Report on
Trenching, Drilling and Metallurgical Testing

on the

Congress Property

Lillooet Mining Division
British Columbia
Canada

N.T.S.: 092 J/15W

UTM co-ord.: 514,000 m E, 5,638,000 m N
UTM Zone 10

Owner/Operator:
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Summary

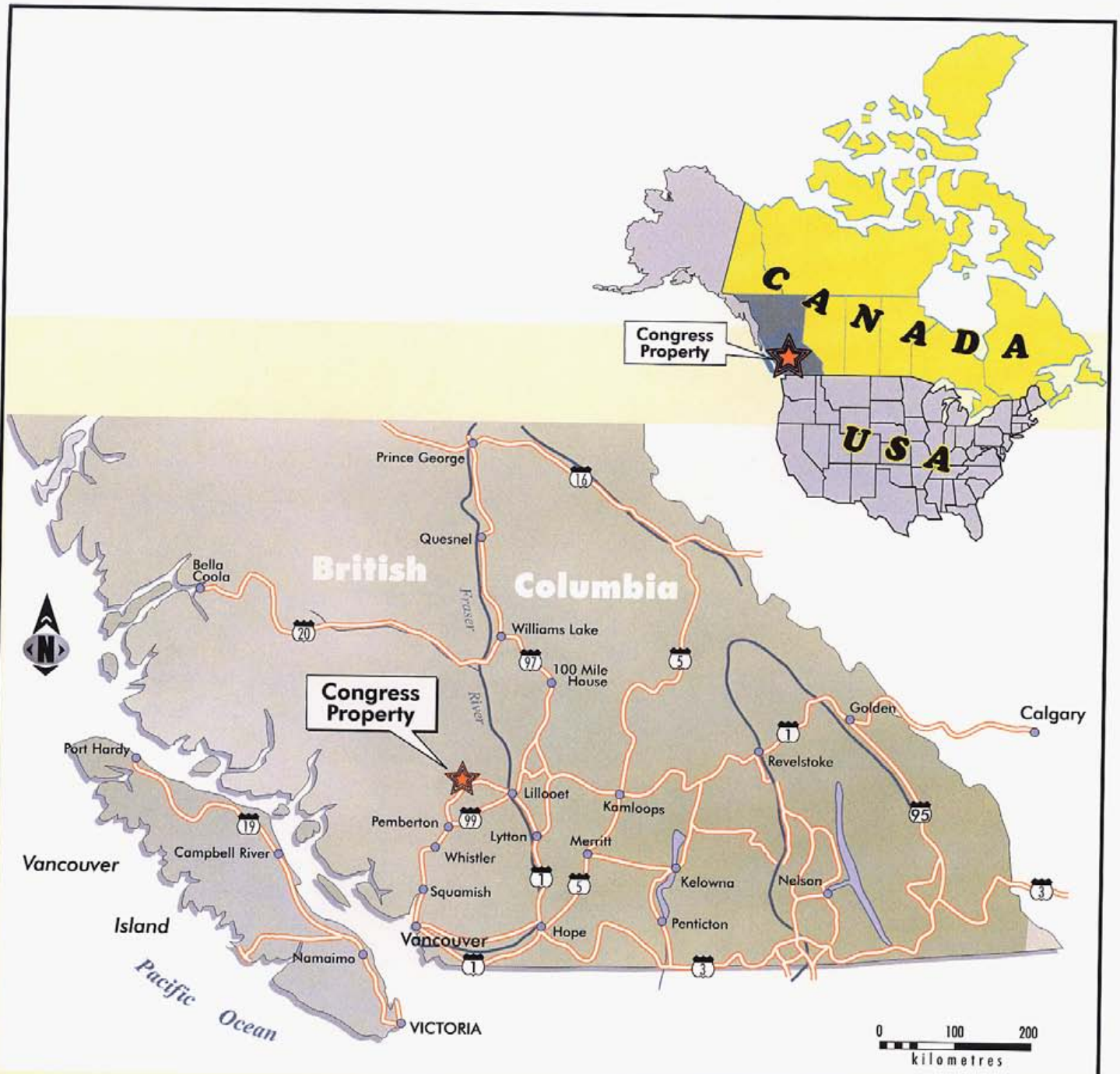
The Congress Property ("the property") is located on the north side of Carpenter Lake, 90 kilometers west of the town of Lillooet (Fig. 1 & 2). The property can be accessed by automobile from Lillooet by taking B. C. Highway 40 for 96 kilometers west to the property.

The property consists of one reverted crown granted mineral claim, 3 mineral leases and 11 mineral claims totaling 2432.756 hectares (Fig. 2, Table 1) located on the north side of Carpenter Lake 4 kilometers northeast of Goldbridge in the Lillooet Mining Division, NTS 092J15W. The property is owned by Levon Resources Ltd. ("the company") and Veronex Resources Ltd. The property is easily accessible by automobile on the Goldbridge to Lillooet road, B. C. Highway 40, which crosses the southern part of the property. The Slim Creek forest access road, which turns off the highway on the property and crosses the property in a northwesterly direction, and numerous access trails and roads built on the property during previous exploration programs provide good access to the rest of the property (Map 1).

The property covers Mississipian to Middle Jurassic rocks of the Bridge River Complex, mainly submarine basalt and andesite with minor chert, argillite and mafic intrusives (Fig. 3). These rocks are cut by northwest trending regional scale structures, in some cases with contained Tertiary feldspar porphyry dacite dykes, sub-parallel to the Ferguson and Cadwallader Structures. The structures on the property are roughly the same distance from the Upper Cretaceous-Tertiary granitic Bendor Intrusions as the Bralorne/Pioneer mines. The Bendor Intrusions are a postulated source for the gold mineralization at the Bralorne mine.

The structures on the property are mineralized with gold and silver in quartz- carbonate veins and in altered vein selvages for up to 5 metres from the veins. These veins have received considerable past work, including 6 adits with more than 2,235 metres of underground workings (Map 1). The following resources have been developed:

	Tonnes	oz/ton	g/tonne	Mineral Resource Category
Howard	273,402.5	0.264	8.2	inferred
Howard	25,909	0.367	11.4	indicated
Howard	40,192	0.280	9.68	measured
Lou underground	189,548	0.350	10.9	inferred
Lou open pit	124,300	0.077	2.4	inferred
Congress	106,678	0.238	7.4	indicated



Levon Resources Ltd.

CONGRESS PROJECT

Lillooet M.D. British Columbia

General Location Map

Scale	as shown	UTM	NAD83 Zone 10	Fig 1
Date	April 2006	TRIM	92J.086,087,096,097	
By	DSD/AGB	NTS	92J/15	

These resources were outlined in the 1930's, 1950's, 1960's and 1980's but were not mined because of the refractory nature of the mineralization. In the Howard Zone, most of the gold is contained in fine grained arsenopyrite, which is intimately associated with quartz-ankerite gangue. The best recovery by cyanide with a very fine grind has been just over 20%. Flotation has been more successful, with the results from the 2004 testing being 91% gold recovered in 52.5% of feed. Metallurgical testing was carried out by Process Research Associates Ltd. Oxidizing the sulphides using a bio-leaching or pressure leaching system was recommended as the best approach to maximize gold recovery. Tests are presently ongoing with the Research and Productivity Council in New Brunswick to test the cost effectiveness of oxidizing the mineralization using bulk microwave technology.

Numerous other showings including the Ozone, Gun, Slide, Paul and Golden Ledge exist on the property, mainly in the Gun Creek canyon. These are generally narrow structures with erratic orientations very disrupted by the major structures following the Gun Creek canyon. No NI 43-101 quantifiable resources have been developed in these showings to date but further work is recommended, particularly on the Golden Ledge.

Further work to increase and upgrade the mineral resources on the property, including surface diamond drilling and underground development on the Lou and Howard Zones, is recommended.

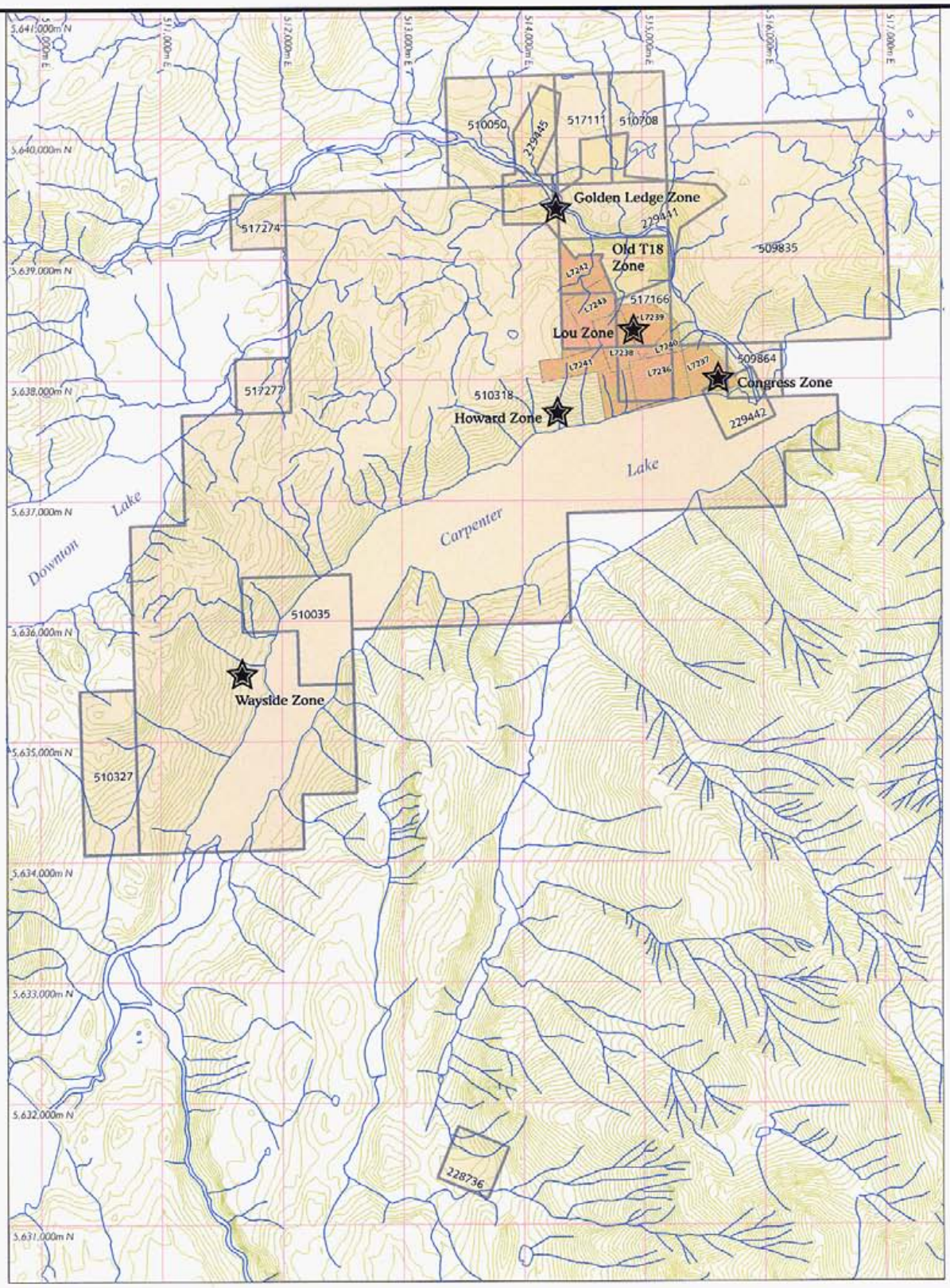
Introduction and Terms of Reference

The author was commissioned by Mr. Louis Wolfin, President and CEO of Levon Resources Ltd., to review historic work on the Congress Property then plan and implement a mineral exploration program designed to increase the known mineral resources on the property. This program and historic work were then to be documented by a NI 43-101 compliant report.

The author supervised the exploration program on the Congress Property. The program commenced on April 25th, 2005 and continued until August 10th, 2005. The author was on site April 26th to May 3rd, May 20th to 24th, June 6th to 11th, August 4th to 9th, 2005.

Disclaimer

This report relies heavily on information on historic work supplied by the company. This work was carried out by mineral exploration professionals known to the author to be reputable and is deemed reliable.



0 0.5 1 1.5 2 2.5

kilometres

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Lillooet M.D. British Columbia

Claim Location Map

Scale	1:50,000	UTM	NAD27 Zone 10	Fig 2
Date	April 2006	TRIM	92J.086,087,096,097	
By	DSD/AGB	NTS	92J/15	

Property Description and Location

The property consists of one reverted crown granted mineral claim, 3 mineral leases and 11 mineral claims totaling 109 cells covering approximately 2432 hectares as listed in Table 1 below:

Table 1: List of Mineral Claims

Tenure No	Claim Name	Map No	Good To	Mining Div	Area
228736	REFER TO LOT TABLE	092J	2008/DEC/25	LILLOOET	25.000
229441		092J	2005/DEC/18	LILLOOET	116.130
229442		092J	2006/OCT/26	LILLOOET	21.070
229445		092J	2006/SEP/20	LILLOOET	48.280
509835		092J	2012/DEC/25		326.113
509864		092J	2012/OCT/13		20.387
510035		092J	2013/DEC/08		61.183
510050		092J	2012/OCT/13		81.512
510318		092J	2012/DEC/25		1488.461
510327		092J	2006/MAR/30		61.191
510708		092J	2012/DEC/25		40.756
517111		092J	2006/JUL/12		40.756
517166		092J	2006/JUL/12		61.151
517274		092J	2006/JUL/12		20.380
517277		092J	2006/JUL/12		20.386
					2432.756

The reverted crown granted mineral claim is treated the same as a mineral claim cell. These claims are kept in good standing by paying $200 per cell or carrying out and documenting $200 in work per cell in the claim block per year. Present expiry dates are recorded in the claim list. The mineral leases are kept in good standing by paying rental fees totaling $1,854.80 per year on the dates shown in the List of Mineral Claims. All claims and leases are contiguous.

The claims are located on National Topographic System map 92J/15W in the Bridge River mining camp, Lillooet Mining Division, British Columbia, Canada (Figs. 1 & 2). The main showings on the property are located around UTM co-ordinates 515,000mE and 5,637,750mN. *(NAD27 zone 10)*

The property is owned 50% by the company and 50% by Veronex Resources Ltd.

Accessibility, Climate, Local Resources, Infrrastructure And Physiography

The property is easily accessible from Vancouver by all weather government maintained roads. Proceed north from Vancouver on paved Highway 99 through Squamish, Whistler and Pemberton 233 kilometres to Lillooet, then proceed west 96 kilometres on Highway 40 to the property (Fig. 1). Highway 40 is approximately 80% paved from Lillooet to the property and is maintained and ploughed year round, mainly for logging and tourist access. This route takes approximately 5.5 hours driving time. Alternatively, in spring summer and fall, it is possible to drive to Pemberton on Highway 99 then northwest 20 kilometres to Pemberton Meadows and northeast 50 kilometres over the gravel Hurley River Forest Access Road to the property. This route takes approximately 4.5 hours driving time from Vancouver, but the road is not ploughed in the winter. All services necessary to operate a mine are available in Lillooet or Pemberton.

The property lies on the boundary between West Coast Marine and Interior climatic zones and is in the rain shadow created by the Coast Mountains. Precipitation is moderate, with generally warm, dry summers. Moderate to heavy snowfall occurs in winter months, with accumulations exceeding 2 metres on the property. Surface work is generally curtailed during winter months due to these conditions.

The town of Bralorne lies 16 kilometres south of the property. This town was built to support historic mining operations and had a population of approximately 5,000 during historic operations. There are approximately 50 full time residents now and over 100 structurally sound houses in the town. A restaurant operates year round in the town. The town of Goldbridge lies 5 kilometers southwest of the main showings on the property. Goldbridge has an area population of approximately 200. There are two motels, a restaurant, gas station, grocery store, and kindergarten to grade 7 school in Goldbridge. There are some trained miners living in the Bridge River Valley.

The town of Bralorne and Goldbridge are connected to the B.C. electric power grid. The Lajoie Dam and power generation facility on Downton Lake, operated by B.C. Hydro, are located approximately 4 kilometers west of the main showings on the property. A high tension power line follows Highway 40 across the property.

The property lies in mountainous terrain with deeply incised stream valleys and moderate to steep slopes. The property covers a plateau north of Carpenter Lake and a steep cliff and talus covered slope extending down to the lake edge. The Gun Creek canyon, 100 to 200 metres wide and 100 to 200 metres deep, crosses northwesterly across the northeastern quadrant of the property. Elevations range from 655 metres on Carpenter Lake in the south part of the property to 1035 metres on the plateau in north central part of the property. Vegetation on the property consists of mature spruce, pine and interior fir. Approximately 60% of the property has been clear cut.

The local population is generally pro-mining and would like to see a mine developed for the benefits it would generate for the local communities.

Sufficient water for mining and milling purposes is available from Gun Creek in the eastern part of the property and a number of ponds and swamps on the plateau part of the property.

History

The Congress Zone was discovered in 1913 and has been explored and mined intermittently since then. Significant periods of activity occurred in 1933, when a 1,000 ton bulk sample was mined for metallurgical tests, and 1945-1950, when the vein was developed on 5 underground levels and some mineralized material stoped.

The Howard Zone was discovered in 1959 and explored by Bralorne-Pioneer Mines Ltd. who put in approximately half of the Lower Howard workings between 1960 and 1964. Levon Resources Ltd. carried out surface and underground drilling and drifting between 1976 and 1988 when the rest of the Lower Howard and the Upper Howard workings were excavated.

The Lou Zone was discovered following up on soil geochemical anomalies and VLF-em geophysical anomalies in 1984. Extensive surface drilling was carried out from 1984 to 1988 and a 300 metre trackless decline was driven in the footwall of the zone in 1989.

Significant work was suspended until 2004 because of low gold prices. A mechanized trenching program on the northern extensions of the Lou and Congress zones was carried out in the fall of 2004. A diamond drill program was carried out on the Howard Zone in December 2004 and January 2005.

The 2004 surface exploration program consisted of approximately 120 metres of mechanized trenching in 6 trenches and 4 NQ diamond drill holes totaling 820.5 metres. The trenches were targeted at new mineral occurrences uncovered by logging activity in the central part of the property and on historic soil geochemical anomalies on strike with the projected northern extensions of the Lou and Congress zones (Map 1). Drilling was targeted at better defining the Howard Zone north of the face of the Lower Howard drift (Map 2).

Trenches 1, 2 and 3 in the central part of the property did not return any values of economic interest. Trenches 4 and 5 (Map 1) were dug at the break in slope west of Gun Creek on historic high gold and arsenic soil geochemical anomalies. They cut a massive stibnite vein, probably the northern extension of the Congress Zone, more than 250 metres north of the most northerly mapped outcrop of the Congress Zone. Trench 6 was a western extension to 1988 Trench 18 and exposed the northern extension of the Lou Zone, a further 175 metres north of its previously most northerly known exposure.

