

HYDROMETALLURGY

Calculate countercurrent washing efficiency with dirty wash solution

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ABSTRACT

Countercurrent decantation circuits (CCD) are often employed in hydrometallurgy to effect solids-liquid separation in combination with a barren solution wash to reduce losses of valuable solution component(s) from the tails. Bechtel Corporation uses its BEMSIM flowsheet-modelling PC computer program to simulate a CCD circuit where the "dirty" wash solution contains significant amounts of soluble components. An algorithm successfully models this condition and is simplified to a single expression when the wash solution is a "clean" solution containing no solubles. The CCD calculation module is compared to operating data from uranium and nickel-cobalt CCD circuits.

Introduction

Hydrometallurgical process schemes have often relied on countercurrent washing methods to reduce soluble losses of valuable product in the tails. Commonly used equipment to effect this operation is a CCD circuit of N thickeners in series, where a slurry is fed to thickener 1 and a wash solution to thickener N; clear pregnant solution overflows thickener 1 and washed tails exit as underflow from thickener N. CCD circuits of this type are typically employed when the solids tend to slime heavily and where a filter is not economic.

A typical CCD circuit usually employs from two to five thickeners and can employ either a "clean" washing solution of, say, pure water or else the hydrometallurgical circuit may generate a "dirty" barren solution containing an appreciable amount of soluble component which may be advantageously returned to the process as CCD wash solution. Various authors have treated the case for clean washing solutions (Svarkovsky 1977, Merritt 1970). For the case of dirty wash solutions, an algorithm has been developed to enable efficient modelling by the BEMSIM computer program, because the need for an accurate method to calculate soluble losses

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of the Mining and Metals Operations of Bechtel where he sponsors use of their BEMSIM computer program.

is required when performing an economic trade-off analysis of the value of product lost in the tails as a function of the number of thickeners.

Technical Background

The steps involved in developing a calculation method start from the basic description of one thickener in a CCD circuit shown schematically in Figure 1. (In actual practice the two feed streams can be mixed together in a launder, feed well, pump sump or externally agitated mix tank.) Reference to Figure 1 shows two feed streams 1 and 2, two product streams 3 and 4, and two paths for stage inefficiency due, respectively, to solution by-pass and to imperfect mixing. In the former case, feed pulp solution travels directly to the thickener exit stream 3 without undergoing displacement washing, and in the latter case the mixing of the two feed streams is non-uniform.

For thickener terminology, the over-all liquid phase volumetric flowrates of each of the two input and the two output streams is designated V1, V2, V3 and V4. For an individual stage, a subscript is added to indicate the stage number n. The mass flowrate of soluble component(s) in the stage n solutions is designed by subscript j for $1 \le j \le m$ soluble components, such as $Cl_{j,n-1}$ for the feed pulp.

Stage-to-stage calculations for mass flows use the following assumptions, assuming for a first iteration that underflow pulp per cent solids are identical from each stage:

• Overflow volume from each stage is uniform except for n = 1.

$V4_n = V2_{n+1}$				2≤n≤N	•••••	(1)
	_					

- $V4_1 = V1_{n-1} + V2_{n+1} V3_1$ n = 1(2)
- Underflow volume of solution from each stage is uniform, fixed by the volume per cent solids.

$$V3_n = V3_{n+1}$$
 $1 \le n \le N$ (3)

 Underflow solids undergo no change in volume nor is there any leaching of soluble components during residence time in the CCD circuit. Solids volume is ignored in the calculations.

 Except for flow quantities due to stage inefficiency, the mixing of the remaining quantities of pulp and wash solution assumes ideal mixing.

