	3.05	3.04	100%	1.29			1.0	
255.12	258.17	3.05	3.05	100%	1.55			2.0	highly altered at 256.97
258.17	261.22	3.05	3.02	99%	1.66			2.0	all highly altered core
261.22	264.26	3.04	2.96	97%	2.73			2.0	highly altered

Hole:	D07-08			Logger:	Brandon Barker				
264.26	267.31	3.05	3.00	98%	2.85			2.0	highly altered w/ some areas
									with little to no alteration
267.31	270.36	3.05	3.03	99%	2.80			2.0	change of type at 268.13
270.36	273.41	3.05	3.04	100%	1.98			2.0	270.51 fault
									271.43 grinding
273.41	276.46	3.05	2.97	97%	2.63			2.0	highly fractured w/ some broken core
276.46	279.50	3.04	3.04	100%	2.64			1.0	gouge at 277.69
									broke core at 280.36
									hydrothermal deposit at 282.25
279.50	282.55	3.05	2.97	97%	2.63			2.0	
282.55	285.60	3.05	3.00	98%	2.79			1.0	altered hydrothermic
285.60	288.65	3.05	2.98	98%	2.08			2.0	altered hydrothermic
288.65	291.70	3.05	2.99	98%	2.64			2.0	bleached
291.70	294.75	3.05	3.05	100%	2.76			2.0	bleached
294.75	297.79	3.04	3.03	100%	2.91			2.0	missing piece at 294.75
297.79	300.84	3.05	3.05	100%	2.63			2.0	299.06 highly bleached white
300.84	303.89	3.05	3.00	98%	2.91			2.0	301.95 grinding
303.89	306.94	3.05	3.04	100%	2.66			2.0	grinding at 303.89
306.94	309.99	3.05	3.01	99%	2.54			2.0	not as altered (hornblendite)
309.99	313.03	3.04	3.02	99%	2.73			2.0	" "
313.03	316.08	3.05	3.05	100%	2.27			2.0	317.66 not as altered
									317.66 fault
316.08	322.18	6.10	2.90	48%	2.58			2.0	orange alteration
322.18	325.23	3.05	3.02	99%	2.43			2.0	partially beached
325.23	328.27	3.04	3.05	100%	2.60			2.0	change type at 328.27
328.27	331.32	3.05	1.32	43%	0.97			2.0	hematite w/ magnetite
									small viening a328.34t
									oxidized pyrite
									from 328.44 to 328.78

DO7-0004
1500N 1145W 930 M EL

QTIL

NEBA

FALZ

900.00

NEBA

NEAF

NEBA

850.00

FALT

NEBA

FALZ

TRACE DISSEMINATED NATIVE COPPER

NEBA

FALZ FALZ

TRACE DISSEMINATED PYRITE

KFPA

IP TARGET

800.00

FALZ

TRACE DISSEMINATED NATIVE COPPER

NEBA

KFPA

NEBA

KFHP

NEBA

750.00

KFHP

NEBA

25 METRES

1100 W

217.00

1200 W

LEGEND

QTIL Quaternary till

KFPA - Kamloops Group Feldspar porphyry
KFHP - Kamloops Gp. Feldspaqr hornblende porphry

NEBA - NICOLA EASTERN Gp. Basalt
NEAF - NICOLA EASGTERN Gp. Andesite flow

FALT - Fault < 30 cm
FALZ - Fault Zone > 30 cm

RENAISSANCE GEOSCIENCE SERVICES INC.

LAKEWOOD MINING LTD.
AFTON AREA CLAIMS
KAMLOOPS MINING DIVISION
DORADO ZONE
GEOLOGY
CROSS SECTION 1500 N
SECTION FACING 45 Az.

DATE	*18/APRIL/2008*
AUTHOR	*L. J. LINDINGER, P.Geo*
FIGURE	*7*
SCALE	*1:500*
PROJECTION	
FILE	

PROFESSIONAL SEAL

900.00
850.00
800.00
750.00
700.00
650.00
600.00
550.00
500.00

DO7-0007
[-50.00]

NEBB
FALZ
TANK
FALZ
NEBB
FALZ
TANK
NEBB
TANK

34.10
[-50.00]

[2.00]
DO7-06 DO7-08

OVERBURDEN

NEBB
FALZ
NEBB
NEBB
TANK
NEBB

NEBB
NEBB
FALZ
NEBB
FALZ
NEBB
TANK
FALZ
NEBB
NESS
KVBX
KVTF
KVTF
KVTF
KVLT
NEAT
KVLT
NEAT
NEAT
NEAT
FALZ
NEAT
KFPA
NEAT
NELT
FALZ
NEAT
NEAT
TANK
TANK
NEAT

331.30
[2.00]