Four drill holes totaling 820.5 metres, C-04-01, 02, 03 and C-05-04, were drilled from the same set-up, two at -60° and two at -80° (Map 2). All four holes intersected at least one of the Howard Zones over wide intervals. The intersections most mineralized in gold are shown below:

Table 2: List of Mineralized Diamond Drill Hole Intersections

Drill Hole	Intersection (metres)	Estimated True width	Grade Gold g/t	Zone
C-04-01 (-60°)	135.2-137.2	1.85m	3.4	West Howard
	153.2-154.2	0.93m	2.2	Howard
	154.2-155.2	0.93m	3.7	Howard
	155.2-156.7	1.39m	0.9	Howard
	224.0-225.5	1.39m	1.9	East Howard
	225.5-227.0	1.39m	0.3	East Howard
	227.0-228.5	1.39m	0.9	East Howard
	228.5-229.5	0.93m	Trace	East Howard
	229.5-231.0	1.39m	0.6	East Howard
	255.4-256.0	0.56m	2.5	Far East Howard
C-04-02 (-60°)	141.8-142.4	0.56m	1.2	West Howard
	154.0-155.5	1.39m	1.2	Howard
	155.5-157.0	1.39m	2.4	Howard
	157.0-158.5	1.39m	0.9	Howard
	158.5-159.5	0.93m	2.4	Howard
	159.5-160.7	1.11m	0.3	Howard
	160.7-162.1	1.30m	1.5	Howard
	162.1-164.3	2.04m	1.5	Howard
C-04-03 (-80°)	149.4-150.3	0.60m	0.2	West Howard
	150.3-151.5	0.80m	1.13	West Howard
	151.5-153.0	1.00m	0.03	West Howard
	153.0-154.1	0.74m	0.37	West Howard
	154.1-154.7	0.40m	0.10	West Howard
	154.7-155.5	0.54m	1.33	West Howard
	155.5-156.8	0.87m	0.30	West Howard

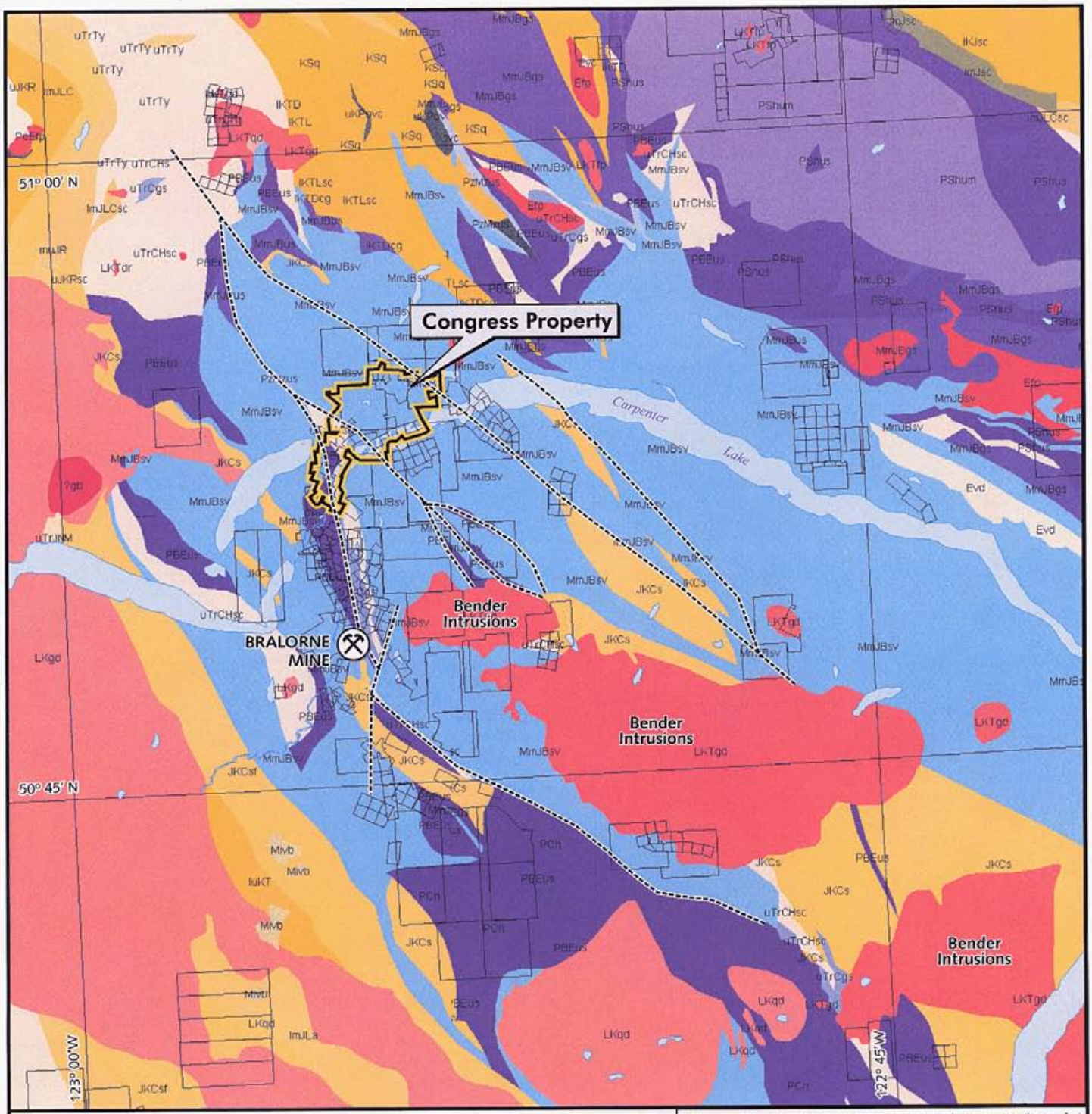
Table 2: List of Mineralized Diamond Drill Hole Intersections (cont.)

Drill Hole	Intersection (metres)	Estimated True width	Grade Gold g/tonne	Zone
C-04-03 (-80°)	166.7-167.3	0.40m	1.13	West Howard
	176.0-176.4	0.27m	0.07	Howard
	176.4-177.2	0.54m	0.13	Howard
	177.2-178.0	0.54m	12.14	Howard
C-05-04 (-80°)	152.1-153.0	0.60m	7.93	West Howard
	153.0-154.5	1.00m	0.30	West Howard
	154.5-156.0	1.00m	0.60	West Howard
	156.0-156.8	0.54m	0.40	West Howard
	156.8-158.3	1.00m	0.30	West Howard
	158.3-159.0	0.47m	0.03	West Howard
	159.0-160.2	0.80m	0.30	West Howard
	160.2-161.6	0.97m	1.37	West Howard

These drill holes show that there are 4 Howard Zones, en echelon, designated here West Howard, Howard, East Howard and Far East Howard. Bralorne Mines Ltd. initially developed the East Howard and Far East Howard in 1959. Levon's development in the 1980's was largely on the Howard. The West Howard is a blind vein discovered by the 2004/2005 drill program. These zones strike north-south and are 20 to 30 metres apart in an east-west direction. The more mineralized sections of the zones are further to the north in each more westerly zone. The zones exceed 10 metres true width in the widest mineralized areas.

Geological Setting

The property covers Mississippian to Middle Jurassic rocks of the Bridge River Complex, mainly submarine basalt and andesite, with minor chert, argillite and mafic intrusives (Fig. 3, Map 1). These rocks are cut by northwest trending regional scale structures, some with contained Tertiary feldspar porphyry dacite dykes, sub-parallel to the Ferguson and Cadwallader Structures, which bound the historic Bralorne/Pioneer mines. The structures on the property are roughly the same distance from the Upper Cretaceous-Tertiary granitic Bendor Intrusions as the Bralorne/Pioneer mines. The Bendor Intrusions are the same age as the mineralization in the Bralorne/Pioneer mines and are a postulated source for the gold mineralization at these mines and on the Congress Property.



(source: http://webmap.em.gov.bc.ca/mapplace/minpot/bcgs.cfm)

Lithological Legend

Mivb	Miocene unnamed: basaltic volcanic rocks
Efd	Eocene: unnamed feldspar porphyritic intrusive rocks
Egd	Eocene: unnamed granodioritic intrusive rocks
Evd	Eocene: unnamed dacitic volcanic rocks
LKTgd	Late Cretaceous to Paeleogene: unnamed granodioritic intrusive rocks
LKgd	Late Cretaceous: unnamed granodioritic intrusive rocks
LKqd	Late Cretaceous: unnamed quartz granodioritic intrusive rocks
uKPo	Upper Cretaceous: Powell Creek Formation - undivided volcanic rocks
lKTs	Lower Cretaceous: Taylor Creek Group - undivided sedimentary rocks
luKT	Lower Cretaceous to Upper Cretaceous: unnamed volcaniclastic rocks
uJKRsc	Upper Jurassic to Lower Cretaceous: Relay Mountain Group - undivided sedimentary rocks
JKCsf	Jurassic to Cretaceous: Cayoosh Assemblage - undivided sedimentary rocks
uTrTy	Upper Triassic: Tyaughton Group - conglomerate, coarse clastic sedimentary rocks
uTrCHsc	Upper Triassic: Cadwallader Group, Hurley Formation - undivided sedimentary rocks
MmJBsv	Mississippian to Middle Jurassic Bridge River Complex: marine sedimentary and volcanic rocks
MmJBgs	Mississippian to Middle Jurassic Bridge River Complex: greenstone, greenschist metamorphic rocks
MmJBbs	Mississippian to Middle Jurassic Bridge River Complex: blueschist metamorphic rocks
PBEus	Permian: Bralorne East Liza Complex: serpentinite ultramafic rocks
PShum	Permian: Shulaps Ultramafic Complex - Hartzburgite Unit: ultramafic rocks
PShus	Permian: Shulaps Ultramafic Complex - Serpentenite Melange Unit: serpentenite ultramafic rocks
PzMzus	Paeleozoic to Mesozoic unnamed serpentenite ultramafic rocks
?gb	Age Unknown: gabbroic to dioritic intrusive rock

Symbol Legend

geological fault

0 5 10
kilometres

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Lillooet M.D. British Columbia

Regional Geology Map

Scale	as shown	UTM	NAD83 Zone 10	Fig 3
Date	April 2006	TRIM	92J.086,087,096,097	
By	DSD/AGB	NTS	92J/15	

Deposit Types

The deposits on the company's property are members of a well recognized group of deposits referred to as mesothermal, orogenic or greenstone hosted quartz-carbonate gold vein deposits. These deposits include the Mother Lode and Grass Valley districts in California and most of the greenstone hosted gold deposits in the Canadian shield, including the Timmins-Val d'Or, Red Lake and Hemlo camps. These deposits are quartz-carbonate veins in moderately to steeply dipping brittle-ductile shear zones and, locally, in shallow dipping extensional fractures.

Mineralization

Mineralization in the Howard Zones consists of quartz-carbonate veins or stringer zones one to 1.5 metres wide, with altered, mineralized selvages (pyrite, siderite) up to 10 metres total width hosted in basalt and gabbro. The zones strike north to a few degrees west of north and dip steeply to the west. The Howard Zones contain the largest and highest grade resource on the property, with over 100,000 ounces of gold contained in all resource categories totaling more than 300,000 tonnes greater than 10 grams per tonne gold. These resources are refractory and would require oxidation of sulphides to recover the gold.

Mineralized areas in the Lou Zone are stockwork quartz carbonate stringers and silicified zones on the flank of a feldspar porphyry dyke hosted in mafic volcanics. The zone strikes north and dips steeply west. The better mineralized zones are 1.5 to 4.0 metres wide and grade 5 to 11 grams gold/tonne and contain abundant stibnite. The Lou Zone has been oxidized for 2 to 5 metres below surface near the decline portal where an open pit resource has been outlined.

The better mineralized areas in the Congress Zone, including the 2004 trenches, are massive stibnite veins, 1.25 to 1.5 metres wide, grading 6 to 8 grams gold per tonne hosted in argillite, chert and very sheared mafic volcanic rocks and again, striking north and dipping steeply west.

Exploration

Exploration carried out by the company and documented in this report consisted of 2 pits, 27 trenches totaling approximately 300 metres, 6 NQ diamond drill holes totaling 1,060.68 metres and 102 MMI geochemical samples in seven lines.

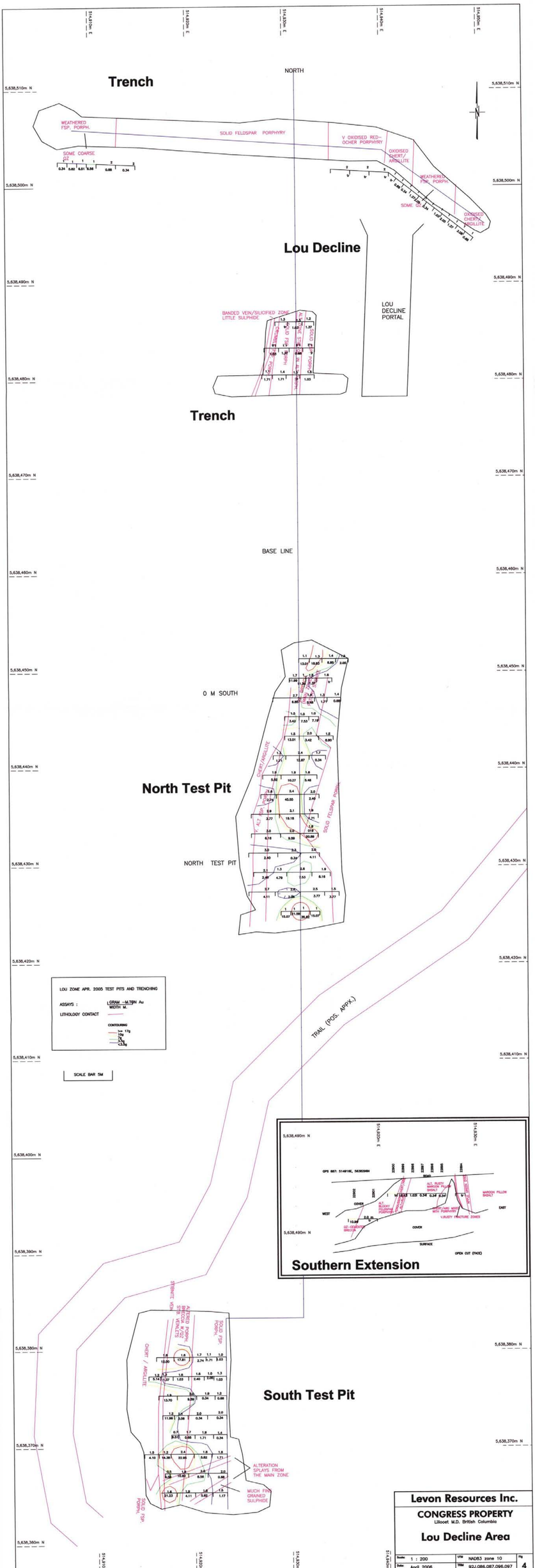
The pits were dug on the Lou Zone to strip 2 areas approximately 10 metres by 30 metres and 10 metres by 20 metres of the zone and carry out detailed sampling (Fig. 5). The pits were dug south of the Lou Decline portal to gain access to the proposed open pit resource to test its metallurgical properties. Results were much higher than historic drilling under this area indicating some surface enrichment. Metallurgical tests on this material indicate it responds very well to cyanide leaching. Further stripping to expose more of this material is recommended.

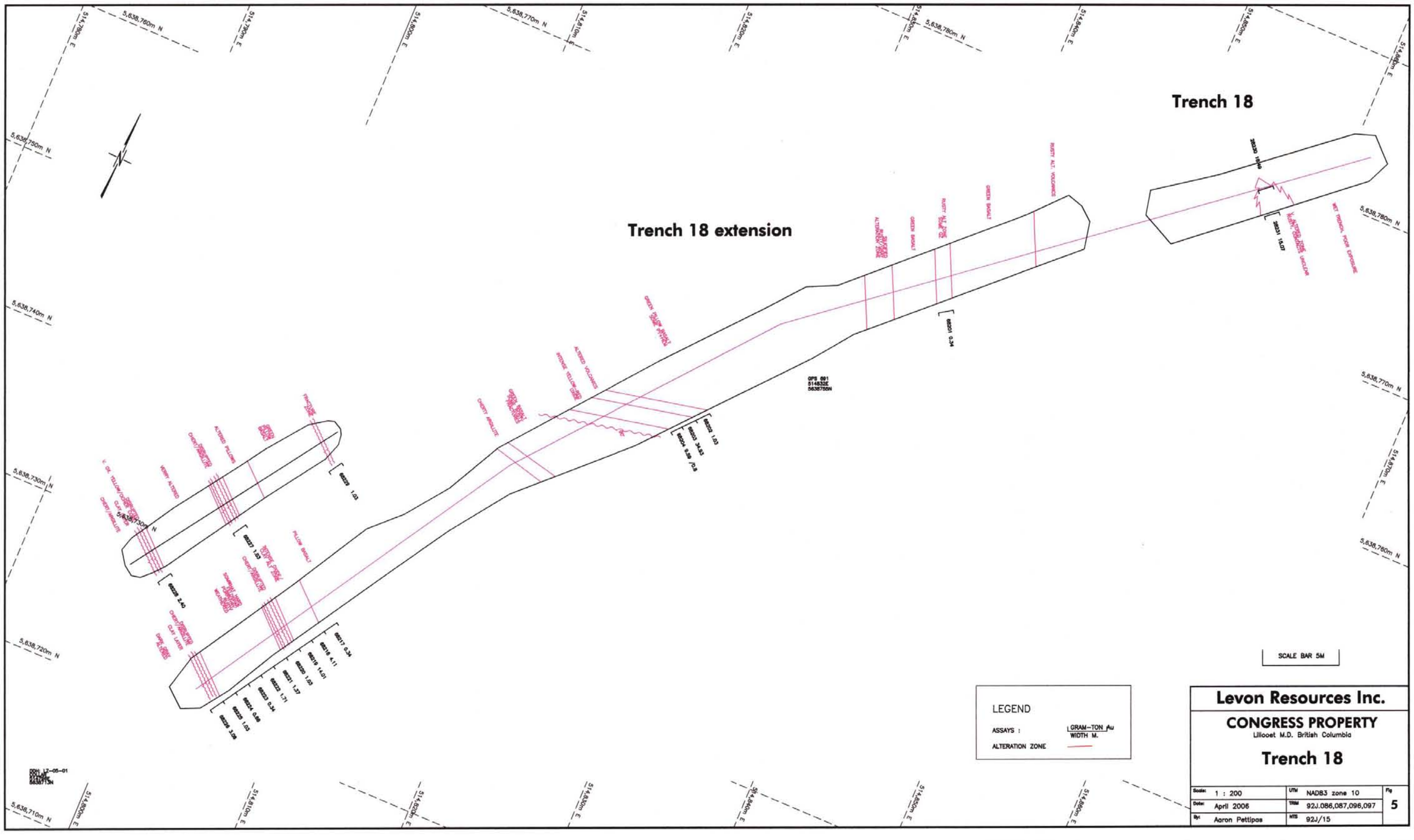
The trenches were excavated in the northern part of the property, north of the LouZone, in an attempt to extend that zone to the north and on a new showing, "Golden Ledge", in the Gun Creek Canyon north of the Howard Zone. The excavator was unable to reach bedrock in many of the Lou Zone trenches north of historic Tr 18 and Tr-6-04 due to deep overburden. Where bedrock was reached unmineralized maroon and green andesite volcanics were exposed. Trenching on the Golden Ledge exposed a 1.0 to 1.5 metre wide silicified fault zone striking 116° and dipping 70° to the west with gold values up to 26.4 grams per tonne gold over 1.2 metres.

A Mobile Metal Ion (MMI) geochemical survey totaling 102 samples was carried out. This type of survey measures the weakly attached cations in the near surface environment and has been used to successfully outline mineralization in areas with tens of metres of overburden, well beyond the limits of traditional soil geochemical sampling.

An orientation survey was carried out over a mineralized part of the Lou Zone with approximately 4 metres of overburden where traditional soil sampling had not detected the mineralization. MMI samples were taken every 10 metres along a line at 90° to the Lou Zone. The 2 samples that were taken directly above the Lou Zone returned highly anomalous values in Au, Ag, As, Cu, and Sb with moderate depletion in Pb and Zn. Au was more than 1,000 times background, Ag 30 times background, As 30 times background, Cu 7 times background and Sb 30 times background. Pb showed a relative depletion of 10 times to background and Zn also showed a depletion of 10 times background. This orientation survey shows that the MMI technique can easily detect Lou Zone mineralization through 4 to 5 metres of overburden.

Three lines were sampled at approximately 100, 200, and 300 metre intervals north of the Tr 18 area across the projected extension of the Lou Zone. Gold and supporting accompanying elements clearly outline the northern extension of the zone with a very strong and wide anomaly on the most northerly line.


Trench
NORTH
WEATHERED FSP. PORPH.
SOLID FELDSPAR PORPHYRY
V OXIDISED RED-OCHER PORPHYRY
OXIDISED CHERT / ARGILLITE
WEATHERED FSP. PORPH.
SOME COARSE QZ
OXIDISED CHERTY ARGILLITE
Lou Decline
LOU DECLINE PORTAL
BANDED VEIN/SILICIFIED ZONE LITTLE SULPHIDE
Trench
BASE LINE
0 M SOUTH
North Test Pit
NORTH TEST PIT
CHERT/ARGILLITE
SOLID FELDSPAR PORPHY
LOU ZONE APR. 2005 TEST PITS AND TRENCHING
ASSAYS : LGRAM ~M.TPN Au WIDTH M.
LITHOLOGY CONTACT
CONTOURING
SCALE BAR 5M
TRAIL (POS. APPX.)
Southern Extension
COVER
WEST
EAST
SURFACE
OPEN CUT (PACE)
South Test Pit
CHERT / ARGILLITE
SOLID FSP PORPHY
ALTERATION SPLAYS FROM THE MAIN ZONE
MUCH FINE GRAINED SULPHIDE
Levon Resources Inc.
CONGRESS PROPERTY
Lillooet M.D. British Columbia
Lou Decline Area
Scale: 1 : 200
Date: April 2006
By: Aaron Pettipas
UTM NAD83 zone 10
TRIM 92J.086,087,096,097
NTS 92J/15
Fig 4



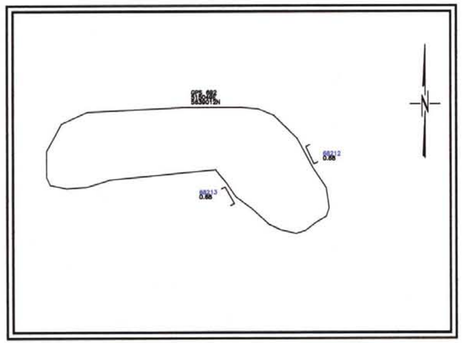
Levon Resources Inc.