Stage Efficiency

The key to calculation of soluble component(s) loss is the definition of stage efficiency. Figure 1 shows the two major sources of stage efficiency which can be identified as E_1 and E_2 . It is improbable that the user of the computer model will know E_1 separately from E_2 and, what is more, the values of both can vary appreciably from stage to stage (Emmett 1981). As a simplification, the two efficiency parameters E_1 and E_2 can be lumped together into a single stage efficiency E_1 , which is equated to the percentage of solution $V1_{n-1}$ which reports directly to $V3_n$ without having been subjected to a mixing or washing effect. Using CCD circuit operating data, it is possible to replace stagewise variations of E_1 with
 TABLE 1. CCD circuit at 3.3:1 wash ratio with dirty wash solution

Feed pulp	Dirty wash	Tailings solution	Pregnant liquid		
Stream 1	Stream 2	Stream 3	Stream 4		
Liq. U ₃ O ₈ in	Liq. U ₃ O ₈ in	Liq. U ₃ O ₈ in	Liq.		
Flow Solution	Flow Solution	Flow Solution	Flow		
m ³ /h mg/L	m ³ /h mg/L	m ³ /h mg/L	m ³ /h		
112.0 170.6	325 4.9	98.5 7.9	333.8		

a global stage efficiency value, which can be backed-fitted when concentration values $C_{j,n}$ are not known for each stage. Then, one can establish the following relations when E_1 is the global efficiency (by-pass plus mixing inefficiency for incoming solution in the stage n pulp feed); $B3V_n$ is the volume of feed pulp solution to stage n which bypasses and reports to tails, and $B3C_{j,n}$ is the mass flow of soluble component j in the bypass solution B3V:

$$B3V_n = (1 - E_1/100) * V1_{n-1}$$
 (4)

$$B3C_{j,n} = (B3V_n/V1_{n-1})*C1_{j,n-1}$$
(5)

$$C3_{j,n} = B3C_{j,n} + [(V3_n - BV3_n)/(V1_{n1} + V2_{n+1})] * \dots (6)$$

(C1_{j,n-1} + C2_{j,n+1} - BC3_{j,n})

If the per cent soluble loss of a component is defined as $\ensuremath{\mathsf{PSL}}_j,$ then

 $PSL_{j} = 100*[C3_{j,N}/C1_{j,0}]$ (7)

and the total per cent soluble loss for all soluble components is

(PSL) _{total} =	m Σ	PSLj	 (8)
	j = 1		

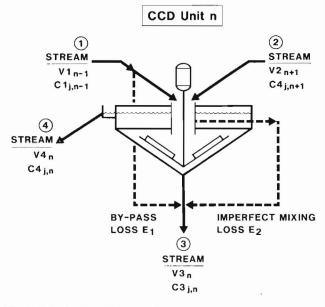


FIGURE 1. Washing thickener unit.

A numerical example for a single soluble component is illustrated in Figure 2 for an assumed stage efficiency of 85% and for a single stage of CCD washing with a dirty wash solution at a wash ratio of 3:1, where the wash ratio is defined as the volume of wash solution per volume of solution in the final tails.

 $B3V_1 = (1.0 - 0.85) * 200 = 30 L$ (9)