TOP OF IP ANOMALY

LEGEND

QTIL Quaternary till

KVBX - Kamloops Gp. Volcanic Breccia
KVTF - Kamloops Gp. Volcanic tuff
KVLT - Kamloops Gp. Lapilli tuff
KFPA - Kamloops Group Feldspar porphyry
KFHP - Kamloops Gp. Feldspaqr hornblende porphry
NEBB - NICOLA EASTERN Gp. breccia
NEBA - NICOLA EASTERN Gp. Augite ppy Basalt
NEAF - NICOLA EASTERN Gp. Andesite flow
NEAT - NICOLA EASTERN Gp. Andesitic tuff
NELT -NICOLA EASTERN Gp. Lapilli tuff
NESS - NICOLA EASTERN GROUP - Siltstone
FALT - Fault < 30 cm
FALZ - Fault Zone > 30 cm

0 25 METRES

50.00 100.00 150.00 200.00 250.00 300.00 350.00

RENAISSANCE GEOSCIENCE SERVICES INC.

LAKEWOOD MINING LTD.
AFTON AREA CLAIMS

KAMLOOPS MINING DIVISION
DORADO ZONE
GEOLOGY
CROSS SECTION 3950 N
SECTION FACING 45 Az.

DATE	18/APRIL/2008
AUTHOR	L. J. LINDINGER, P.Geo
FIGURE	8A
SCALE	1:500
PROJECTION	
FILE	

PROFESSIONAL SEAL

900.00

850.00

800.00

750.00

700.00

650.00

600.00

550.00

500.00

DO7-06 [2.00] **DO7-05** **DO7-08**

DO7-07 [-50.00]

OVERBURDEN

TOP OF IP ANOMALY

D07-06 RESULTS

Au	As		Sb	Cu
2	2		2	73
1	9		2	44
5	6		2	44
3	5	42.00 [2.00]	2	42
4	9		2	57
3	10		2	67
		50.60 [2.00]		

Au	As		Sb	Cu
4	6		2	110
1	26		2	59
3	9		2	65
2	15		2	91
2	86	538	2	78
2	28		2	65
3	10		2	80
2	15		2	68
				63
		34.10 [-50.00]		

Au	As		Sb	Cu
3	10	130	3	176
3	15		3	53
5	15		3	50
25	0		3	48
3	15		3	92
3	10		3	53
3	10		3	45
5	10		3	88
5	70		5	34
5	5		3	9
30	10		3	39
10	3		3	27
5	3		3	31
5	25		3	45
10	110		10	49
25	15		3	25
30	3		3	15
10	3		3	24
10	5		3	40
10	5		3	113
5	10		3	96
25	15		3	98
25	10		3	67
60	20		5	132
20	5		3	113
5	15		5	179
10	20		3	62
20	20		3	81
15	20		3	110
5	190		20	159
20	50		3	127
40	30		10	153
10	95		5	130
10	375		20	134
5	135		5	118
30	60		3	131
30	335		15	182
15	235		15	197
10	970		75	161
10	850		55	183
25	270		15	219
25	330		15	199
15	510		20	179
5	500		20	130
10	155		5	100
10	45		5	110
50	90		5	85
15	20		3	189
15	15		3	110
10	15		3	121
5	25		3	133
15	20		3	64
35	30		3	85
5	15		3	26
3	10		3	7
10	15		3	89
10	25		3	52
15	15		3	133
20	20		3	135
20	20		3	138
15	15		3	37
10	20		3	137
25	15		3	123
10	30		3	99
15	35		3	103
20	35		3	98
10	35		3	109
10	35		3	100
10	20		3	72
5	20		3	71
10	25		3	78
30	15		3	62
20	30		3	94
10	25		3	86
5	15		3	67
5	10		3	219
10	10		3	88
		331.32 [2.00]		

50.00 100.00 150.00 200.00 250.00 300.00 350.00

DRILL TRACE

Au ppb	As ppm	Sb	Cu ppm
>0 to <25	>0 to <100	>0 to <40	>0 to <200
>=25 to <50	>=100 to <200	>=40	>=200
>=50	>=200 to <=500		
	>=500		

0 25 METRES

GEOLOGICAL SURVEY BRANCH ASSESSMENT REPORT 29,230

RENAISSANCE GEOSCIENCE SERVICES INC.

LAKEWOOD MINING LTD.
AFTON AREA CLAIMS
KAMLOOPS MINING DIVISION
DORADO ZONE
GOLD, ARSENIC, ANTIMONY AND COPPER ANALYTICAL RESULTS
CROSS SECTION 3950 N
SECTION FACING 45 Az.

DATE	18/APRIL/2008
AUTHOR	L. J. LINDINGER, P.Geo
FIGURE	8B
SCALE	1:500
PROJECTION	
FILE	

PROFESSIONAL SEAL