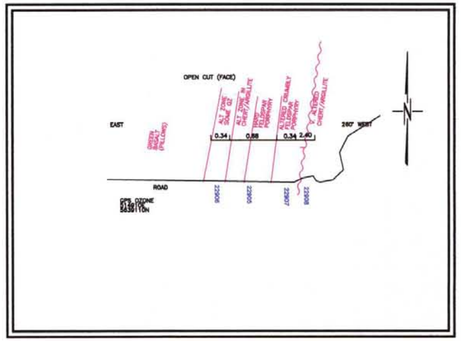
CONGRESS PROPERTY

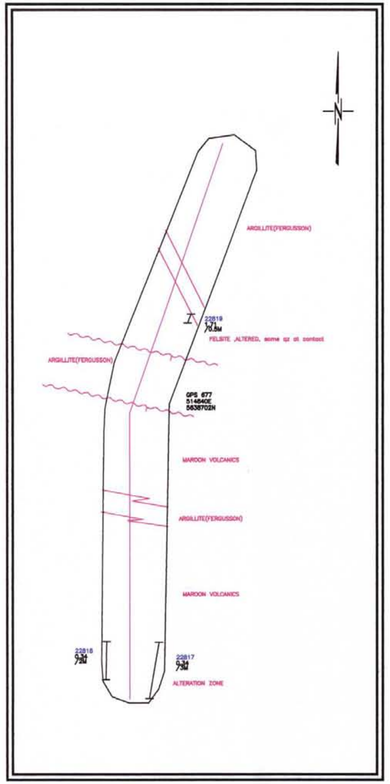
Lillooet M.D. British Columbia

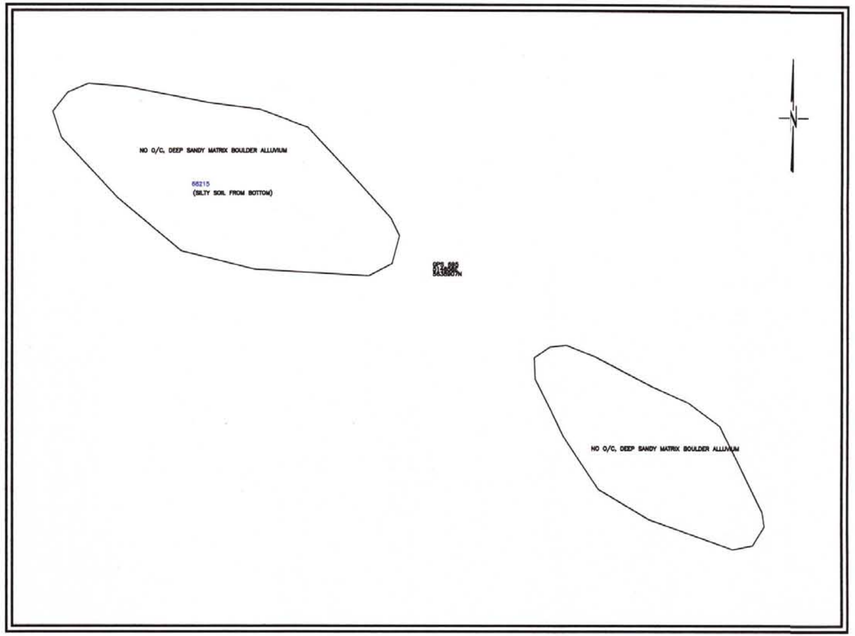
Trench 18

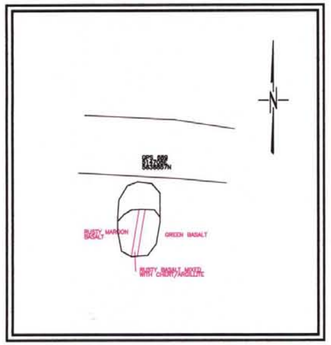
Scale: 1 : 200	UTM	NAD83 zone 10	Fig 5
Date: April 2006	TRIM	92J.086,087,096,097	
By: Aaron Pettipas	NTS	92J/15	

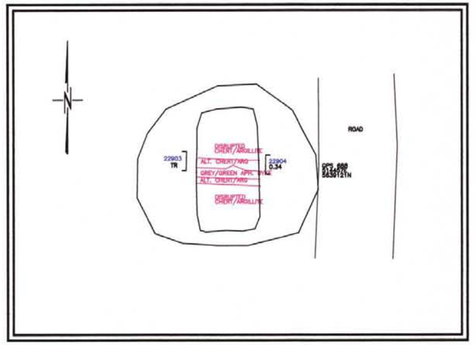












SCALE BAR 5M

LEGEND

ASSAYS : GRAM-TON Au / WIDTH M.

ALTERATION ZONE

Levon Resources Inc.

CONGRESS PROPERTY

Lillooet M.D. British Columbia

Old T18 Zone Trenching

Scale:	1 : 200	UTM	NAD83 zone 10	Fig
Date:	April 2006	TRIM	92J.086,087,096,097	6
By:	Aaron	NTS	92J/15	

Three lines were also sampled at 200 metre intervals across the projected northern extension of the Howard Zone. No clear anomaly outlining the Howard Zone extension was seen. A weak arsenic anomaly trending 05° might reflect the zone.

Drilling

A program of 6 NQ diamond drill holes totaling 1060.68 metres was carried out on the property in the spring and summer of 2005.

The first hole of the summer 2005 drill program was drilled at -45° Inclination, Azimuth 190° under the Golden Ledge showing (Map 1). This hole intersected 3.9 metres of 5.14 grams gold per tonne. The zone appeared to be dipping to the south at 70° where intersected. The true width of the zone is estimated to be 1.65 metres where intersected. This zone might be a rotated extension of the Howard Zone. More drilling to better define and quantify resources in this zone is recommended.

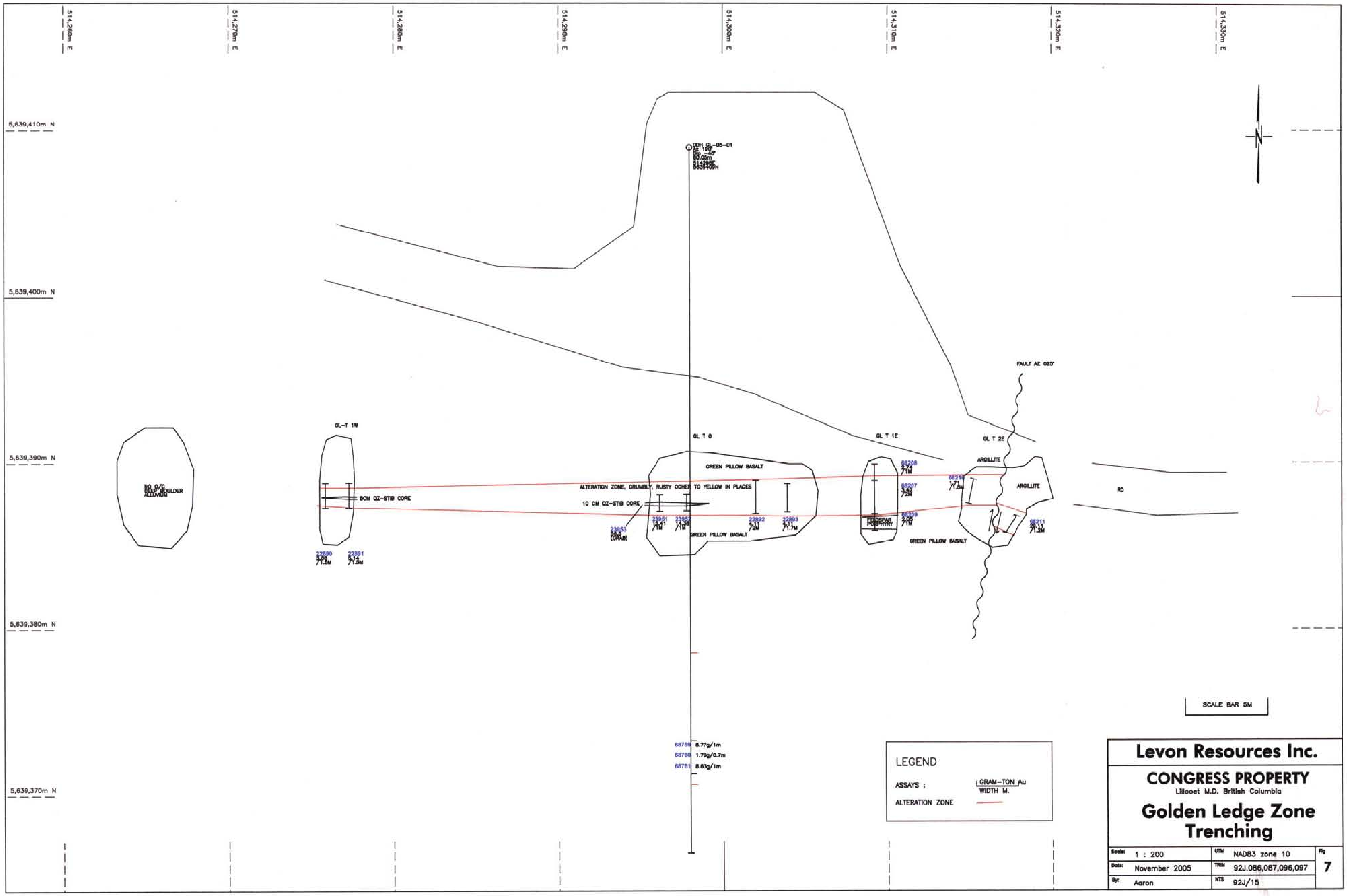
The next four holes of the summer 2005 drill program were drilled on the Lou Zone, one approximately 100 metres north of Tr 18/Tr-6-04, two under Tr-10 from opposite directions and one under Tr 27 (Maps 1 & 3). These holes added to the resource in the Lou Zone and better defined structural controls in the Tr 10 area.

The last hole of the summer 2005 program was drilled to intersect the zone just below the level of the Lower Howard workings and below the fall 2004 winter 2005 drill holes (Maps 1 & 2). The summer 2005 intersected the West Howard and Howard Zones and added to the mineral resources in the Howard Zone.

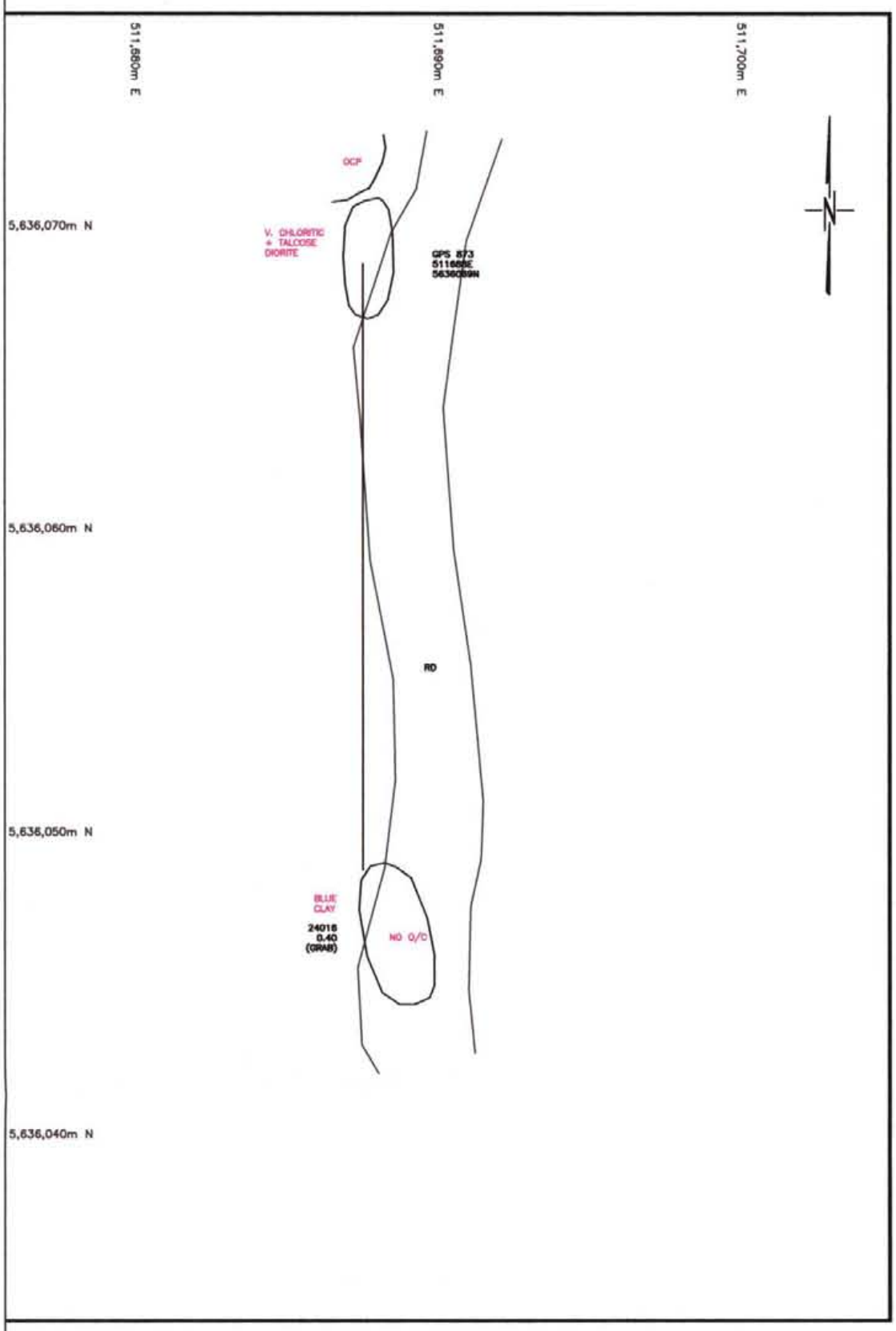
Sampling Method and Approach

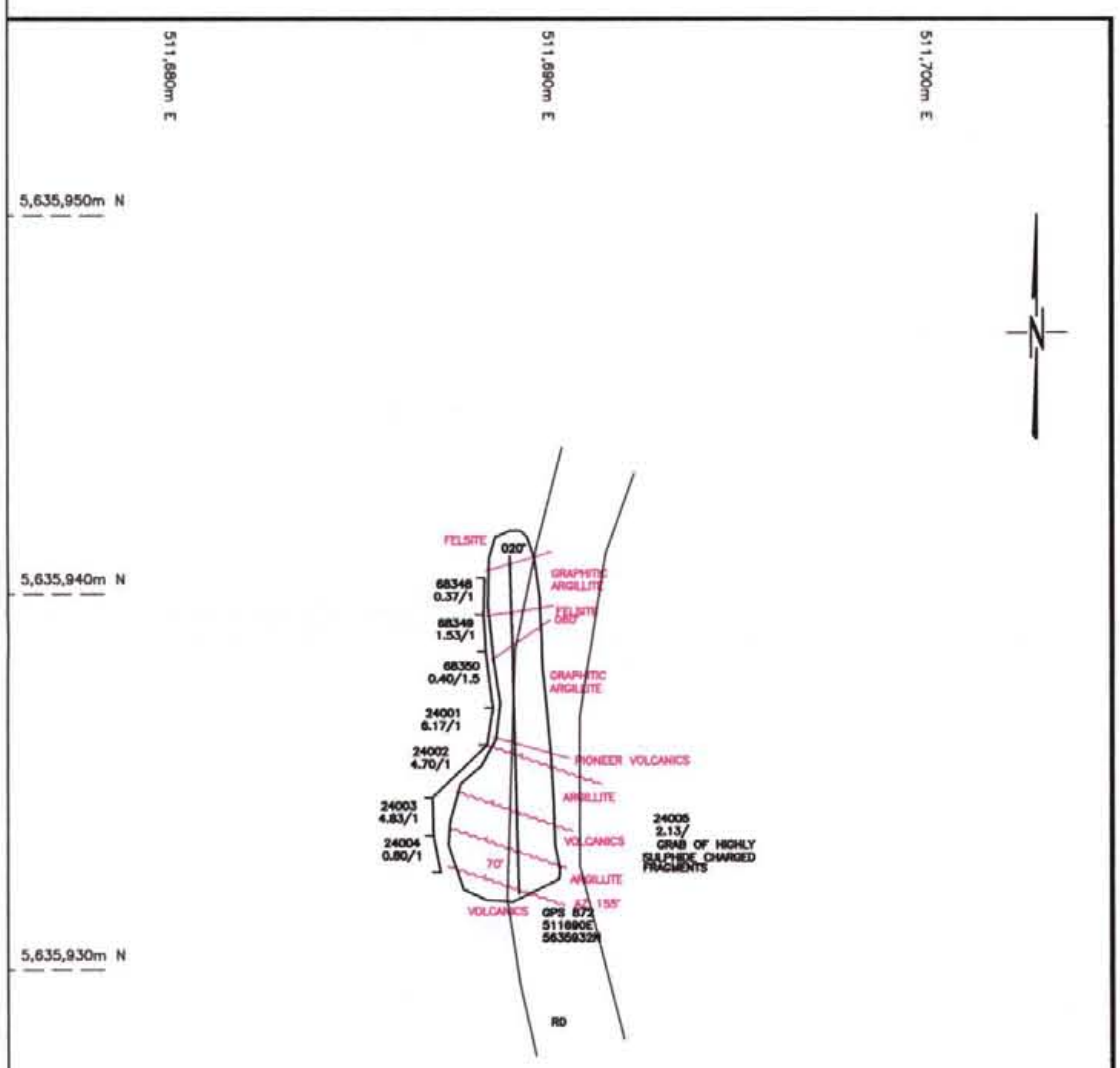
Trenches that reached bedrock were chip sampled along the long axis of the trench. Samples were from one to 2 metres in length, depending on lithology and mineralization. Samples vary from 2 to 3 kilograms per metre sampled. Where bedrock was not reached a soil sample was generally taken of the deepest material recovered. Generally, trenches were dug, sampled and reclaimed the same day.

MMI samples were taken every 10 metres from 10 to 20 centimetres depth from whatever soil was present at this depth along lines that were run at right angles to the projected extensions of the Howard and Lou Zones. Approximately 0.5 kilos of soil, with larger rocks and organics removed, was placed in a sealable plastic bag with the location marked on the outside with a felt pen.



514,260m E
514,270m E
514,280m E
514,290m E
514,300m E
514,310m E
514,320m E
514,330m E
5,639,410m N
5,639,400m N
5,639,390m N
5,639,380m N
5,639,370m N
DDH GL-05-01
NO O/C
DEEP BOULDER
ALLUVIUM
GL-T 1W
5CM QZ-STIB CORE
22890
22891
GL T 0
GREEN PILLOW BASALT
ALTERATION ZONE, CRUMBLY, RUSTY OCHER TO YELLOW IN PLACES
10 CM QZ-STIB CORE
23953
(GRAB)
23951
22892
22893
GREEN PILLOW BASALT
GL T 1E
68208
68207
68209
FELDSPAR PORPHYRY
GREEN PILLOW BASALT
GL T 2E
FAULT AZ 025°
ARGILLITE
ARGILLITE
68211
RD
68759 0.77g/1m
68760 1.70g/0.7m
68761 0.83g/1m
SCALE BAR 5M
LEGEND
ASSAYS :
GRAM-TON Au
WIDTH M.
ALTERATION ZONE
Levon Resources Inc.
CONGRESS PROPERTY
Lillooet M.D. British Columbia
Golden Ledge Zone
Trenching
Scale: 1 : 200
UTM NAD83 zone 10
Fig 7
Date: November 2005
TRIM 92J.086,087,096,097
By: Aaron
NTS 92J/15





LEGEND

ASSAYS : GRAM–TON Au / WIDTH M.

ALTERATION ZONE

SCALE BAR 5M

Levon Resources Inc.