TABLE 2. Results from the uranium CCD circuit simulation

Bechtel M&M Operations: B	EMSIM Simulator	Program			[13:59 9/ 4/19	Page 1					
Uranium Leach Solids/Liquid CCD Circuit - Blyvooruitzicht Mill, RSA 5-Stage Counter-Current Decantation Washing @ 75.5% Stage Efficiency Reference: CIM Bulletin, May 1979, pages 127-134 Authors: D.W. Boydell, P.A. Paxen, D.W. Bosch, W.G. Craig Material Balance from Table 5, pg 137, Day -2.											
CD100 : CCD CIRCUIT	[CCD : 11] [MASS BALAN	ICE CLOSURE	.0000]	ENERGY BALANCE CLOSURE						
[OUTLET TEMP OF U/FL [NUMBER OF STAGES (1 [STAGE EFFICIENCY, %	-7) =	.0] (OUTLET TEM 5.] [UNDERFLOW .50]	P OF PREG. SOLIDS, X	= .0] = 55.901	(HEAT LOSS FACTOR [OVERFLOW SOLIDS, PPM [PCT SOLUBLE LOSS	= .0) = .00) = 3.856)					
	=====INP	UT STREAMS=====		OUTPUT STREAMS=	=====						
	1 PULP FEED TO	2 CCD BARREN WASH	3 TAILS PULP	4 PREG LIG	UOR						
IQUID PHASE											
LIQUID MASS FLOW - t/h LIQUID VOL FLOW - m3/h LIQUID S.G U308 - t/h H20 - t/h	1.026 ===	=== 1.000 ===	049 .00080	.0081 .01990	gpl ===== .0587						
OLID PHASE											
SOLID MASS FLOW - t/h SOLID S.G LEACHED ORE - t/h	128.1000 Wt 2.600 === 128.1000 100	000	Pct 128.1000 4 === 2.600 = .00 128.1000 1	2.600							
OTAL STREAM				÷							
TOTAL MASS FLOW - t/h Total vol flow - m3/h WT. PCT. Solids - percent Total S.G	243.0000 161.3 52.72 1.507	325.0000 325.0 .00 1.000	229.1592 147.8 55.90 1.551	338.8408 338.8 .00 1.000							

TABLE 3. Results from the Ni-Co circuit simulation

			Nickel-Cob						ot Plant (Data			
			5-Stages o Uses 98% S			h "Dirty	Con	nputes H	eat Baland		х.		
CCD	: CCD (CIRCUIT	[CCD :	11]	(MASS E	ALANCE (.0001]	[ENERGY BAL		••••	.00
	COUTLET T	TEMP OF U/FL	0⊌ =	63.61	COUTLET	TEMP OF	PREG.	=	42.11	[HEAT LOSS F/	CTOR =		48.71
	INUMBER (OF STAGES (1	-7) =	5.1			DS, X		47.00]	OVERFLOW SOL	IDS, PPM =		.001
	(STAGE EI	FFICIENCY, X	z	98.001						[PCT SOLUBLE	LOSS =	: 12	2.895]
	(HEAT OF	REACTION	=	.00	(EXTERN		ADDED	=	.001	[AMBIENT HEAT	LOSS	. 777	04.011
				= I NPUT					STREAMS=				
			7		100		11		12				
			CCD Feed		Total Was		CCD Under		Preg. So				
Liquid I	Phace												
	d Mass Flow	- kg/h	1217.91		800.09		379.23		1638.77				
	d S.G.	-	1.198		1.102		1.102		1.170	2.21			
	d Sp. Ht. d Vol Flow		.806. 1016.6	gpl	.866	gpi ======	.860 344.1	gpl	.806	gpl ======			
Ni		- kg/h	8.56	8.42	.04	.06	.08	.23		6.08			
Co		- kg/h	.275	.27	.005	.01	.004	.01					
Fe		- kg/h	4.25	4.18	.13	.18	.09	.26					
Mg		- kg/h	25.36	24.95	16.80		7.96	23.13		24.42			
AL		- kg/h	3.01	2.96	1.48	2.04	.71	2.05		2.70			
Mn		- kg/h	1.74	1.71	1.13	1.56	.54	1.56		1.67			
Cr		- kg/h	.47	.46	.39	.54	.18	.53	.68	.48			
SiOZ		- kg/h	25.44	25.02	8.42	11.60	4.08	11.85	29.78	21.26			
Free /	Acid	- kg/h	70.15	69.00	5.54	7.63	3.05	8.87	72.64	51.86			
Sulfat	te	- kg/h	157.55	154.98	74.36	102.42	35.55	103.31	196.36	140.19			
Water		- kg/h	921.10	906.05	691.80	952.84	326.99	950.20	1285.91	918.07			
Solid P	hase												
Solid	Mass Flow	- kg/h	336.30	Wt Pct	.00	Wt Pct	336.30	Wt Pct	.00	Wt Pct			
Solid	S.G.	-		=====	.000	******		======					
Gangue	e	- kg/h	336.300	100.00	.000	.00	336.300	100.00	.000	.00			
fotal Si	tream										*		
Total	Mass Flow	- kg/h	1554.21		800.09		715.53		1638.77				
	Vol Flow	- L/h	1115.5		726.0		443.0		1400.7				
	t. Solids	- PERCENT	21.64	•	.00		47.00		.00				
	n S.G.	1	1.393		1.102		1.615		1.170				
	rature	- deg C	97.6		80.0		63.6		42.1				
	n Sp.Ht.	- cal/g-C	.686		.866		.573		.806				
Stream	n Enthalpy	- kcal/h	104013.0		55430.6		26102.0		55637.5				