CONGRESS PROPERTY

Lillooet M.D. British Columbia

North Wayside Zone

Scale:	1 : 200	UTM	NAD83 zone 10	Fig 8
Date:	April 2006	TRIM	92J.086,087,096,097	
By:	Aaron Pettipas	NTS	92J/15	

511,680m E | 511,690m E | 511,700m E | 511,710m E | 511,720m E

5,635,650m N

5,635,640m N

5,635,630m N

5,635,620m N

5,635,610m N

5,635,600m N

5,635,590m N

5,635,580m N

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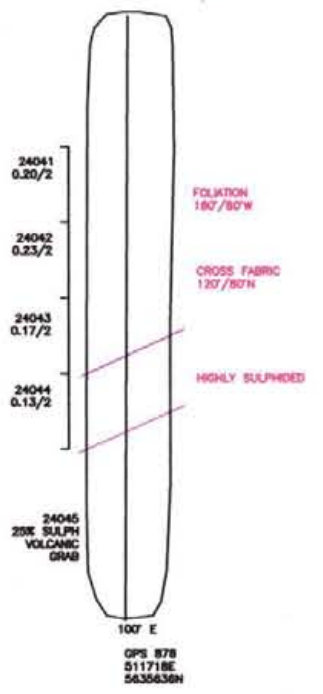
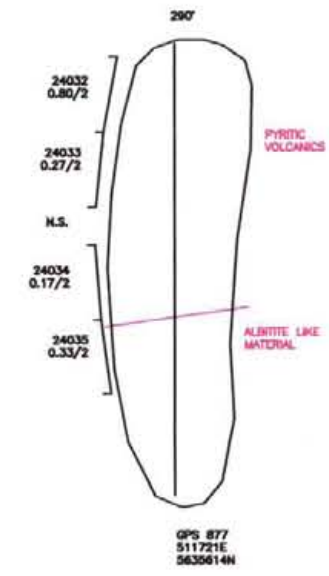
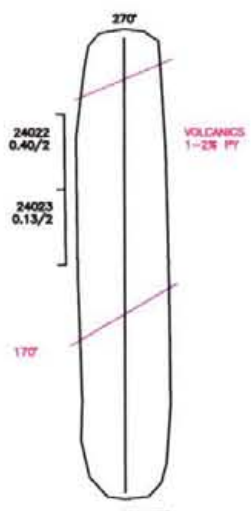
5,635,560m N

5,635,550m N

5,635,540m N







LEGEND

ASSAYS : GRAM–TON Au / WIDTH M.

ALTERATION ZONE

SCALE BAR 5M

Levon Resources Inc.

CONGRESS PROPERTY

Lillooet M.D. British Columbia

Wayside Zone

Scale:	1 : 200	UTM	NAD83 zone 10	Fig 9
Date:	April 2006	TRIM	92J.086,087,096,097	
By:	Aaron Pettipas	NTS	92J/15	

511,680m E | 511,690m E | 511,700m E | 511,710m E

Mineralized sections of drill core were sampled, generally to lithological or alteration boundaries. Two metre or less of core was split on site with a manual Longyear core splitter and half core splits placed in 6 mil plastic bags with a unique sample tag. The tag number was then written twice on the outside of the bag and the bags sealed with winter grade flagging. Core recovery was greater than 95% for all holes. The remaining core was stored in a secure core storage facility at the Bralorne Mines Ltd. office and mill site.

Sample Preparation, Analysis and Security

Sample preparation was as described above. No further sample preparation was carried out in the field. Samples were taken by the author, A. Pettipas, BSc., an employee of an associated company and by G. Polischuk, a contractor to the company. Two hundred forty-seven rock and core samples were in the possession of one of these individuals until they were delivered to Bralorne Mines Ltd. assay facility where the samples were fire assayed for gold using a one assay ton split. Pulps were transported by the author to Acme Analytical Laboratories Ltd. in Vancouver for 41 element, 4 acid digestion with analysis by ICP-MS (Appendix C). Thirty-four samples were taken directly to Acme Analytical Laboratories Ltd. for one assay ton fire assay for gold and 41 element analysis by ICP-MS. Bralorne Gold Mines Ltd. assay lab is not certified. Acme is an ISO 9001 accredited company.

Bralorne's assay laboratory carries out one check sample for every 10 samples run. Acme runs a check sample and a Standard for every 35 samples assayed. Correlation between the two laboratories was very good with less than 10% discrepancy for 90% of the samples.

The MMI samples were in the possession of the author or in a locked house before they were shipped via Canada Post to SGS Geochemical Laboratories, an ISO/IEC 17025 accredited laboratory, in Toronto, Ontario. No sample preparation was carried out in the field.

Data Verification

Quality control measures were carried out by the laboratories, as described above. The comparison of fire assays and ICP-MS gold results provided excellent verification of assay procedures.

None of the historic underground or drill data has been verified in this phase of work. Underground workings on the Howard, Lou and Congress Zones need rehabilitation to be safely accessed. Historic drill core was left on the property and has been vandalized to the point where it is not possible to obtain reliable check assays.

Adjacent Properties

The most important adjacent properties are the northwest trending low sulphidation, mesothermal quartz veins, variably mineralized with auriferous pyrite, arsenopyrite, galena, sphalerite, chalcopyrite and free gold of the Bralorne and Pioneer Mines, located 11 kilometers south of the property. These mines have the largest reported gold production in the Canadian Cordillera with 129.24 tonnes or 4,155,627 ounces of gold recovered from 7.2 million tones averaging 17.95 grams gold per tonne. Productive veins were typically 1.3 metres to 2.5 metres in width, with ore shoots locally up to 6.0 metres in width. The best ore shoot in the Pioneer Mine was mined over 800 metres vertically and 400 metres horizontally. These are "true fissure veins", following regional scale structures and exhibiting good vertical and horizontal continuity. The best ore shoots occur in the northwest trending ribbon veins where there is a flexure in the vein or where the veins are in contact with or cutting serpentinite.

This is the type of mineralization sought on the company's property.

Mineral Processing and Metallurgical Testing

Approximately 20 kilograms of vein material was collected in the Lower Howard adit from the East Howard zone in the first cross-cut to the west, approximately 60 metres in from the portal in 2004. This material was sent to Process Research Associates Ltd. for metallurgical testing. Samples from this material assayed 12.3 g/tonne or 0.36 ounces/ton gold. This grade is representative of the better mineralized areas in the Howard zones. The sample was taken from an area relatively close to the surface in a drift developed in 1959, so was partly oxidized. This probably accounts for the relatively high (for the Congress Property) gravity recovery of 23.1% of gold content. A relatively complex, 4 stage flotation process was able to recover a further 67.8% of gold for a total recovery of 90.9% of gold. Gold in the flotation concentrate was contained in 52.3% of the feed. The concentrate produced would grade roughly 0.5 ounces per ton or 15.5 g/tonne, not high enough to bear the cost of shipping offsite. Process Research Associates Ltd. recommended bio-leaching the concentrate on site and then using cyanide to recover the gold from the residue.

The milling characteristics of fresh Lou Zone material have not been adequately tested. Approximately 40 kilograms of the oxide material from the top of the Lou Zone was sent to PRA for beneficiation tests (Appendix D). Gravity and flotation recovery, similar to that used in the Bralorne mill did not produce adequate recovery due to the very fine habit of the gold. Cyanidation of the tailings showed that this material would be very amenable to standard cyanide heap leach or vat leach recovery.

Previous metallurgical work by PRA shows that the Congress material could produce a high value gold-antimony flotation concentrate (Personal communication, F. Wright, 2005).

A test is presently being carried out on fresh Howard Zone material by the Research and Productivity Council in New Brunswick to see if microwaving bulk mineralization could be a cost effective way of oxidizing the mineralization.

Mineral Resource and Reserve Estimate

Howard Zone

Christoffersen estimated that the following resources existed in the Howard Zone in 1988 (Christoffersen, 1988):

	Tonnes	Grade (grams gold/tonne)
Measured	40,192	8.7
Indicated	25,909	11.4
Inferred	218,540	9.45

These are in-situ, uncut resources, diluted to a minimum of 1.0 metre at zero grade. The resources conformed to a USGS classification and should be considered inferred, indicated and measured mineral resources under NI 43-101 compliant classifications.

Drilling in 2004 and 2005 has added a further 54,862.5 tonnes grading 3.3 grams/tonne gold to the inferred mineral resource. The 2004/2005 drilling raises the total inferred mineral resource to 273,402.5 tonnes grading 8.2 grams/tonne gold.

Lou Zone

Christoffersen estimated that there are the following resources in the Lou Zone:

	Tonnes	Grade (grams gold/tonne)
Inferred	105,673	9.4

(Underground resource: Uncut, diluted to 1.0 m with 0 grade.)

Inferred	124,300	2.4

(Open pit resource: Uncut, undiluted, 1:1 stripping ratio)

Two of the four holes drilled on the Lou Zone in 2005 increased the resources in the zone. Drill hole LZ-05-01 intersected 2.3 metres of 5.4 grams/tonne gold. Combined with the 3.3 metres of 19.0 grams/tonne gold in Trench-1-05 above hole LZ-05-01 an inferred mineral resource of 77,000 tonnes grading 13.6 grams/tonne gold is estimated for this area. Drill hole LZ-05-04 intersected 1.0 metres of 3.3 grams/tonne gold. This adds an inferred mineral resource of 6,875 tonnes grading 3.3 grams/tonne in this area.

The 2005 exploration program brings the present underground inferred mineral resource in the Lou Zone to 189,548 tonnes grading 10.9 grams/tonne gold.

There is an inferred open pittable mineral resource in the upper part of the Lou Zone, previously quantified by Cooke and Christoferssen as 124,300 tonnes grading 2.4 grams/tonne gold. Detailed sampling of the area exposed in the test pits shows that approximately 50% of the exposed 4 to 5 metre wide structure grades better than 10 grams/tonne gold. This material is broken and oxidized and these higher values are partly the result of surface enrichment. Oxidation extends 2 to 3 metres in depth. Overburden depths are sufficiently shallow so that an excavator can easily expose the Lou Zone for 700 metres from the area of Tr-56 to Tr-05-1. It is estimated that there is a minimum of 8,000 tonnes of oxidized enriched material in this zone.

Congress Zone

Mineral resources estimated for the Congress Zone are based on underground sampling, generally at 1.5 metre intervals along the exposed vein, carried out by several companies in the past including Bralorne-Pioneer Mines (1961), Sheep Creek Mining, Congress Gold Mines (1930's) and Congress Operating Company. Only composite grades, as shown on Map 4, are available. These resources were recalculated by Christoffersen in 1988 as 106,675 tonnes grading 7.4 grams gold per tonne (Christofferson, 1988). These are in-situ, uncut and undiluted resources which Christofferson categorized as "Indicated Ore Reserves" at the time using a USGS classification. These resources qualify as indicated mineral resources under NI 43-101 compliant classifications. Resampling of some of the drifts to verify historic sampling would raise these resources to the measured mineral resource category.

Other Relevant Data and Information

Detailed maps and cross sections were supplied by the company to verify the resources, particularly Cooke, 1985 and 1986 and Christoffersen, 1988. Reference should be made to these reports to verify resources.

Considerable information exists on the Wayside showing on the west side of the property. This showing was mined in the 1930's. Historical reports indicate one mineralized shoot was mined and further extensive drifting did not expose further ore grade mineralization.

Extensive surface work has been carried out intermittently on the property, particularly by Chevron in 1987. The scope of this report did not include a review of the Wayside information.

Interpretation and Conclusions

A substantially larger resource will need to be developed to justify the capital cost of any type of oxidation system. The best areas to quickly develop further mineral resources are in the Howard and Lou Zones because of their substantial widths. The medium term objective should be to develop a resource containing a minimum of 500,000 ounces of gold.

Deeper drilling of the Howard Zones in 2004 and 2005 has extended the mineralization to below the elevation of the projected extension of the Lower Howard drift and distinguished 4 mineralized zones en echelon. The most effective method of increasing the mineral resources in the Howard Zone is to advance the drift on the Lower Howard level. The drift should hit the down plunge extension of the mineralized shoot developed on the Upper Howard level in approximately 100 metres. A crosscut should be driven to the west for 30 metres at 50 metres from the present face to test the West Howard Zone. The results of this work should be evaluated and an updated mineral resource calculated. Further work should be planned based on this evaluation. Further MMI surveys and drilling should be carried out to trace the Howard Zone to the north and test the zone for more mineralization.

Trenching, drilling and MMI geochemical surveys have extended the Lou Zone 500 metres to the north and increased the resources in this zone. The most effective way to increase and upgrade the mineral resources in the Lou Zone is to drift on the zone. Access is possible at just above the level of Carpenter Lake through the Congress workings. This would involve 250 metres of crosscutting. Headings should then be drifted 250 metres to the north and 500 metres to the south. The southern drift should daylight between Highway 40 and Carpenter Lake. Further work should be then planned based on an evaluation of the mineral resources exposed.

Trenching in 2004 on the property succeeded in extending the strike extent of the Congress Zone 250 metres to the north. A one to 1.5 metre massive stibnite vein grading 6 to 8 grams/tonne gold was exposed. The 2004 trenching shows that more of this material could be developed but the low grade and relatively narrow widths make this material marginal at present metal prices. No further development should be carried out on the Congress zone at this time.

Initial work on the Golden Ledge showing was very encouraging. Further drilling should be carried out to expand and better define this showing.

Recommendations

Further development should be carried out on the Lou and Howard Zones to increase the known resources to a minimum of 500,000 contained ounces of gold. The Golden Ledge should be drill tested for continuity and grade. These objectives are best accomplished with a combined program of diamond drilling and underground development.

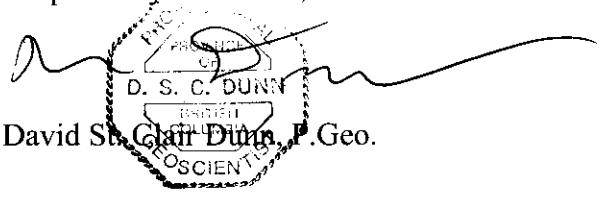
Specifically, 1,500 metres of drilling and 200 metres of underground development should be carried out on the Howard Zone. This should entail advancing the Lower Howard drift 100 metres, crosscutting 30 metres to West Howard and drifting 50 metres on the West Howard. Drilling should be mainly directed at the zone at 50 metre intervals in front of the drifts with a few short holes drilled to try to trace the zone to the north.

One thousand metres of drilling should be directed at testing the Lou Zone to the north beneath where it has been outlined by trenching and MMI sampling. One thousand metres of underground development should be carried out, 250 metres cross-cutting from the Congress Mine and 750 metres of drifting on the zone, 250 metres to the north and 500 metres to the south.

The Golden Ledge should be drill tested with at least 500 metres of drilling.

This program should take six months to complete at an estimated cost of $2,500,000.

Respectfully submitted,



David St. Clair Dunn, P.Geo.

References

Ash, C. 2001: Ophiolite *Related Gold Quartz Veins in the North American Cordillera*, British Columbia Geological Survey Branch, Bulletin 108.

Ash, C. and Alldrick, D., 1996: Au-Quartz Veins; *in* Selected British Columbia Mineral Deposit Profiles, Volume 2, D.V. Lefebure and T. Hoy, Editors, *British Columbia Ministry of Energy, Mines and Petroleum Resources,* pages 53-56.

Chapman, Wood and Griswold Ltd., 1965: *Memorandum Reports Nos.1, 2, and 3 Ace Mining Co.Ltd.*

Church, B.N., 1996: *Bridge River mining camp, geology and mineral deposits*; British Columbia Geological Survey Branch, Paper 1995-3, 159 p.

CIM, 1976: Special Volume 15; *Porphyry Deposits of the Canadian Cordillera*, A. Sutherland Brown, Editor.

Christofferson, J.E., 1987: *Status Report on the Congress Property.*

Christofferson, J.E., 1988: *Report on Gold Ore Reserves, Congress Property.*

CIM, 1995: Special Volume 46; *Porphyry Deposits of the Northwestern Cordillera of North America*, T. G. Schroeter, Editor.

Dolmage, Mason and Stewart Ltd, 1964: *Report on Examination of the Ace Mining Property.*

Dolmage, Mason and Stewart Ltd., 1965: *Ace Mining Company Limited N.P.L.,Final Report.*

Emmons, W. E., 1937: *Gold Deposits Of The World*, pages 99-100.

Geological Survey of Canada, 1995: *Geology of Canada*, no. 8; Geology of Canadian Mineral Deposit Types, pages 324, 328, 351-357, 360, 363. O.R. Eckstrand, W.D. Sinclair, and R.I. Thorpe, Editors.

Martin, W., 1961*: Ore Potentialities Bralorne Pioneer Mines Limited*, Unpublished Report.

Miller-Tait, J., and Sampson, C., 1995: *Exploration and Development Programmes, October 1993-July 1995, Bralorne-Pioneer Property*, Bralorne, British Columbia, Lillooet Mining Division, Bridge river Area, NTS 92-j/15, Latitude 50° 46'N, Longitude 122° 48' W.

Miller-Tait, J., Morris, A., and Hawthorn, G., 1996: *Bralorne Pioneer Gold Mines Ltd., Plan and Production Schedule for: 150 Ton per Day Mining Operation*, Bralorne Gold Mine, Bralorne, B.C. Unpublished Corporate Report.

Mindat Consultants, 1984; *Progress Report on the Congress Property.*

Process Research Associates Ltd., 2003. *Project 0304003, 0304006, 0403504.* Unpublished company reports.

Sawyer Consultants Inc., 1979; *Report on the Howard Property.*

Seraphim, R.H., 1981; *Progress Report Bridge River Claims of New Congress Resources Ltd.*

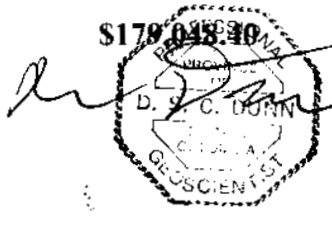
Schiarrizza, P., Gaba, R.G., Glover, J.K., Garver, J.I., and Umhoefer, P.J., 1996: *Geology and Mineral Occurences of the Taseko-Bridge River Area; Geological Survey of British Columbia,* Bulletin 100.

Appendix A

Statement of Costs

Appendix A
Statement of Costs

Personnel:	
D. Dunn 34.25 days @ $400/day	$13,700.00
C. Sampson 2.0 days @ $500/day	1,000.00
A. Pettipas 50 days @ $215/day	10,750.00
R. Reid 5.5 days @ $400/day	2,200.00
Transportation:	
Truck Rental-Oniva	4,125.00
Fuel	935.00
Lodging:	
House rental Goldbridge: V. Ross: 6 months @ $375/month:	1,650.00
D. Dunn: Room, board, fuel and truck rental:	3,807.49
Assays:	
SGS	211.88
Eco Tech	232.00
Bralorne Gold Mines	3,649.10
Acme Analytical	3,769.27
Drafting and Claim verification:	
Terracad	10,397.80
Terracognita	3,360.81
Accurate Mining Services	1,000.00
Excavator:	
Volvo Ex 210 BLC for 4 months @ 12,656.25/month:	50,625.00
Operator: Veritas (G. Polischuk) Wages, fuel, and p/u rental:	19,607.55
Drilling:	
ABC Drilling: 1,060.68 metres @ $38/m	40,348.50
Mob-Demob Drill and Excavator	
Buster's and Associates Hauling	4,079.00
Metallurgical Test Work	
F. Forgeron: 2 days @ $500/day	1,000.00
New Brunswick Research and Productivity Council	2,600.00
Project Total	**$179,048.40**



Appendix B

Diamond Drill Logs

Diamond Drill Record

Sheet # 1

Property Congress lou zone

Hole Number lz-05-01

Dip Test		
	Angle	
Depth	Reading	Corrected

UTM 514796E, 5638713N

AZ 045D

dip -45

EL.

Total Depth 178.91

Grid Location .....

Cross Section .....

Core Size ..... NQ

Date Begun 6/16/2005

Date Finished 6/22/2005

Date Logged 6/22/2005

Logged By ..... AP

Depth from	Depth to	Approx. width	Description	sample number	from	to	approx. width	rec.	Au g/t	Cu (%)	Ag (ppm)	Zn (%)
0	5.18		CASING									
5.18	17.1		CHERT/ARGILLITE (FERGUSSON)									
			v.broken with occasional volcanic									
			inclusions, 10% irregular qz-str's									
			< 1cm									
		R.Q.	70% <10cm									
17.1	64.7		PILLOW VOLCANICS, mostly									
			green									
			altered. Some inclusions of									
			chert/argillite									
			22.7-23.47: highly charged with py	68763	22.7	23.2			0.4			
		60deg. Fol	(10%), foliated almost schistise	68764	23.2	23.9			8.17			
			23.47-23.9:20%qz,10%py+as-py	68765	23.9	24.9			0.23			
			in talc alt volc.	68766	24.9	25.9			0.37			
		60deg fabri	dark bands with strieated slip									
			surfaces									
		70deg flt	23.6: icm gouge									
			23.9-24.9:less qz, 5-10%									
			py, some as-py +sph. +/- stib									

			24.9-28: less sulphide									
			28.0-29: 5cm clots of f.g. py	68767	28	29			0.33			
		10deg flt	29.0: fault+breccia(qz)									
		30deg	35.5: 50 cm inclusion of contorted									
			chert/argillite									
		40-55deg	37.2: 10cm gouge									
			and small flat slips									
		80deg	41.0-43.7: breccia, tuff like matrix									
			< 5% <1cm irregular veinlets									
			43.7-64.62: mixed alt volcanic+	68768	56	57			0.33			
			some tectonic breccia-qz infilling									
			v. broken ,rusty fractures, chloritic									
			patches									
		R.Q.	50%> 10cm									
64.7	87.73	35deg	MIXED: CHERT-ARGILLITE, PILL	68619	70	71			1.2			
			VOLCANICS	68620	71	72			0.8			
		60deg flt	72.24-72.60: 1-2%py	68769	72	73			7.67			
			brecciated with gouge	68770	73	74.3			3.63			
		50deg flt	73.7-74.04: v.f.g white sulphide	68621	74.3	75			0.73			
			(as-py?) ~1/2%, some qz	68622	75	76			1.03			
			75.95-76.35: cataclastic zone									
		5deg	**76.8 and 78.5: low angle faults**									
			minor py									
		30-60deg	84.0: py veinlets									
		R.Q.	~60% > 10cm									
				68771	87.48	88.04			0.1			
87.73	96.1		FELDSPAR PORPHYRY: 10%									
			~5mm spar pale green in a fine									
			grained tan matrix									
		R.Q.	100%									

96.1	135.2	50deg ctct	VOLCANICS: much py @ ctct	68772	96.1	96.9			0.27			
			somwat brecciated									
		60deg ctct	103.7-105.7: scanty mineralized	68773	103.7	104.7			0.03			
			zone	68774	104.7	105.7			0.1			
			111.83-118: transition to maroon									
			volcanics									
		80deg	121.2: 5cm zone of f.g. sulph.									
		R.Q.	80% > 10 cm									
135.2	178.91	30deg ctct	CHERT/ARGILLITE: tectonic									
			breccia, conglomerate like, with									
			minor inclusions of volcanic or									
			dyke rock:									
		30 deg	144.3: chalcedonic veinlet									
			barren									
			145.55-146.6:volcanic rock incl-	68775	145.55	146.6			1.77			
		70 deg	uding gougy slip									
		80deg	147.88-148.43: aph. Dyke,	68776	147.88	148.43			0.23			
			2% py, tan									
			150.2-151.2: 5% sulphide	68777	150.2	151.2			1.23			
		80deg	151.2-151.82: gray aph. Dyke	68778	151.2	151.82			0.83			
			151.82-152.4: chert/arg. W/much									
		,	py	68779	151.82	152.4			0.1			
		30deg ctct	154.35-155.35: dyke, green-yellow	68780	154.35	155.35			0.17			
			with <1%py									
			fractured contact									
			157.9-158.9: chert/arg w/much	68781	157.9	158.9			0.27			
			py, some qz+ aspy? (v.f.g.)									
		20deg	158.9: graphitic slip+gouge	68782	158.9	159.9			1			
			other flat fracture @ 160.62									
			163.5-164.15:dyke? Green									
			(maraposite) silicified resembles									

			listwanite										
			164.7-174.8: mixed with aph dyke	68784	164.7	165.7			0.1				
			chloritic, much py in places,	68785	165.7	166.7			0.07				
			brecciated with qz-carb infillings	68786	166.7	167.7			1.27				
			172.15: cinnabar patch in 1cm qz	68787	167.7	168.7			0.77				
				68788	168.7	169.7			0.23				
			173.65: 5cm 50% py segment	68789	169.7	170.7			0.1				
			then 10% py to 174.8, 5%qz-carb	68790	170.7	171.7			0.03				
			irregular infillings	68791	171.7	172.7			0.17				
				68792	172.7	173.7			0.57				
				68793	173.7	174.8			0.3				

R.Q. ~75% > 10 cm

178.91 E.O.H.

			minor py									
		R.Q.	50%>10cm									
73.6	80.26		VOLCANIC(PILLOWS), after									
			1.5m breccia zone, 0.5mm									
			grainsize, light green with 1%									
			qz str's and black bands @									
			random orientations									
80.26	**82.3**	75deg ctct	TUFF, water lain fine grained									
		40deg	layering									
		R.Q.	25%>10CM									
82.3	151.48		VOLCANICS ,maroon-green									
			with chert/arg inclusions									
			90.9-96.0: alteration zone, bleachi	24006	90.9	92			0.87			
		40deg fol	,dark bands <1cm (v.f.g arg.?)	24007	92	93			0.27			
			little sulphide	24008	93	94			0.3			
			94.65: 10cm qz-bx	24009	94	95			0.07			
		35deg flt	95.5: 1cm mud slip	24010	95	96.1			0.23			
		40deg flt	96.1: 1cm gouge									
			96.1-115.2:maroon basalt,									
			random qz-carb veinlets<1cm									
			115.2-119.4:alteration zone,	24011	115.2	116			0.07			
			qz-carb str's,hematite, buff in	24012	116	117			0.3			
			color, little sulphide	24013	117	118			0.1			
		30deg fol	118.5: 60cm black banding(arg?)	24014	118	119.2			0.2			
			119.4-127.4: basalt,maroon									
		40deg	127.4-128.1									
			alt zone,bleached,little sulphide									
			143.1-144.6: alt zone, bleaching									
		45deg fol	fabric ,fault zone like,py<1%									

Diamond Drill Record

Sheet # 1

Property Congress

Hole Number LZ-05-02

Dip Test		
	Angle	
Depth	Reading	Corrected

UTM: 514961E, 5638175N

AZ 272

DIP -45

EL

Total Depth 151.48

Grid Location .....

Cross Section 320S

Core Size ..... NQ

Date Begun 7/15/2005

Date Finished 7/18/2005

Date Logged 7/18/2005

Logged By ..... DD

Depth from	Depth to	Approx. width	Description	sample number	from	to	approx. width	rec.	Au g/t	Cu (%)	Ag (ppm)	Zn (%)
0	1.9		CASING									
1.9	39.6		FELDSPAR PORPHYRY: 30%									
			phenocrysts up to 1 cm									
		55deg	10-11.28: minor veining									
			26-30m: matrix dark gray									
			30-35: altered,bleached, fractured									
			open qz lined cavaties									
39.6	64.8	35deg ctct	CHERT/ARGILLITE(FERGUSSON)									
			mixed with tuff like material for									
			first 3m									
			41.2-41.75: v alt chalky dyke? +	24615	41.2	41.75			0.17			
			gouge									
			44.0: minor py									
64.8	67.7	bx ctct	VOLCANIC (PILLOWS),alt. Green									
			gray with 10%qz and dark bands									
			sulph <<1%, v.f.g.									
67.7	73.6	85deg ctct	CHERT/ARGILLITE(FERGUSSON									

			maroon pillows becoming gougy,										
			1m core loss in last run										
		R.Q.	~75%>10CM except indicated										
	151.48		E.O.H.										

Diamond Drill Record

Sheet # 1

Property Congress

Hole Number LZ-05-03

Dip Test		
	Angle	
Depth	Reading	Corrected

UTM: 514838E, 5638133E
AZ 084
DIP -70
EL 939M

Total Depth 142.34
Grid Location .....
Cross Section .....
Core Size ..... NQ

Date Begun 6/23/2005
Date Finished 7/13/2005
Date Logged 7/13/2005
Logged By ..... AP

Depth from	to	Approx. width	Description	sample number	from	to	approx. width	rec.	Au g/t	Cu (%)	Ag (ppm)	Zn (%)
0	2.13		CASING									
2.13	87.9		PILLOW BASALT,chloritic,maroon					80%				
			many random qz-carb veinlets,									
			vesicular in places									
			to 17m: v.broken,poor recovery									
		35deg ctct	56.4m: 20cm minor alt zone, <1%									
			py,bleaching (carb alt)									
			71.4: transition from mostly green									
			to mostly maroon and more solid									
			basalt.									
87.9	89.2	65deg	FELDSPAR PORPHYRY, beginnir	68346	87.84	88.84			0.13			
			with 20 cm of cemented gouge,	68347	88.84	89.84						
			then 5% 1-3mm fetdspar in an app	68338	89.84	90.53						
			green matrix,aphanitic becoming	68339	90.7	91.7						
			granular toward ending in 20 cm	68340	91.7	92.7						
			of alteration	68341	92.7	93.7						
				68342	93.7	94.7						
89.2	89.84	ctct broken	DYKE?, V ALTERED granular	68343	94.7	95.7						
			like alt 1mm grainsize dio.,	68344	95.7	96.13						

		70deg	89.5m: thin striaeted slips.	68345	96.13	96.9						
89.84	92.48	45deg flt	VOLCANICS,apple green, fine									
			grained,altered,brecciated with									
			~2%py									
		80deg flt	90.53m:15cm clay gouge									
92.48	93.9	60deg flt	MIXED, GOUGY ALT VOLCANIC									
			AND F.S.P. DYKE,broken, 1-2%									
			py, <1%fg gray(as-py?)									
93.9	95.24		ALT VOLCANIC,cataclastic									
		45deg fol	qz+10%py+5% v.f.g.white (as-py?									
95.24	95.55	70deg	FELDSPAR PORPHYRY, <									
			5% 1mm feldspar,									
			py clots and bands									
95.55	96.9		VOLCANICS,alt,some 20deg/ca									
			qz bands <1cm, locally up to 10%									
			py									
		60deg	**96.62: 1cm mud slip, dark gray**									
			clay like, including some arg. Like									
			material <5%									
		R.Q.zone	22>10cm, 8<10 cm									
96.9	142.34	65deg	VOLCANICS,basalt, maroon									
			solid with minor carb alt zones:									
		60deg	106.6-107.65m									
		60deg	111.8-114.95m									
		50deg fol	sulphide (py) <1%									
			121.9-122.4: gray F.S.P.									

		50deg	123-123.3:graphitic+carb alt									
	142.34		E.O.H.									

Diamond Drill Record

Sheet # 1

Property Congress

Hole Number C-05-06

Dip Test		
ACID	Angle	
Depth	Reading	Corrected
EOH	-83	

UTM 514090E, 5638019N
AZ 075
DIP -80
EL

Total Depth 273.4
Grid Location .....
Cross Section .....
Core Size ..... NQ

Date Begun 7/24/2005
Date Finished 8/6/2005
Date Logged 8/6/2005
Logged By ..... AP

Depth from	Depth to	Approx. width	Description	sample number	from	to	approx. width	rec.	Au g/t	Cu (%)	Ag (ppm)	Zn (%)
0	11.28		CASING									
11.28	13.3		VOLCANICS,pillows,olive green									
			granular									
13.3	50.4		DIORITE,green-gray, v.chloritic on									
			vitreous slips, 50%mafics 1-3mm									
			granoblastic sub vol intrusion.									
			14.2:10cm carb veinlet									
			38:2.5m ves.basalt inclusion									
			then finer grained dio.									
		R.Q.	too poor to estimate									
50.4	273.4		BASALT,maroon-green,pillows,									
			chloritic, <5% qz-carb veinlets									
			137.5:10cm gritty mod slip									
			138-140m:20% barren qz bx-infillir									
			179.9: 1m carb alt, breccia, minor	24051	179.9	190.9			0.17			
			qz-sulphide									
			181.3:70cm carb alt zone	24052	181.3	182			0.23			
			182.3: 1m v.bleached, up to 20%	24053	182	182.3			0.2			

		30deg band	py+ dark banding	24054	182.3	183.3			0.33			
			py in up to 2cm f.g. clots									
			188m: 1m minor alt zone									
		35deg	192.0m:20cm alt zone with 5cm									
			qz-sulphide									
203.4	211		HOWARD ZONE:	24055	203.4	204.3						
		45deg ctct	alt volcanics,qz,gougy slips+some	24056	204.3	205.3						
			brecciation	24057	205.3	206.3						
		35deg	203.7m:1cm slip	24058	206.3	207.3						
			203.8m: 10cm qz	24059	207.3	208.2						
		25deg	204.3m: 1cm slip(broken core)	24060	208.2	209.8						
			204.5-206.2m: much py+f.g d.gray	24061	209.8	211						
			(as-py?) +breccia									
			205.5m: 10cm qz, 10-20%py in									
			clots, ~1% sph.									
		45deg	206.2m: sharp slip									
			207.3-208.2m: maroon v. with									
			magnetite									
			208.2-209.8m:v.bleached,little									
			sulphide									
		70deg band	209.8-211m: friquent qz-sulphide									
			infillings in breccia									
		40deg	**211 abrupt end to zone**									
		R.Q.zone	**30>10cm, 10<10cm**									
			212m: much chlorite,serp like									
			229m: 2m much chlorite									
		70deg	254.9m:2.8m inclusion of graph.									
			chert/arg									
		60deg	264m:10cm graphitic banding,									
			minor qz									

			270-273.4m:core becomes v.									
			broken, much vitreous chl. Slips									
		R.Q.	~25%pcs>10cm									
	273.4		E.O.H.									

Appendix C

Assay Results and Assay Procedure

HEAD ASSAY REPORT

Client: Levon Resources Ltd. - Congress Lou Zone Project
Sample: as per ID

Date: 17-May-05
Project: 0503904

Elements	Units	Sample ID					Detection Limits		Analytical
		North	RE North	avg. North	South	1:1 Comp.	Minimum	Maximum	Method
Au	g/mt	14.40	14.80	14.60	12.10	13.35	0.01	5000	FA/AAS
Ag	g/mt	43.40	44.80	44.10	10.40	27.25	0.3	9999	FAGrav
S(tot)	%	--	0.53	0.53	0.90	0.72	0.01	100	Leco
As	%	0.40	0.40	0.40	0.60	0.50	0.001	100	Assay
Sb	%	1.14	1.22	1.18	1.34	1.26	0.001	100	AsyMuA
Fe	%	4.22	4.20	4.21	3.96	4.08	0.001	100	AsyMuA
Al	ppm	42837	44283	43560	70079	56820	100	50000	ICPM
Sb	ppm	10948	11302	11125	13130	12128	5	2000	ICPM
As	ppm	4172	4148	4160	4462	4311	5	10000	ICPM
Ba	ppm	270	274	272	354	313	2	10000	ICPM
Bi	ppm	<2	<2	<2	<2	<2	2	2000	ICPM
Cd	ppm	<0.2	<0.2	<0.2	<0.2	<0.2	0.2	2000	ICPM
Ca	ppm	4003	4159	4081	7249	5665	100	100000	ICPM
Cr	ppm	135	136	136	102	119	1	10000	ICPM
Co	ppm	23	23	23	20	22	1	10000	ICPM
Cu	ppm	195	201	198	77	138	1	20000	ICPM
Fe	ppm	41707	41829	41768	38867	40318	100	50000	ICPM
La	ppm	7	7	7	10	9	2	10000	ICPM
Pb	ppm	82	130	106	38	72	2	10000	ICPM
Mg	ppm	3690	3655	3673	4694	4183	100	100000	ICPM
Mn	ppm	1326	1337	1332	811	1071	1	10000	ICPM
Hg	ppm	<3	<3	<3	<3	<3	3	10000	ICPM
Mo	ppm	5	6	6	5	5	1	1000	ICPM
Ni	ppm	33	31	32	35	34	1	10000	ICPM
P	ppm	407	396	402	511	456	100	50000	ICPM
K	ppm	16226	16404	16315	23646	19981	100	100000	ICPM
Sc	ppm	10	11	11	10	10	1	10000	ICPM
Ag	ppm	43	44	44	12	28	0.5	1000	ICPM
Na	ppm	2943	2722	2833	5136	3984	100	100000	ICPM
Sr	ppm	81	85	83	204	144	1	10000	ICPM
Tl	ppm	<2	<2	<2	<2	<2	2	1000	ICPM
Ti	ppm	3920	3977	3949	3560	3754	100	100000	ICPM
W	ppm	14	15	15	18	16	5	1000	ICPM
V	ppm	86	86	86	101	94	1	10000	ICPM
Zn	ppm	429	393	411	227	319	1	10000	ICPM
Zr	ppm	46	58	52	46	49	1	10000	ICPM

HEAD ASSAY REPORT - WHOLE ROCK

Client: Levon Resources Ltd. - Congress Lou Zone Project
Sample: as per ID
Geochemical Fusion

Date: 17-May-05
Project: 0503904

Compounds	Units	Sample ID					Detection Limits		Analytical
		North	RE North	avg. North	South	1:1 Comp.	Min.	Max.	Method
Al2O3	%	8.50	8.02	8.26	11.29	9.78	0.01	100	WRock
BaO	%	0.03	0.03	0.03	0.05	0.04	0.01	100	WRock
CaO	%	0.79	0.77	0.78	1.43	1.11	0.01	100	WRock
Fe2O3	%	6.12	6.20	6.16	5.41	5.79	0.01	100	WRock
K2O	%	2.39	2.16	2.28	3.05	2.66	0.01	100	WRock
MgO	%	0.65	0.62	0.64	0.83	0.73	0.01	100	WRock
MnO	%	0.20	0.19	0.20	0.13	0.16	0.01	100	WRock
Na2O	%	0.41	0.48	0.45	0.74	0.59	0.01	100	WRock
P2O5	%	0.07	0.10	0.09	0.08	0.08	0.01	100	WRock
SiO2	%	73.32	73.58	73.45	71.02	72.24	0.01	100	WRock
TiO2	%	0.69	0.67	0.68	0.69	0.69	0.01	100	WRock
LOI	%	5.18	5.23	5.21	5.23	5.22	0.01	100	2000 F
Total	%	98.35	98.05	98.20	99.95	99.08	0.01	105	WRock

GRAVITY + FLOTATION TEST PROCEDURE

Client: Levon Resources Ltd. - Congress Lou Zone Project
Test: GF1
Sample: Lou Zone 1:1 Mix

Date: 3-May-05
Project: 0503904

Objective: To recover gold and gold bearing sulphide minerals by gravity and flotation at a grind size of 60% passing 74 microns. Cleaner flotation on combined rougher and scavenger concentrate

STAGE	TIME (min)	pH	ADDITION		COMMENTS
			Reagent	g/tonne	
Grind(2.0kg)	12	7.9			Target grind size P60=74um
Gravity					200G, 1.0psi, 28' bowl
Bulk Flotation *(on combined pan tails and gravity tails*					
Condition	2	7.9	PAX	50	
Rougher Float 1	10	7.9	DF250	43	to barren
Condition	2		PAX	25	
Rougher Float 2	7	7.9	DF250	6	to barren
Scavenger Flotation					
Condition	2	7.7	CuSO4	200	
			PAX	25	
Scavenger Float	8	7.8	DF250	17	
Cleaner Flotation *(on combined Rougher and Scavenger concentrate)*					Combine all concentrates
Cleaner Float	13	7.4	DF250		

GRAVITY + FLOTATION TEST METALLURGICAL BALANCE

Client: Levon Resources Ltd. - Congress Lou Zone Project
Test: GF1
Sample: Lou Zone 1:1 Mix

Date: 3-May-05
Project: 0503904

Objective: To recover gold and gold bearing sulphide minerals by gravity and flotation at a grind size of 60% passing 74 microns. Cleaner flotation on combined rougher and scavenger concentrate

Flotation Balance

Product	Weight		Assay					Distribution				
			Au	Fe	As	Sb	$S_{(T)}$	Au	Fe	As	Sb	$S_{(T)}$
	(g)	(%)	(g/t)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)
Cleaner Concentrate	99.5	5.1	52.60	9.40	2.11	12.63	8.44	22.8	11.8	23.4	50.0	58.3
Cleaner Tails	196.3	10.1	23.70	7.89	0.59	1.17	0.97	20.2	19.5	12.9	9.1	13.2
Ro & Sc. Concentrate	**295.9**	**15.3**	**33.42**	**8.40**	**1.10**	**5.03**	**3.48**	**43.0**	**31.3**	**36.3**	**59.2**	**71.5**
Final Tail	1638.5	84.7	8.00	3.33	0.35	0.63	0.25	57.0	68.7	63.7	40.8	28.5
Calculated Head	**1934.4**	**100.0**	**11.89**	**4.10**	**0.46**	**1.30**	**0.74**	**100.0**	**100.0**	**100.0**	**100.0**	**100.0**
Expected Head			13.20	4.08	0.50	1.26	0.72					