 $B3C_{1,1} = (30/200)*100 = 15 g$ (10)

 $C3_{1,1} = 15 + [(100 - 30)/(200 + 300)]*(100 + 51 - 15) = 34 g$ (11)

 $PSL_1 = 100*(34/100) = 34\%$ (12)

Equations (1) through (7) were employed to construct an iterative procedure for the general case where $2 \le N \le 7$, $1 \le j \le 40$ and

TABLE 4.	CCD	circuit	performance	with	dirty	wash
solution						

	•	5 tails am 11	Stage 1 preg. liq. Stream 12			
	Measured	Calculated	Measured	Calculated		
Temp., °C	65.0	63.6	41.7	42.1		
Ni, kg/h	0.12	0.08	8.52	8.52		
Co, kg/h	0.008	0.004	0.272	0.276		
Fe, kg/h	0.22	0.09	4.16	4.29		
Mg, kg/h	0.40	7.96	41.76	34.20		
Al, kg/h	0.67	0.71	3.82	3.78		
Mn, kg/h	0.64	0.54	2.23	2.33		
Cr, kg/h	0.22	0.18	0.64	0.68		
SiO ₂ , kg/h	5.18	4.08	28.68	29.78		
Acid, kg/h	5.20	3.05	70.48	72.64		

for each stage or varying values of E_1 and E_2 for each stage when their values are known.

Examples

Two examples are given where the washing solution contains soluble components. The simpler Example 1 is a uranium washing circuit with only one soluble component. In Example 2 the wash solution contains 10 soluble components.

where stage n underflow pulps have different volume per cent solids. If warranted, this procedure can employ a different value of E_1

Example 1 - Uranium Circuit

This example of CCD circuit from an operating uranium plant (Boydell, *et al.* 1979) is selected to illustrate the calculation method; it uses the data for washing at a wash ratio of 3.3:1. Operating data was modified in a minor way for internal compatibility and ignored any solubilization of uranium during CCD washing.

Data from Table 1 were used as input to the BEMSIM program at an over-all stage efficiency of 75.5%. Output is shown in Table 2, where it is seen that the calculated U_3O_8 loss in the tails solution of 8.1 mg/L is quite close to the value reported in Table 1. Soluble loss is calculated at 3.86%, compared to the reported value of 4.14%.

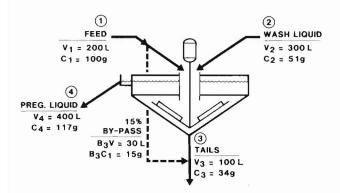


FIGURE 2. Numerical example of thickener at 85% stage efficiency and wash ratio of 3:1.

Example 2 — Nickel-Cobalt Circuit

This example of a five-stage CCD circuit came from operating data of a pilot plant. The feed slurry to the CCD at 97.6°C was the flash slurry from autoclave leaching of nickel-cobalt laterite, and contained 8.56 kg/h Ni⁺⁺, 0.275 kg/h Co⁺⁺ and 25.36 kg/h Mg⁺⁺. The dirty wash solution at 80°C contained 0.04 kg/h Ni⁺⁺, 0.005 kg/h Co⁺⁺ and 16.8 kg/h Mg⁺⁺.