Gravity + Flotation Balance

Product	Weight		Assay					Distribution				
			Au	Fe	As	Sb	$S_{(T)}$	Au	Fe	As	Sb	$S_{(T)}$
	(g)	(%)	(g/t)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)
Gravity Concentration												
Gravity Pan Concentrate	1.9	0.1	150.74					1.2				
Flotation Concentration												
Cleaner Concentrate	99.5	5.1	52.60	9.40	2.11	12.63	8.44	22.5	11.8	23.4	50.0	58.3
Cleaner Tails	196.3	10.1	23.70	7.89	0.59	1.17	0.97	20.0	19.5	12.9	9.1	13.2
Ro & Sc. Concentrate	**295.9**	**15.3**	**33.42**	**8.40**	**1.10**	**5.03**	**3.48**	**42.5**	**31.3**	**36.3**	**59.2**	**71.5**
Gravity Pan Conc + Flotation Concentrate	**297.8**	**15.4**	**34.18**	**8.34**	**1.10**	**4.99**	**3.46**	**43.7**	**31.3**	**36.3**	**59.2**	**71.5**
Final Tail	1638.5	84.6	8.00	3.33	0.35	0.63	0.25	56.3	68.7	63.7	40.8	28.5
Calculated Head	**1936.3**	**100.0**	**12.03**	**4.10**	**0.46**	**1.30**	**0.74**	**100.0**	**100.0**	**100.0**	**100.0**	**100.0**
Expected Head			13.35	4.08	0.50	1.26	0.72					

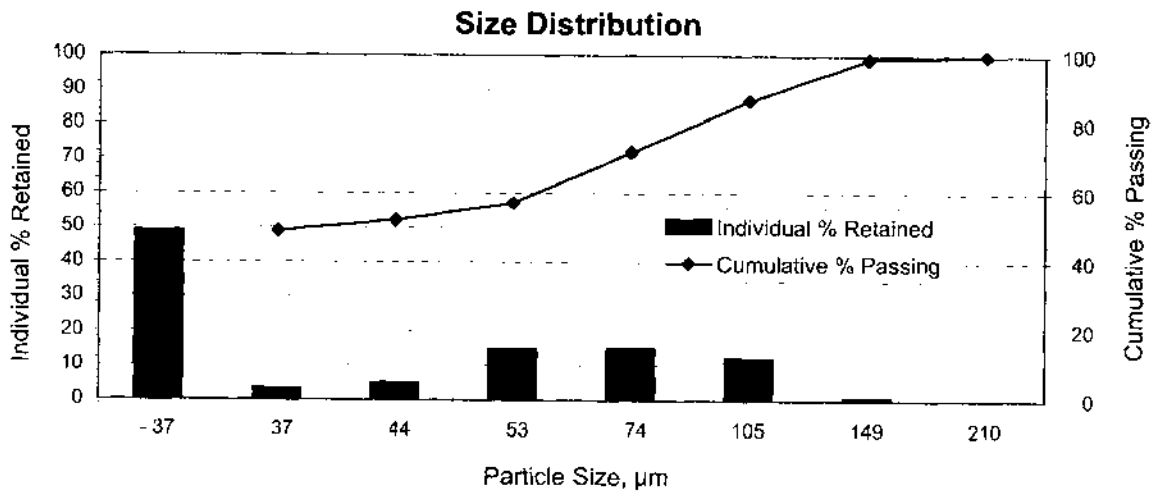
SIZE ANALYSIS REPORT

Client: Levon Resources Ltd. - Congress Lou Zone Project
Test: GF1
Sample: Flotation Head (Gravity Tails)
Grind: Gravity Feed, 2 kg for 12 minutes at 65% solids in stainless steel mill #2.

Date: 3-May-05
Project: 0503904

Sieve Size		Individual	Cumulative
Tyler Mesh	**Micrometers**	**% Retained**	**% Passing**
65	210	0.0	100.0
100	149	0.9	99.1
150	105	12.1	87.0
200	74	15.1	71.9
270	53	14.8	57.2
325	44	5.0	52.2
400	37	3.3	48.9
Undersize	- 37	48.9	-
TOTAL:		100.0	

60 % Passing Size (µm) = 56



CYANIDATION TEST REPORT

Client: Levon Resources Ltd. - Congress Lou Zone Project
Test: CGF1
Sample: Lou Zone - GF1 Tailings

Date: 1-Jun-05
Project: 0503904

Objective: To determine Au & Ag extraction by cyanidation of flotation tailings

TEST CONDITIONS

Solids:	1,473 g
Solution:	2,210 g
Solids:	40.0 %
80% Passing Size:	n.a. µm
Initial NaCN:	1.0 g/L
Target pH:	10.5
Test Duration:	48 hours

TEST DESCRIPTION

- repulped to 40% solids
- adjusted to and maintained pH 10.5
- adjusted to and maintained at 1.0 g/L NaCN
- sampled at 6 and 24 hours
- test ended after 48 hours
- filtered and displacement washed with hot cyanide solution followed by two hot water displacement washes
- solution and solids fire assayed for Au and Ag content

HEAD GRADE

	Au	Ag
Calculated Total:	7.50 g/t	9.5 g/t
Measured Total:	8.00 g/t	n.a. g/t

LEACH TEST DATA

Time	NaCN		Lime	pH		dO_2	Slurry	Solution				
							Weight	Vol.	Assay Vol.	Au		Ag	
(hours)	(g/L)	(g)	(g)	before	after	(mg/L)	(g)	(mL)	(mL)	(mg/L)	(mg)	(mg/L)	(mg)
0	1.00	2.21	2.00	7.7	10.7		2,500	1,069					
1	1.00		0.80	10.0	10.8	8.9			5				
3	1.00			10.5					5				
6	0.90	0.22	1.08	10.1	10.8		3,727	2,295	55	4.0	9.4	5.0	11.7
24	0.70	0.66	1.08	10.1	.10.8		3,762	2,330	55	4.1	9.8	5.3	12.7
30	0.90	0.22	1.07	10.4	10.7				5				
48	0.70			10.3			3,767	2,335		4.1	10.1	5.1	12.6
Total		3.31	6.03										

SOLIDS

Time	Residue				
	Weight	Au		Ag*	
(hours)	(g)	(g/t)	(mg)	(g/t)	(mg)
48	1,432	0.68	0.97	1.0	1.4

Time	Assay (ppm)	
(hours)	As	Sb
6	2.5	6.1
24	4.9	12.7
48	9.1	24.2
Residue	3470	6520

CYANIDATION RESULTS

Time	Distribution		Reagent Consumption		Reducing Power
	Au	Ag	NaCN	$Ca(OH)_2$	0.1 N $KMnO_4$/L
(hours)	(%)	(%)	(kg/t)	(kg/t)	(mL)
6	84.7	83.7	0.85		
24	88.8	90.6	0.54		
48	91.2	89.8	1.14	4.09	50
Residue	8.8	10.2			
Total	100.0	100.0			

HEAD ASSAY REPORT

Client: Levon Resources Ltd. - Congress Golden Ledge Project
Sample: as per ID

Date: 23-May-05
Project: 0504305

Elements	Units	Sample ID					Detection Limits		Analytical
		Sample A	RE A	avg. A	Sample B	1:1 Comp.	Minimum	Maximum	Method
Au	g/mt	28.60	28.60	28.60	5.56	17.08	0.01	5000	FA/AAS
Ag	g/mt	6.00	6.40	6.20	3.00	4.60	0.3	9999	FAGrav
S(tot)	%	1.27	1.24	1.26	0.58	0.92	0.01	100	Leco
As	%	--	2.13	2.13	--	2.13	0.001	100	Assay
Sb	%	--	0.32	0.32	--	0.32	0.001	100	AsyMuA
Fe	%	6.53	6.74	6.64	6.22	6.43	0.001	100	AsyMuA
Al	ppm	--	57782	57782	--	57782	100	50000	ICPM
Sb	ppm	--	3157	3157	--	3157	5	2000	ICPM
As	ppm	--	16619	16619	--	16619	5	10000	ICPM
Ba	ppm	--	309	309	--	309	2	10000	ICPM
Bi	ppm	--	<2	<2	--	<2	2	2000	ICPM
Cd	ppm	--	<0.2	<0.2	--	<0.2	0.2	2000	ICPM
Ca	ppm	--	6029	6029	--	6029	100	100000	ICPM
Cr	ppm	--	164	164	--	164	1	10000	ICPM
Co	ppm	--	44	44	--	44	1	10000	ICPM
Cu	ppm	--	52	52	--	52	1	20000	ICPM
Fe	ppm	--	64828	64828	--	64828	100	50000	ICPM
La	ppm	--	10	10	--	10	2	10000	ICPM
Pb	ppm	--	<2	<2	--	<2	2	10000	ICPM
Mg	ppm	--	4683	4683	--	4683	100	100000	ICPM
Mn	ppm	--	577	577	--	577	1	10000	ICPM
Hg	ppm	--	<3	<3	--	<3	3	10000	ICPM
Mo	ppm	--	5	5	--	5	1	1000	ICPM
Ni	ppm	--	29	29	--	29	1	10000	ICPM
P	ppm	--	352	352	--	352	100	50000	ICPM
K	ppm	--	23102	23102	--	23102	100	100000	ICPM
Sc	ppm	--	20	20	--	20	1	10000	ICPM
Ag	ppm	--	5	5	--	5	0.5	1000	ICPM
Na	ppm	--	2429	2429	--	2429	100	100000	ICPM
Sr	ppm	--	158	158	--	158	1	10000	ICPM
Tl	ppm	--	<2	<2	--	<2	2	1000	ICPM
Ti	ppm	--	8661	8661	--	8661	100	100000	ICPM
W	ppm	--	29	29	--	29	5	1000	ICPM
V	ppm	--	178	178	--	178	1	10000	ICPM
Zn	ppm	--	65	65	--	65	1	10000	ICPM
Zr	ppm	--	106	106	--	106	1	10000	ICPM

HEAD ASSAY REPORT - WHOLE ROCK

Client: Levon Resources Ltd. - Congress Golden Ledge Project
Sample: as per ID
Geochemical Fusion

Date: 23-May-05
Project: 0504305

Compounds	Units	Sample ID					Detection Limits		Analytical
		Sample A	RE A	avg. A	Sample B	1:1 Comp.	Min.	Max.	Method
Al2O3	%	9.45	9.69	9.57	9.78	9.68	0.01	100	WRock
BaO	%	0.05	0.05	0.05	0.04	0.05	0.01	100	WRock
CaO	%	1.18	1.19	1.19	1.65	1.42	0.01	100	WRock
Fe2O3	%	9.53	9.78	9.66	9.04	9.35	0.01	100	WRock
K2O	%	2.93	3.06	3.00	3.15	3.07	0.01	100	WRock
MgO	%	0.82	0.83	0.83	1.18	1.00	0.01	100	WRock
MnO	%	0.10	0.10	0.10	0.17	0.14	0.01	100	WRock
Na2O	%	0.34	0.39	0.37	0.44	0.40	0.01	100	WRock
P2O5	%	0.09	0.13	0.11	0.16	0.14	0.01	100	WRock
SiO2	%	67.89	67.94	67.92	66.67	67.29	0.01	100	WRock
TiO2	%	1.50	1.53	1.52	1.52	1.52	0.01	100	WRock
LOI	%	4.94	5.02	4.98	5.84	5.41	0.01	100	2000 F
Total	%	98.82	99.71	99.27	99.64	99.45	0.01	105	WRock

GRAVITY + FLOTATION TEST PROCEDURE

Client: Levon Resources Ltd. - Congress Golden Ledge Project
Test: GF1
Sample: Golden Ledge 1:1 Mix

Date: 23-May-05
Project: 0504305

Objective: To recover gold and gold bearing sulphide minerals by gravity and flotation at a grind size of 60% passing 74 microns. Cleaner flotation on combined rougher and scavenger concentrate

STAGE	TIME (min)	pH	ADDITION		COMMENTS
			Reagent	g/tonne	
1 kg each, Head A + B					
Grind(2.0kg)	13	7.9			Target grind size P60=74um
Gravity					200G, 1.0psi, 28' bowl
Bulk Flotation					
(on combined pan tails and gravity tails					
Condition	2	7.6	PAX	50	
Rougher Float 1	7	7.8	DF250	41	to barren
Condition	2		PAX	25	
Rougher Float 2	3	7.8	DF250	4	to barren
Scavenger Flotation					
Condition	2	7.7	CuSO4	200	
			PAX	25	
Scavenger Float	4	7.6	DF250	6	
Cleaner Flotation					Combine all concentrates
(on combined Rougher and Scavenger concentrate)					
Cleaner Float	11	7.8	DF250		

GRAVITY + FLOTATION TEST METALLURGICAL BALANCE

Client: Levon Resources Ltd. - Congress Golden Ledge Project
Test: GF1
Sample: Golden Ledge 1:1 Mix

Date: 23-May-05
Project: 0504305

Objective: To recover gold and gold bearing sulphide minerals by gravity and flotation at a grind size of 60% passing 74 microns. Cleaner flotation on combined rougher and scavenger concentrate

Flotation Balance

Product	Weight		Assay					Distribution				
			Au	Fe	As	Sb	$S_{(T)}$	Au	Fe	As	Sb	$S_{(T)}$
	(g)	(%)	(g/t)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)
Cleaner Concentrate	182.4	9.4	74.00	12.73	5.36	0.69	5.16	44.5	20.5	40.7	45.6	64.4
Cleaner Tails	255.6	13.2	21.20	8.12	1.81	0.06	0.98	17.9	18.3	19.3	5.9	17.1
Ro & Sc. Concentrate	**438.0**	**22.6**	**43.18**	**10.04**	**3.29**	**0.32**	**2.72**	**62.4**	**38.8**	**60.0**	**51.6**	**81.5**
Final Tail	1499.3	77.4	7.60	4.62	0.64	0.09	0.18	37.6	61.2	40.0	48.4	18.5
Calculated Head	**1937.3**	**100.0**	**15.64**	**5.85**	**1.24**	**0.14**	**0.75**	**100.0**	**100.0**	**100.0**	**100.0**	**100.0**
Expected Head			16.57	6.43	2.13	0.32	0.92					

Gravity + Flotation Balance

Product	Weight		Assay					Distribution				
			Au	Fe	As	Sb	$S_{(T)}$	Au	Fe	As	Sb	$S_{(T)}$
	(g)	(%)	(g/t)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)	(%)
Gravity Concentration												
Gravity Pan Concentrate	2.6	0.1	334.86					2.8				
Flotation Concentration												
Cleaner Concentrate	182.4	9.4	74.00	12.73	5.36	0.69	5.16	43.3	20.5	40.7	45.6	64.4
Cleaner Tails	255.6	13.2	21.20	8.12	1.81	0.06	0.98	17.4	18.3	19.3	5.9	17.1
Ro & Sc. Concentrate	**438.0**	**22.6**	**43.18**	**10.04**	**3.29**	**0.32**	**2.72**	**60.6**	**38.8**	**60.0**	**51.6**	**81.5**
Gravity Pan Conc + Flotation Concentrate	**440.6**	**22.7**	**44.92**	**9.98**	**3.27**	**0.32**	**2.70**	**63.5**	**38.8**	**60.0**	**51.6**	**81.5**
Final Tail	1499.3	77.3	7.60	4.62	0.64	0.09	0.18	36.5	61.2	40.0	48.4	18.5
Calculated Head	**1940.0**	**100.0**	**16.08**	**5.84**	**1.24**	**0.14**	**0.75**	**100.0**	**100.0**	**100.0**	**100.0**	**100.0**
Expected Head			17.08	6.43	2.13	0.32	0.92					

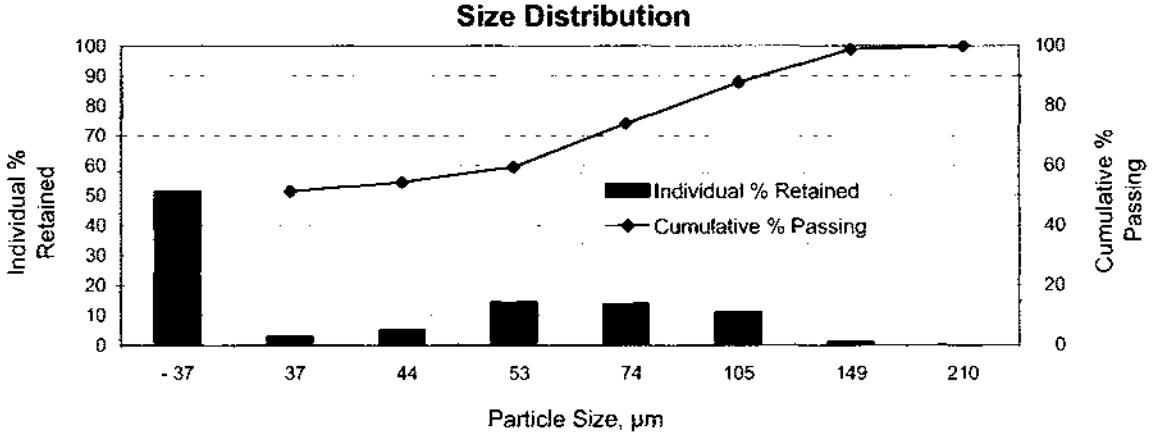
SIZE ANALYSIS REPORT

Client: Levon Resources Ltd. - Congress Golden Ledge Project
Test: GF1
Sample: Flotation Head (Gravity Tails)
Grind: Gravity feed, 2 kg for 13 minutes at 65% solids in stainless steel mill #2.

Date: 23-May-05
Project: 0504305

Sieve Size		Individual	Cumulative
Tyler Mesh	Micrometers	% Retained	% Passing
65	210	0.1	99.9
100	149	1.1	98.9
150	105	11.1	87.8
200	74	13.8	74.0
270	53	14.4	59.6
325	44	5.1	54.5
400	37	3.1	51.4
Undersize	- 37	51.4	-
TOTAL:		100.0	

60 % Passing Size (µm) = 53



BRALORNE GOLD MINE Ltd.

Assay Report

Sample wt.	Au wt.	Sample #	Au oz/t
29.166	0.005	22801	0.01
29.166	0.049	22802	0.05
29.166	0.024	22803	0.02
29.166	0.190	22804	0.19
29.166	0.009	22805	0.01
29.166	0.007	22806	0.01
29.166	0.088	22807	0.09
29.166	0.352	22808	0.35
29.166	0.017	22809	0.02
29.166	0.005	22810	0.01
29.166	0.277	22811	0.28
29.166	0.400	22812	0.40
29.166	0.266	22813	0.27
29.166	0.098	22814	0.10
29.166	0.002	22815	0.00
29.166	0.03	22816	0.03
29.166	0.010	22817	0.01
29.166	0.011	22818	0.01
29.166	0.049	22819	0.05
29.166	0.033	22820	0.03
29.166	0.062	22821	0.06
29.166	0.066	22822	0.07
29.166	0.025	22823	0.03
29.166	0.041	22824	0.04
29.166	0.145	22825	0.15
29.166	0.034	22826	0.03
29.166	0.048	22827	0.05
29.166	0.083	22828	0.08
29.166	0.523	22829	0.52
29.166	0.353	22830	0.35
29.166	0.034	22831	0.03
29.166	0.021	22832	0.02
29.166	0.068	22833	0.07
29.166	0.279	22834	0.28
29.166	0.306	22835	0.31
29.166	0.021	22836	0.02
29.166	0.054	22837	0.05
29.166	0.163	22838	0.16
29.166	0.200	22839	0.20
29.166	0.004	22840	tr
29.166	0.039	22841	0.04
29.166	0.325	22842	0.33
29.166	0.354	22843	0.35
29.166	0.059	22844	0.06
29.166	0.197	22845	0.20
29.166	0.551	22846	0.55
29.166	0.378	22847	0.38
29.166	0.031	22848	0.03
29.166	0.002	22849	tr
29.166	0.047	22850	0.05
29.166	0.053	22851	0.05
29.166	0.004	22852	tr
29.166	0.023	22853	0.02
29.166	0.044	22854	0.04
29.166	0.025	22855	0.03
29.166	0.042	22856	0.04
29.166	0.025	22857	0.03
29.166	0.001	22858	tr
29.166	0.019	22859	0.02
29.166	0.090	22860	0.09
29.166	0.040	22861	0.04
29.166	0.057	22862	0.06
29.166	0.029	22863	0.03
29.166	0.011	22864	0.01
29.166	0.060	22865	0.06
29.166	0.044	22866	0.04
29.166	0.007	22867	0.01
29.166	0.015	22868	0.02
29.166	0.001	22869	tr
29.166	0.001	22870	tr
29.166	0.001	22871	tr
29.166	0.006	22872	0.01
29.166	0.002	22873	tr
29.166	0.003	22874	tr
29.166	0.004	22875	tr
29.166	0.009	22876	0.01
29.166	0.019	22877	0.02
29.166	0.25	22878	0.25
29.166	0.189	22879	0.19
29.166	0.167	22880	0.17
29.166	0.011	22881	0.01
29.166	0.035	22884	0.04
29.166	0.041	22885	0.04
29.166	0.019	22886	0.02
29.166	0.089	22887	0.09
29.166	0.049	22888	0.05
29.166	0.032	22889	0.03
29.166	0.086	22890	0.09
29.166	0.149	22891	0.15
29.166	0.122	22892	0.12
29.166	0.120	22893	0.12
29.166	0.004	22894	tr
29.166	0.006	22895	0.01
29.166	0.009	22896	0.01
29.166	0.007	22897	0.01
29.166	0.025	22898	0.03
29.166	0.484	22899	0.48
29.166	0.001	22900	tr
29.166	0.003	22901	tr
29.166	0.321	22902	0.32
29.166	0.003	22903	tr
29.166	0.01	22904	0.01
29.166	0.012	22905	0.01
29.166	0.015	22906	0.02
29.166	0.010	22907	0.01
29.166	0.073	22908	0.07
30	0.021	23520	0.70
30	0.001	23584	0.03
30	0.004	23585	0.13
30	0.006	23586	0.20
30	0.010	23587	0.33
30	0.007	23588	0.23
30	0.002	23589	0.07
30	0.010	23590	0.33
30	0.007	23591	0.23
30	0.009	23592	0.30
30	0.006	23593	0.20
30	0.002	23594	0.07
30	0.003	23595	0.10
30	0.002	23596	0.07
30	0.011	23597	0.37
30	0.005	23598	0.17

30	0.002	23599	0.07
30	0.006	23600	0.20
29.166	0.452	23951	0.45
29.166	0.419	23952	0.42
29.166	1.648	23953	1.65
29.166	0.211	23954	0.21
29.166	0.220	23955	0.22
29.166	0.101	23956	0.10
29.166	0.264	23957	0.26
29.166	0.104	23958	0.10
29.166	0.378	23959	0.38
29.166	0.005	23960	0.01
29.166	0.367	23961	0.37
29.166	0.052	23962	0.05
29.166	0.156	23963	0.16
29.166	0.295	23964	0.30
29.166	0.165	23965	0.17
29.166	0.066	23966	0.07
29.166	1.33	23967	1.33
29.166	0.077	23968	0.08
29.166	0.047	23969	0.05
29.166	0.555	23970	0.56
29.166	0.106	23971	0.11
29.166	0.606	23972	0.61
29.166	0.283	23973	0.28
29.166	0.182	23974	0.18
29.166	0.117	23975	0.12
29.166	0.005	23976	0.01
29.166	0.070	23977	0.07
29.166	0.182	23978	0.18
29.166	0.216	23979	0.22
29.166	0.142	23980	0.14
29.166	0.072	23981	0.07
29.166	0.112	23982	0.11
29.166	0.111	23983	0.11
29.166	0.085	23984	0.09
29.166	0.121	23985	0.12
29.166	0.050	23986	0.05
29.166	0.170	23987	0.17
29.166	0.124	23988	0.12
29.166	0.618	23989	0.62
29.166	0.018	23990	0.02
29.166	0.254	23991	0.25
29.166	0.305	23992	0.31
29.166	0.192	23993	0.19
29.166	0.053	23994	0.05
29.166	0.167	23995	0.17
29.166	0.688	23996	0.69
29.166	0.423	23997	0.42
29.166	0.121	23998	0.12
29.166	0.104	23999	0.10
29.166	0.023	24000	0.02
30	0.185	24001	6.17
30	0.141	24002	4.70
30	0.145	24003	4.83
30	0.024	24004	0.80
30	0.064	24005	2.13
30	0.026	24006	0.87
30	0.008	24007	0.27
30	0.009	24008	0.30
30	0.002	24009	0.07
30	0.007	24010	0.23
30	0.002	24011	0.07
30	0.009	24012	0.30

30	0.003	24013	0.10
30	0.002	24014	0.07
30	0.006	24015	0.20
30	0.012	24016	0.40
30	0.008	24017	0.27
30	0.025	24018	0.83
30	0.002	24019	0.07
30	0.007	24020	0.23
30	0.005	24021	0.17
30	0.012	24022	0.40
30	0.004	24023	0.13
30	0.032	24024	1.07
30	0.008	24025	0.27
30	0.099	24026	3.30
30	0.006	24027	0.20
30	0.003	24028	0.10
30	0.008	24029	0.27
30	0.003	24030	0.10
30	0.002	24031	0.07
30	0.024	24032	0.80
30	0.008	24033	0.27
30	0.005	24034	0.17
30	0.010	24035	0.33
30	0.010	24036	0.33
30	0.004	24037	0.13
30	0.007	24038	0.23
30	0.006	24039	0.20
30	0.004	24040	0.13
30	0.006	24041	0.20
30	0.007	24042	0.23
30	0.005	24043	0.17
30	0.004	24044	0.13
30	0.013	24045	0.43
30	0.001	24046	0.03
30	0.004	24047	0.13
30	0.002	24049	0.07
30	0.046	24050	1.53
30	0.005	24051	0.17
30	0.007	24052	0.23
30	0.006	24053	0.20
30	0.010	24054	0.33
30	0.048	24055	1.60
30	0.006	24056	0.20
30	0.192	24057	6.40
30	0.004	24058	0.13
30	0.003	24059	0.10
30	0.002	24060	0.07
30	0.034	24061	1.13
29.166	0.014	68201	0.01
29.166	0.031	68202	0.03
29.166	1.019	68203	1.02
29.166	0.282	68204	0.28
29.166	0.095	68207	0.10
29.166	0.077	68208	0.08
29.166	0.064	68209	0.06
29.166	0.054	68210	0.05
29.166	0.852	68211	0.85
29.166	0.022	68212	0.02
29.166	0.023	68213	0.02
29.166	0.029	68214	0.03
29.166	0.019	68217	0.02
29.166	0.115	68218	0.12
29.166	0.409	68219	0.41
29.166	0.027	68220	0.03

29.166	0.043	68221	0.04
29.166	0.048	68222	0.05
29.166	0.012	68223	0.01
29.166	0.016	68224	0.02
29.166	0.033	68225	0.03
29.166	0.090	68226	0.09
29.166	0.030	68227	0.03
29.166	0.026	68228	0.03
29.166	0.071	68229	0.07
29.166	0.541	68230	0.54
29.166	0.437	68231	0.44
29.166	0.828	68232	0.83
29.166	0.101	68233	0.10
29.166	0.126	68234	0.13
29.166	1.403	68235	1.40
29.166	2.571	68236	2.57
29.166	1.056	68237	1.06
29.166	0.743	68238	0.74
29.166	0.311	68239	0.31
29.166	0.048	68240	0.05
29.166	0.020	68241	0.02
29.166	0.029	68242	0.03
29.166	0.040	68243	0.04
29.166	0.039	68244	0.04
29.166	0.019	68245	0.02
29.166	0.043	68246	0.04
29.166	0.018	68247	0.02
29.166	0.036	68248	0.04
29.166	0.022	68249	0.02
30	0.013	68332	0.43
30	0.012	68333	0.40
30	0.022	68338	0.73
30	0.001	68339	0.03
30	0.039	68340	1.30
30	0.018	68341	0.60
30	0.004	68342	0.13
30	0.002	68343	0.07
30	0.003	68344	0.10
30	0.044	68345	1.47
30	0.004	68346	0.13
30	0.005	68347	0.17
30	0.011	68348	0.37
30	0.046	68349	1.53
30	0.012	68350	0.40
30	0.036	68619	1.20
30	0.024	68620	0.80
30	0.022	68621	0.73
30	0.031	68622	1.03
30	0.009	68752	0.30
30	0.001	68753	0.03
30	0.001	68754	0.03
30	0.001	68755	0.03
30	0.001	68756	0.03
30	0.001	68757	0.03
30	0.069	68758	2.30
30	0.263	68759	8.77
30	0.051	68760	1.70
30	0.259	68761	8.63
30	0.001	68762	0.03
30	0.012	68763	0.40
30	0.245	68764	8.17
30	0.007	68765	0.23
30	0.011	68766	0.37
30	0.010	68767	0.33

30	0.012	68768	0.40
30	0.230	68769	7.67
30	0.109	68770	3.63
30	0.003	68771	0.10
30	0.008	68772	0.27
30	0.001	68773	0.03
30	0.003	68774	0.10
30	0.053	68775	1.77
30	0.007	68776	0.23
30	0.037	68777	1.23
30	0.025	68778	0.83
30	0.003	68779	0.10
30	0.005	68780	0.17
30	0.008	68781	0.27
30	0.03	68782	1.00
30	0.005	68783	0.17
30	0.003	68784	0.10
30	0.002	68785	0.07
30	0.038	68786	1.27
30	0.023	68787	0.77
30	0.007	68788	0.23
30	0.003	68789	0.10
30	0.001	68790	0.03
30	0.005	68791	0.17
30	0.017	68792	0.57
30	0.009	68793	0.30
29.166	0.114	68832	0.11
29.166	0.039	68833	0.04
29.166	0.013	68834	0.01
29.166	0.116	68835	0.12
29.166	0.026	68836	0.03
29.166	0.002	68837	tr
29.166	0.069	68838	0.07
29.166	0.056	68839	0.06
29.166	0.012	68840	0.01
29.166	0.026	68841	0.03
29.166	0.001	68842	tr
29.166	0.020	68843	0.02
29.166	0.003	68844	tr
29.166	0.019	68845	0.02
29.166	0.076	68846	0.08
29.166	0.009	68847	0.01
29.166	0.002	68848	tr
29.166	0.036	68849	0.04
29.166	0.008	68850	0.01
29.166	0.010	69901	0.01
29.166	0.044	69902	0.04
29.166	0.077	69903	0.08
29.166	0.034	69904	0.03
29.166	0.079	69905	0.08
29.166	0.006	69906	0.01
29.166	0.045	69907	0.05
29.166	0.050	69908	0.05
30	0.006	69909	0.20
30	0.034	69910	1.13
30	0.001	69911	0.03
30	0.011	69912	0.37
30	0.003	69913	0.10
30	0.040	69914	1.33
30	0.009	69915	0.30
30	0.001	69916	0.03
30	0.001	69917	0.03
30	0.001	69918	0.03
30	0.001	69919	0.03

30	0.001	69920	0.03
30	0.034	69921	1.13
30	0.002	69922	0.07
30	0.004	69923	0.13
30	0.365	69924	12.17
30	0.001	69925	0.03
30	0.001	69926	0.03
30	0.238	69927	7.93
30	0.009	69928	0.30
30	0.018	69929	0.60
30	0.012	69930	0.40
30	0.009	69931	0.30
30	0.001	69932	0.03
30	0.009	69933	0.30
30	0.041	69934	1.37
30	0.010	69935	0.33
30	0.018	69936	0.60
30	0.010	69937	0.33
30	0.019	69938	0.63
30	0.002	69939	0.07
30	0.002	69940	0.07
29.168	0.198	172958	0.20
29.166	0.187	172959	0.19
29.166	0.116	172960	0.12
29.166	0.216	172961	0.22
29.166	0.345	172962	0.35
30	0.014	251793	0.47
30	0.362	251794	12.07
30	0.066	251795	2.20
30	0.011	251796	0.37
30	0.030	251797	1.00
30	0.002	251798	0.07
30	0.033	251799	1.10
30	0.034	251800	1.13
30	0.116	251801	3.87
30	0.477	251802	15.90
30	0.181	251803	6.03
30	0.019	251804	0.63
30	0.168	251805	5.60
30	0.005	251819	0.17
30	0.010	251820	0.33
30	0.015	251821	0.50
30	0.002	251822	0.07
30	0.002	251823	0.07
30	0.012	251824	0.40
30	0.014	251825	0.47
30	0.006	251826	0.20
30	0.015	251827	0.50
30	0.015	251828	0.50
30	0.002	251829	0.07
30	0.001	251830	0.03
30	0.001	251831	0.03
30	0.003	251832	0.10
30	0.002	251833	0.07
30	0.003	251834	0.10
30	0.001	251835	0.03
30	0.001	251836	0.03
30	0.001	251837	0.03
30	0.003	251838	0.10
30	0.004	251839	0.13
30	0.001	251840	0.03
30	0.001	251841	0.03
30	0.002	251842	0.07
30	0.003	251843	0.10

30	0.001	251844	0.03
30	0.002	251845	0.07
30	0.001	251846	0.03
30	0.007	251847	0.23
30	0.002	251848	0.07
30	0.025	251849	0.83
30	0.014	251850	0.47
30	0.001	251851	0.03
30	0.001	251852	0.03
30	0.004	251853	0.13
30	0.002	251854	0.07
30	0.004	251855	0.13
30	0.001	251856	0.03
30	0.001	251857	0.03
30	0.001	251858	0.03
30	0.001	251859	0.03
30	0.003	251860	0.10
30	0.001	251861	0.03
30	0.001	251862	0.03
30	0.002	251863	0.07
30	0.001	251864	0.03
30	0.001	251865	0.03
30	0.002	251866	0.07
30	0.001	251867	0.03
30	0.005	251868	0.17
30	0.001	251869	0.03
30	0.001	251870	0.03
30	0.001	251871	0.03

Appendix D

Metallurgical Test Results

RECOVERY OF PRECIOUS METALS FROM VARIOUS GRAB SAMPLES FROM THE CONGRESS DEPOSIT

Prepared for:

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Attention:

Mr. Jasman Yee

Prepared by:

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PRA Project No.:

0503904 and 0504305

Prepared by
Gie Tan, Ph.D.
Senior Metallurgist

Reviewed by
John Huang, Ph.D.
Senior Metallurgist

Date: July 6, 2005

1.0 INTRODUCTION

Levon Resources Ltd. engaged Process Research Associates Ltd. (PRA) to carry out metallurgical testing on several samples of the Congress deposit. Two PRA projects (0307110 and 0401602) dealt with earlier materials from this area, and in this report the findings for the Lou Zone and Golden Ledge samples will be presented.

The main objectives were to characterize and test the response of the newer materials under the processing conditions established in the past.

2.0 TEST PROCEDURE

Samples brought in by the client during separate occasions were logged in, dried in a low temperature oven and assayed prior to testing. Thus, the head grades of the main constituents such as precious metals (Au, Ag), pnictides (As, Sb), total sulphur, iron, silica and gangue constituents were determined by selected methods. Standard analytical procedures that were applied, included fire assays, whole rock analysis by fusion, inductively coupled plasma (ICP) spectrophotometry with various methods of digestion, wet chemical titration, Leco furnace determinations and gravimetry. Blending and splitting of the head samples into appropriate test charges was by means of mechanical riffling.

Coarse samples were then crushed to 10 mesh prior to blending and grinding, as required. Particle size distributions were measured by screen analyses, and the selected samples were subjected to metallurgical processes that included gravity concentration, flotation, as well as cyanidation, generally in that order and under previously established conditions. The feed samples were ground to a target P_{60} (60% passing size) of 200 mesh Tyler, at a pulp density of 65% solids in a rod mill. The discharge was adjusted to 20% solids immediately before testing.

Gravity separation was conducted as a single pass through a Falcon SB40 centrifugal machine, followed by panning of the gravity concentrate to simulate a cleaner processing stage. The standard 28° bowl was used and the gradient was set to 200G with a back-water pressure of 1 psig. Excess water was decanted from the gravity tailings, which were forwarded with the pan tailings to flotation at natural pH, using ~100g/t Potassium Amyl Xanthate (PAX) and DF250 in the rougher, and $CuSO_4$ activation in the scavenger. Primary flotation was conducted in a 4L cell at 1800 RPM and a fully opened air-intake valve, and followed by one stage of cleaning all combined concentrates. The excess water was decanted from the flotation tailings prior to drying in a hot oven or forwarding to cyanide leaching. Dry weights and assays of all products were used to calculate material balances.

Bottle roll cyanidation on the Lou Zone flotation tailings was conducted with kinetic sampling, for 48 hours in 1 g/L NaCN and pH ~11 maintained by lime slurry additions. Analysis for Au, Ag, As and Sb, required taking 50 mL solution samples at the 6 and 24 hour interval, aside from regular 5 mL samples for monitoring the NaCN level. The slurry weight was tracked, and the sampling losses accounted for in the calculations. The dissolved oxygen level was checked during the run, and NaCN levels as well as the final fouling factor were determined by titration.

All of the data generated was entered into standard PRA spreadsheets that have been appended to this report. The following sections provide a brief summary of the main findings and an overview of the overall program.

3.0 RESULTS AND DISCUSSION

The initial samples received comprised coarse materials from the Lou Zone, as identified by tags and client-described labels, shown in the appended Sample Receiving Log Sheets, but abbreviated in Table 1 below. Similar samples from the Golden Ledge were submitted along with 3 smaller high-graded Lou Zone samples, and the best grade-estimates of interest are provided in Table 1.

Table 1 – Head Assay Results

Sample ID	Au g/t	Ag g/t	As %	Sb %	SiO_2 %	S^T %	Fe %
Lou North	14.6	44.1	0.40	1.18	73.5	0.53	4.21
Lou South	12.1	10.4	0.60	1.34	71.0	0.90	3.96
Lou Stibnite	196	28366	0.75	27.9	n.a.	n.a.	2.77
Lou Magnetics	5.99	362.9	0.023	0.336	n.a.	n.a.	10.3
Lou Light Fraction	1.73	350.3	0.065	0.78	n.a.	n.a.	0.45
Golden Ledge A	28.6	5.66	2.14	0.319	67.9	1.26	6.64
Golden Ledge B	5.56	2.45	0.66	0.006	66.7	0.58	6.22

The high-graded Lou sample-assays indicated that the methods used entrained more of the Ag, as compared with the largely liberated Au.

3.1 Gravity Separation

Crushed 1:1 portions of North and South Lou samples were blended into the LOU Composite, and a GL Composite was likewise prepared from the Golden Ledge A and B materials. The 2kg charges were ground to P_{60} ~200 mesh prior to testing. The pan concentrates comprised <0.1% of the mass, and the gravity test results are summarized in Table 2.

Table 2 – Gravity Separation Results

Sample ID	Gravity Gradient	Tails P_{60} µm	Pan Grade, Au	Pan Recovery
LOU Composite	200 G	56	150.7 g/t	1.2%
GL Composite	200 G	53	334.9 g/t	2.8%

Free gold is preferentially removed from the material by this operation, yielding a flotation feed (gravity + pan tails) that had a P_{60} ~270 mesh representing the bulk of the mass but containing the mineral-associated precious metal values mainly. Only small amounts (1-3%) of coarse free gold were removed prior to flotation.

3.2 Flotation Results

The response of the two composites to flotation is summarized in Table 3, and it would appear that comparable results were achieved, considering the respective differences in head grades. Thus, rougher-scavenger recoveries for Lou material were 43% Au, 36% As and 59% Sb, while those for the GL Composite were 62% Au, 60% As and 52% Sb with lower mass rejection to comparable tailings grades.

Table 3 – Overall Flotation Grade Results

Sample	Grades, % or g/t Au				Recovery, %				
ID	Au	As	Sb	S^T	mass	Au	As	Sb	S^T
LOU Cl. Conc.	52.6	2.11	12.6	8.44	5.1	22.7	23.4	50.0	58.3
Sc.Tailings	8.0	0.35	0.63	0.25	84.7	53.0	63.7	40.8	28.5
Head (calc.)	11.9	0.46	1.30	0.74	100	100	100	100	100
GL Cl. Conc.	74.0	5.36	0.69	5.16	9.4	44.5	40.7	45.6	64.4
Sc. Tailings	7.6	0.64	0.09	0.18	77.4	37.6	40.0	48.4	18.5
Head (calc.)	15.6	1.24	0.14	0.75	100	100	100	100	100

A slight difference in behavior is apparent from the Grade vs. Recovery curves that are plotted in Figures 1 and 2, respectively. For the Lou Zone, only the Sb and S primary recoveries are >50%, whereas for the GL Composite all but the Fe attain this level. This could indicate that the Lou Zone samples, reportedly taken from surface pits, could have been tarnished by oxidation. Hence it was of interest to check if cyanidation could recover the residual values on this material.