This CCD circuit employed separate, agitated interstage mixers and oversize thickeners, and the computation results using 98% stage efficiency demonstrate the high efficiency of this configuration. Input and output data are shown in Table 3. Measured and experimental results are shown in Table 4.

The input value of 98% stage efficiency was selected to make the calculated loss for Ni⁺⁺ equal to the measured value in the pregnant solution. At this value, the measured value for total soluble loss was 10.9%, compared to a calculated value of 12.9%.

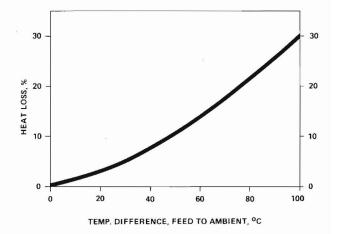


FIGURE 3. Heat loss from a single CCD stage as per cent of input enthalpy (feed + wash).

CCD Washing with Clean Wash Solution

For the special case when there is only one soluble component and the wash solution contains no soluble component, the set of iterative equations (4) through (6) is replaced by a simple expression for N stages. Let

$N = Number of stages \dots (13)$
$r_o = Feed ratio = V4/V1$ (14)
$r_1 = Wash ratio = V2/V3$ (15)
$A = \frac{r_1 + (1 - E_1)}{(1 - E_1) r_0 + 1} $ (16)

TABLE 5. Results from uranium CCD circuit with clean wash

Bechtel	M&M Operat	tion	S: BE	MSIM Simu	lator Pr	ogram					[14:17 9/ 4/	1990]	Page
				5-Stage C	ounter-C : "The E	urrent De xtractive	cantatio	rcuit - Pe n 288.6% S rgy of Ura CSMR	Stage Ef anium",	ficiency	0.		
CD 100	: CCD (IRC	UIT	[CCD	: 11]	[MASS	BALANCE	CLOSURE		.0000]	[ENERGY BALANCE CLOS	URE	
		OF S	OF U/FLO TAGES (1- IENCY, %		.0 5. 88.60		T TEMP O	F PREG. IDS, %	:	.0] 59.00]	[HEAT LOSS FACTOR [OVERFLOW SOLIDS, PPM [PCT SOLUBLE LOSS	ב ב ב	.0] 200.00] .498]
				*******	== I NPUT	STREAMS=	***		=OUTPUT	STREAMS=			
				1 CCD FEED		2 WASH LIQ	Jor	3 TAILINGS		4 PREG SOLU	JTION		
		-		92000. 179. 1.026 268.00 91732.	gpl ===== 2.989		gpl .000	1.33	gpl .020 1025.98		gpl ===== .978 999.022		
OLIDS											10.0-0		
SOLIDS LEACHEI	MASS FLOW D ORE	-	lb/h lb/h	100000. 100000.	Wt Pct 100.00	0.	Wt Pct .00	99945.	Wt Pct ****** 100.00	55.	Wt Pct ===== 100.00		
TOTAL ST	REAM												
TOTAL	MASS FLOW VOL FLOW SOLIDS	-	lb/h gpm PERCENT	192000. 255. 52.08		250000. 500. .00		169399. 211. 59.00		272601. 545. .02			

$$B = \frac{r_1 + (1 - E_1)}{(1 - E_1)r_1 + 1} \dots (17)$$

then,

$$PSL = 100 * \frac{r_o - 1}{r_o A B^{N-1} - 1} \dots (18)$$

As an example to illustrate equation (18), the operating data from a uranium mill (Merritt 1970, pp. 460-461) were used as input to the BEMSIM program. The reported soluble loss of uranium of 0.5% was achieved when the stage efficiency of the 5-stage CCD circuit was set equal to 88.6%, as shown on the computer results in Table 5.

Heat Balance

For the case illustrated by Example 2, the CCD circuit feed was at or near boiling, and because the thickeners were uncovered there was appreciable heat loss due to evaporation from the surface of the thickeners. In fact, the 5-stage circuit lost an estimated 43.6% of its input enthalpy (in streams V1 and V2) to evaporation and only about 2.5% to 2.6% each to radiation and convection from the tank walls.