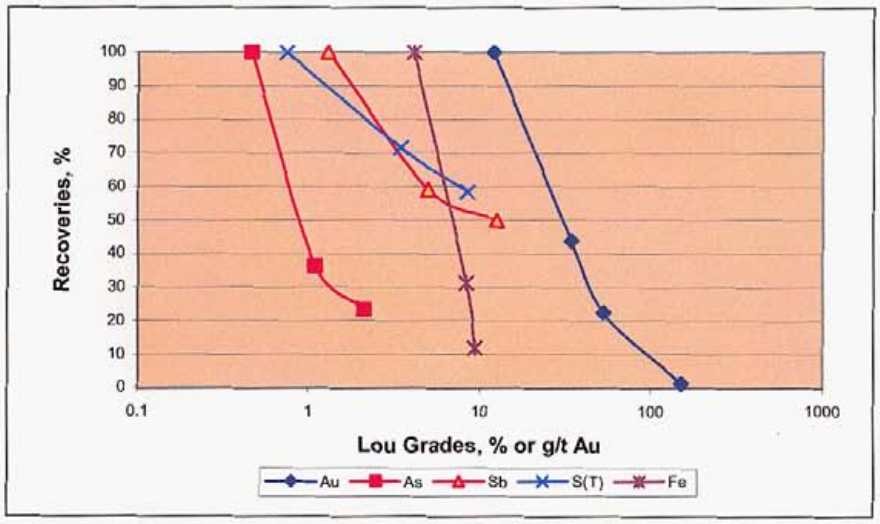

Figure 1 – Lou Zone Grade Recovery Curves

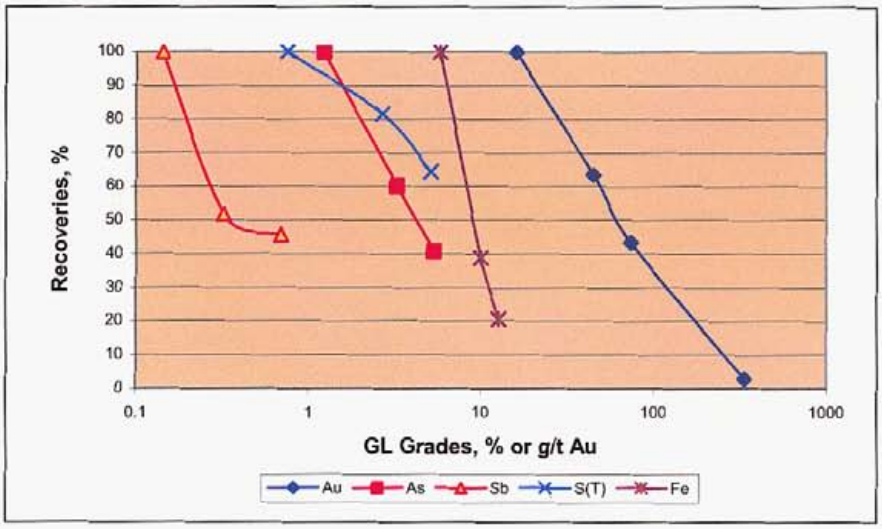

Figure 2 – Golden Ledge Grade Recovery Curves

3.3 Cyanidation Test

Bottle roll cyanidation of Lou Zone flotation tailings containing about 7.5 g/t Au and 9.5 g/t Ag, extracted ~90% of these precious metals in 48 hours at pH 10.5, a pulp density of 40% solids, and with 1 g/L NaCN. Reagent consumptions were 1.14 kg/t NaCN and 4.09 kg/t lime, while the residue graded 0.7 g/t Au and 1 g/t Ag. The solution profiles are shown in Figure 3.

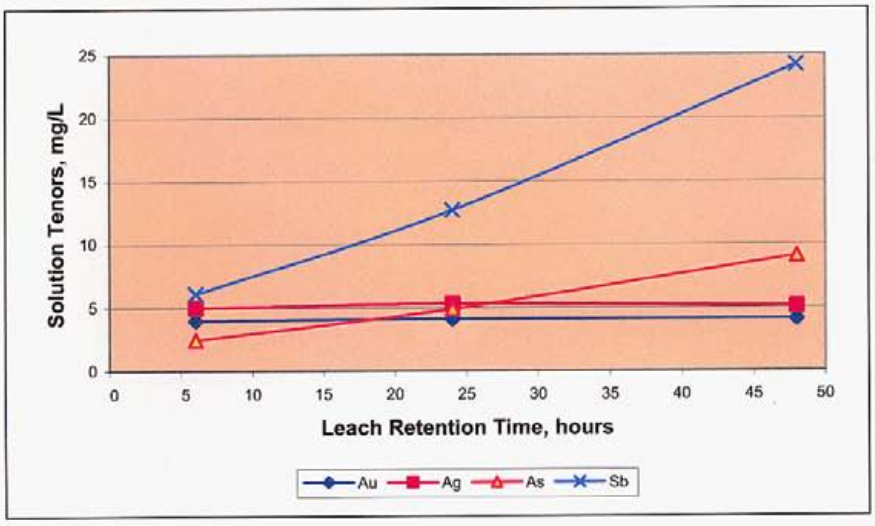

Figure 3 – Lou Zone Cyanidation Kinetics

No increases in the Au and Ag concentrations were noticed past the 6-hour mark, and this retention time and/or a lower NaCN level could lower the reagent requirements substantially.

4.0 CONCLUSIONS AND RECOMENDATION

The metallurgical test results indicated that the silver is less liberated than the gold in the Lou Zone samples. The LOU Composite residues after gravity, flotation and cyanide leaching came down to <1 g/t Au and Ag in <85% of the mass, indicating overall recoveries of 95.7% Au and 96.9% Ag. Leaching of the flotation tailings in <1 g/L NaCN and/or on the order of 6 hours retention time may be attempted.

The flotation tailings of the GL Composite still contained 7.6 g/t Au in 77.6% of the mass, to indicate an overall recovery of 65.6% Au. In general, however, its response to gravity and flotation was much better than that of the Lou Zone, likely due to higher Au and As grades, as well as a lesser degree of tarnishing. Further testing is required to assess alternative means of treating these materials.

Appendix E

Author's Statement of Qualifications

Appendix E: Author's Statement of Qualifications

I, David St. Clair Dunn, Professional Geoscientist, with a business address of 1154 Marine Drive, Gibsons, B.C., Canada, certify that:

1. I am a graduate of the University of British Columbia, Vancouver, B.C. and hold a degree of Bachelor of Science in Geology.

2. I have practiced my profession as a prospector and geologist for 35 years.

3. I am registered as a Professional Geoscientist with the Association of Professional Engineers and Geoscientists of the Province of British Columbia (Reg. # 18,479). I am a Fellow of the Geological Association of Canada and a member of the Association of Exploration Geochemist's, the Canadian Institute of Mining, Metallurgy and Petroleum, the Honorary Advisory Board to the B.C. and Yukon Chamber of Mines, the Society of Economic Geologists and the Mining Exploration Group. I am the qualified persons for the purposes of National Instrument 43-101 in reference to this report.

4. I directly supervised the summer 2005 trenching and diamond drilling program on the Congress Property.

5. I am the sole author of this report.

6. I am not aware of any material fact or material change from the information in this report that would make the report misleading.

7. I consent to the use of this report for the purpose of a private or public financing.

Signed:

David St. Clair Dunn, P.Geo.

LEGEND

3	PLUTONIC ROCKS		CONTACT
2	VOLCANIC ROCKS		BEDDING
1	SEDIMENTARY ROCKS		FOLIATION
a	GREY CHERT		AXIS
b	BLACK ARGILLITE		VEIN
c	GREEN SILTSTONE		TRENCH
d	WHITE LIMESTONE		FAULT
e	GREEN BASALT		OUTCROP
f	PURPLE BASALT		PROPERTY BOUNDARY
g	GREEN TUFF		MAP AREA ZOOMED
h	GREEN GABBRO		ADIT
i	FELDSPAR PORPHYRY		LOU DECLINE
j	FELDSPAR HORNBLENDE		CONTOUR (METERS)
k	APLITE		MMI Sampling
m	SERPENTINITE		Proposed Underground Development
n	MASSIVE		
o	SHEARED		
p	BEDDED		
q	JOINTED		
r	PILLOWED		
s	AMYGDALOIDAL		

SLIDE ZONE
PAUL ZONE
GUN CREEK
GUN ZONE
OZONE ZONE
LOU ZONE
HOWARD ZONE
CONGRESS MINE
PROPOSED ADIT
UPPER HOWARD ADIT
LOWER HOWARD ADIT
ADIT NO-1
ADIT NO-2
CONGRESS ADIT NO-3
MMI 8+50N
MMI 7+50N
MMI 6+50N
MMI 8300
MMI 8100
MMI 7880
LZ
HZ7
166443 / 500 ppb Au Soil
Brushed to here
Proposed Drilling
Cliff
Talus
Steep slope
Steep cliffs
Outcrops of quartzite some orange weathering
All quartzite to Gun Creek black and white color tiny QVs some very strong shear / fault zones
Quartzite / chert
Shale arkosic sst. & quartzite
Rusty quartzite
Green and purple fine grained calcareous andesite (weathers orange)
All sheared mixed sedimentary rocks
Sheared chert
Sheared qzite pebble
Stongly sheared quartzite, argillite with vein quartz, graphite
Black argillite lime
Calcareous chloritized volcanic rocks
Strongly sheared argillite with cc, marlposite, VQ graphite
Pyritic limey green-purple fine-med grained volcanic rocks
Gossaned F P...tr pyrite
Purple / green volcanics
Possible outcrop purple volcanic rocks
(Serpentine Layered Float)
"greenstone" pale altered fine grained volcanics
Strongly sheared volcanic rocks much black chlorite
Limey & chloritic basaltic volcanic rocks
Weakly sheared purple volcanic rocks, minor QVs, much ep, crumbly fracture pattern probably pillowed
Sheared purple volcanic rocks
Limey sheared basalt with some pillow imdures
Fine grained green andesite
Strongly fractured / jointed orange feldspar porphyry dyke some cc vits no quartz
Purple pillow lava
Chloritized fine grained basaltic flows
Fractured hematitic
Dark green / purple volcanic
Chloritic
Brown alteration
Sheared strongly hematic sections
Pale grey volcanic rocks
Purple / green vesicular andesite (pl.)
Serpentinized vesicular pl. ep
Green volcanic rocks serpentinous with ep
Shattered pillowed green andesite
Finegrained volcanics
Volcanics
Sheared pillowed volcanics
Strongly sheared pillowed basalt
Orange weathering
Strongly sheared Fergusson Gp sediments
Orange weathering 10% feldspar phenocryste
Fragmental
F P grey matrix chalky white fs phenocrysts
Crest of ridge
Vesicular texture
Sheared purple / green volcanic rocks
Steep chert (Fergusson?)
Pillowed hematitic volcanic rocks
Black-grey-white quartzite (Fergusson Group sediments) some orange weathering
Minor species of rusty quartz very well mineralized Would need down deep
Property Boundary
Map Area Zoomed
0 500 1,000 meters
0 125 250 metres
After: Cooke, 1984
GEOLOGICAL SURVEY BRANCH ASSESSMENT REPORT
28,346

LEVON RESOURCES LTD.		
CONGRESS PROPERTY		
PROPERTY GEOLOGY		
DAVID ST. CLAIR DUNN, P.GEO.		
Nad 83 (Zone 10)	Scale: 1: ~~2500~~ 2358	MAP: 1
Date: NOV 05, 2005	Drawing: TerraCAD	

LOOKING EAST

CHERT / ARGILLITE

BASALT

TRENCH 5
0.38 / 6.1 m

TRENCH 4
1.39 / 3.0 m

TRENCH 3
0.87 / 0.61 m

800 Level

TRENCH 2
0.32 / 5.4 m

TRENCH 1
0.25 / 3 m

MAIN ROAD

BASALT

GABRO

MICRO GABBRO & BASALT

BASALT

690 Level

660 Level

HOWARD FAULT

X - CUT 3

X - CUT 2

X - CUT 1

84-1 0.08 / 1.40 m
84-2
84-3 0.10 / 1.52 m
84-4 0.10 / 1.19 m
84-5 0.11 / 1.49 m
85 d 20 0.257 / 0.40 m 0.101 / 1.00 m
79-1 0.064 / 0.76 m
79-1 0.202 / 1.01 m
79-3 0.006 / 0.98 m
79-3 0.040 / 1.45 m 0.31 oz / 1 kg
79-3 0.100 / 2.44 m
79-3 0.301 / 0.98 m
79-4 0.006 / 0.92 m
79-4 0.020 / 1.69 m
79-5 0.106 / 1.58 m
72-2 0.546 / 1.39 m
80-2 0.06 / 0.67 m
80-1 0.182 / 0.91 m 0.120 / 1.00 m
80-4 0.14 / 1.05 m
80-2 0.12 / 3.10 m
80-3 0.53 / 1.00 m
80-6 0.005 / 1.98 m
80-7 0.95 / 8.23 m
80-8 0.268 / 0.46 m
88-8 >0.01 / 5.20 m
88-7 0.154 / 7.15 m
88-7 0.249 / 3.05 m
85 d 10 0.059 / 0.46 m
85 d 10 0.43 / 2.70 m
C-04-02 3.7oz of 1.8 g/t
C-04-01 1.9m of 3.0 g/t
C-04-03 0.54m of 12.14 g/t
C-05-04 0.60m of 7.93 g/t (West Howard)
C-05-06 1.0m of 6.4 g/t
85-16 >0.01 / 393 cm
87-20 (Zone 2) 0.181 / 1.31 m
87-6 0.447 / 1.68 m
87-19, 20, 21 0.097 / 1.08 m 0.047 / 1.36 m
SA-28 0.63 / 3.35 m
65-4 0.64 / 1.52 m
88-2 1.491 / 0.46 m 0.20 m true
0.314 / 0.43 m 0.19 m true
88-3 0.214 / 0.67 m 0.31 m true
Zone 2 0.075 / 3.0 m 2.31 m true
88-4 0.054 / 1.22 m
88-5 0.089 / 0.91 m
88-6 0.108 / 0.60 m
88-8 0.004 / 0.30 m
88-9 0.010 / 1.28 m
88-10 tr / 2.27 m
88-7 tr / 0.67 m
88-11 0.326 / 2.0 m 1.26 m true
SA-29 0.16 / 0.91 m 0.146 / 1.00 m
65-5 0.34 / 0.82 m 0.28 / 1.00 m
87-10 0.001 / 0.23 m
87-11 0.023 / 4.57 m
87-12 0.728 / 3.66 m 2.76 m true
87-15 0.321 / 2.30 m 2.26 m true
87-12 0.032 / 4.18 m 3.66 m true
87-14 0.062 / 2.13 m
87-15 0.013 / 3.50 m 2.48 m true

LEGEND

- Diamond drill hole showing gold in grade in grams / tonne & Estimated true width intersections in meters
- Underground survey stations
- TRENCH 1 — Trench showing gold grade in ounces / short ton & width of intersection in meters
- Underground workings on vein showing length of interval in meters, gold grade in ounces / short ton & average width of vein

0 50 100
meters

Assessment Report
28,346

LEVON RESOURCES LTD.

CONGRESS PROPERTY
HOWARD ZONE
LONGITUDINAL SECTION LOOKING EAST
PIERCE POINTS OF DRILL HOLES

DAVID ST. CLAIR DUNN, P.GEO.

N.T.S. 92J / 15W	Scale: 1: ~~600~~ 480	MAP: 2
Date: Nov, 2005	Drawing: TerraCAD	

AFTER CHRISTOFFERSEN J., 1988

-100 mE | 0 mE | 100 mE | 200 mE | 300 mE | 400 mE | 500 mE | 600 mE | 700 mE | 800 mE

1000 m Z

900 m Z

800 m Z

700 m Z

600 m Z

500 m Z

400 m Z

300 m Z

TR-28
1.80m x 1.87 g/t

TR-29
6.00m x 2.49 g/t

TR-13
3.93m x 6.22 g/t

TR-30
3.92m x 4.04 g/t

TR-26
1.74m x 37.32 g/t

TR-31

TR-27
2.71m x 7.78 g/t

TR-32
1.01m x 5.53 g/t

TR-33

ROAD
TR-10
1.01m x 3.11 g/t

TR-56

N 10° E

85d25
3.54m / 1.24g/t

85d15
1.49m / 0.62g/t

2.01m / 1.56g/t 85d21
2.99m / 0.93g/t 85d22
4.51m / 1.24g/t 85d24
NSV 85d12
2.1m / 0.93g/t 85d13
2.93m / 1.87g/t 85d14

85d16
2.01m / 7.78g/t
85d17
3.51m / 0.62g/t
85d19
3.51m / 0.31g/t

85d30
0.34m / 5.91g/t
85d32
3.38m / 11.51g/t
85d33
0m / 0g/t
86d34
1.98m / 1.24g/t
86d35
0.94m / 5.91g/t
86d36
0.46m / 5.91g/t

LZ-05-04
1.0m / 3.3g/t

86d37
2.01m / 11.2g/t
86d39
2.41m / 1.24g/t

LZ-05-03 (NS
NSV

86d41
0.24m / 2.18g/t
86d41
0.7m / 1.56g/t

86d40
0.55m / 10.89g/t
86d42
0.85m / 2.8g/t
86d41
2.01m / 1.56g/t
86d40
0.91m / 8.71g/t

POWERLINE

86d49
2.65m / 9.64g/t

GOLDBRIDGE ROAD

86d43
0.7m / 3.42g/t

CARPENTER LAKE

86d45
2.99m / 2.8g/t
86d46
1.19m / 6.22g/t
86d46
1.89m / 3.73g/t
86d44
2.16m / 2.8g/t

86d47
0.55m / 13.06g/t
86d47
2.99m / 1.87g/t

LEGEND:

- MINERALIZATION: ALTERED/VEINED
- PLUTONIC: FELDSPAR/HORNBLENDE PORPHYRY
- VOLCANIC: GREEN/PURPLE FLOW
- SEDIMENTARY: CHERT/ ARGILLITE

SYMBOLS:

- 85d18 / 6.6 m x 12 — DRILL HOLE
- TR-33 / 6.6 m x 12 — TRENCH
- CONTACT: OBSERVED, INFERED
- FAULT: OBSERVED, INFERED

0 25 50
metres

LEVON RESOURCES LTD

CONGRESS PROPERTY
LOU ZONE
LONGITUDINAL SECTION LOOKING EAST
BRIDGE RIVER AREA, LILLOOET MINING DIVISION, B.C.

DAVID ST. CLAIR DUNN, P.GEO.

N.T.S. 92J / 15W	Scale: 1: ~~1000~~ 750	MAP: 3
Date: Nov. 2005	Drawing: TerraCAD	

50 mE
100 mE
150 mE
200 mE
250 mE
300 mE
350 mE
400 mE
450 mE
500 mE

850 m Z
800 m Z
750 m Z
700 m Z
650 m Z
600 m Z

86d51
4.17/0.85m
6.84/1.49m

86d50
10.89/0.36m

86d52
9.64/2.19m
7.40/0m

BLOCK C1
6824 Tonnes
5.29 (0.187)
15m
28m
5.29 / 28m
5.29 / 1.40m
116m
63m
9.33 / 0.6m

Portal (Level 1)
750.2m

BLOCK C2
28592 Tonnes
6.03 (0.213)
59m

Portal (Level 2)
710.0m

151m
48m
10.51 / 0.44m
6.86 / 1.25
79.5m
6.78 / 0.4m

BLOCK C3
39,624 TONNES
6.22 (0.222)
71m

2.36 / 1.53m
49.5m
141m
5.91 / 1.54m

Portal (Level 3)
653.6 (Lake Level)

44m

BLOCK C4
19065 Tonnes
7.00 (0.249)
9.42 / 1.62m
61m

(Level 4)
622.5m

15m

BLOCK C5
4076 Tonnes
9.42 (0.334)

0 12.5 25
metres

25m = 44mm
1 : 568

Legend:

86-1
9.6 / 1.65m
Diamond drill hole showing gold grade in grams / short ton @ intersection width in meters

m
m
9.6 / 1.65m
15m
Ore Reseve Block showing tonnage in metric tonnes & gold grade in grams / short ton with ounces / ton following in brackets; true external dimensions of block shown in metres

LEVON RESOURCES LTD

CONGRESS PROPERTY
LONGITUDINAL SECTION
CONGRESS VEIN
ORE RESERVE BLOCKS
LOOKING EAST
BRIDGE RIVER AREA, LILLOOET MINING DIVISION, B.C.

DAVID ST. CLAIR DUNN, P.GEO.

N.T.S. 92J / 15W	Scale: 1: ~~500~~ 568	MAP: 4
Date: Nov, 2005	Drawing: TerraCAD	