The operating data for the CCD circuit in Example 2 indicated that a correlation can be established for heat loss from a single stage CCD thickener based on the temperature difference of the feed slurry and the ambient temperature. When temperatures are expressed in $^{\circ}$ C, then Figure 3 shows that the percentage heat loss (PHL)₁ is given by:

 $(PHL)_1 = 0.129 \ \Delta t \ + \ 1.823 \times 10^{-3} \ \Delta t^2 \ \dots \ (19)$

and the heat loss from N stages ($N \ge 2$) is then found from

 $(PHL)_N = (1.209)^N * (PHL)_1 \dots (20)$

When these equations were applied to Example 2, calculations of the pregnant liquor outlet temperature yielded 42.1°C against a measured value of 41.7° C, while the underflow temperature of 63.6° C is compared to a measured value of 65° C.

Conclusions

An iterative method is presented for computer calculation of CCD wash circuits when the wash solution contains soluble component(s) present in the feed slurry. A special case for a single soluble component and a clean wash solution results in a simplified expression.

Two examples of calculations using an in-house computer program show the application of the method to washing with dirty wash solutions at a wash ratio of 3.3:1 in an operating uranium plant and a wash ratio of 1.8:1 in a nickel-cobalt laterite pilot plant. In these examples the stage wash efficiencies were 75.5% and 98%, respectively, and the calculated losses of individual soluble component values is compared to measured values, as well as the total loss of soluble components.

An example of an operating uranium circuit with a clean wash solution was used to check the simplified expression in Equation (18). The results compared with the measured soluble loss at a stage efficiency of 88.6%.

When individual stage efficiency values are not known the method presented uses a single over-all value to represent the stage efficiency and gives reasonable good results and provides a method for calculations in the computer program.

A special case of hot or near-boiling feed was used to determine an empirical correlation to find the heat losses and do the CCD circuit heat balance for uncovered thickeners operating under this feed condition.

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2nd Symposium on Application of Mathematical Methods and Computers in Geology, Mining and Metallurgy

The Yugoslav Committee for Application of Mathematical Methods and Computers in Geology, Mining and Metallurgy is organizing the Second Symposium to take place in Belgrade, Yugoslavia, from October 21-25, 1991. The symposium will provide the opportunity for the exchange of information, including recent developments in research and technology and education in the fields of geology, mining and metallurgy.

The symposium will include all fields of the application of mathematical methods and computer techniques in the field. The main topics are:

 development and application of engineering methods based on mathematical approaches and computer techniques (CAD — technique, modelling, simulation, planning, design, graphics etc.);

operational researches (methods and appli-

cation);

- geostatistics;
- · information systems and data bases;
- · application of computer techniques;
- artificial intelligence and expert systems; and
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The official languages of the symposium are English and Russian, for both the presentation of papers and publication.

Authors interested in submitting papers should send three copies of short abstracts (up to 150 words) to the address given below no later than **January 15, 1991**. All accepted papers from registered delegates will be published in a proceedings volume which will be available to participants at the symposium. Papers not written in English will have extended English abstracts.

The registration fee is US\$250 which includes access to the symposium and technological exhibition, all volumes of the proceedings, coffee breaks and cocktails.

A 4-day field trip, from October 26 to 29, is being organized. Included are guided tours of the Kostolac open pit coal mine; Bor copper mine; Madjanpek open pit copper mine; Rajko's Cave; Gamzigrad's Spa; Grza karstic spring; Djerdap hydropower plant. The field trip will also include visits to the Serbian medieval monasteries, Manasija and Ravanica. The cost of the field trip is US\$350, and includes transportation, accommodation, meals, all tickets and a guide book.

For the submission of abstracts and further enquiries, contact: SITRGM — KPMR, II Symposium, Kneza Milosa 9/IV, Yugoslavia; Tel.: +38-11-334-357; Fax: +38-11-342-613.

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