# Metallurgical Testwork to Support Development of the Kintyre Project

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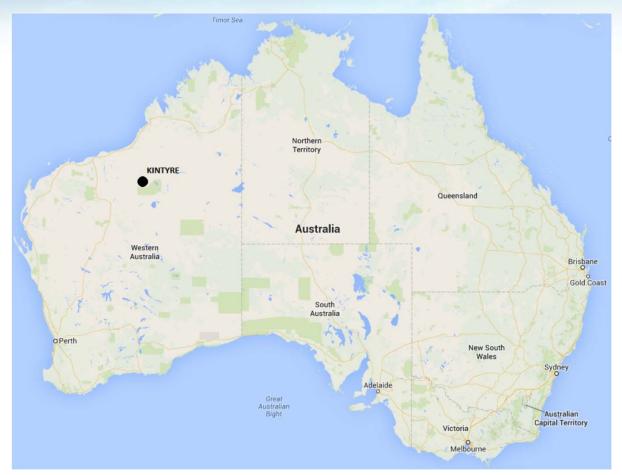


# **ANSTO Minerals**

- Studied processing of uranium ores for over 30 years;
- Extensive variety of deposits:
  - 6 operating mines in Australia (3 open, 3 closed)
  - 15 potential mines in Australia
  - 14 operating and potential mines outside Australia (mainly Southern Africa)
- Fundamental and diagnostic leach studies
- Detailed geometallurgical studies



# **KINTYRE URANIUM DEPOSIT**



- Discovered in 1985 by CRA (Rio Tinto)
- Acquired in 2008 by Cameco/Mitsubishi JV (70/30)



# **KINTYRE URANIUM DEPOSIT**

- 55 Mlb  $U_3O_8$  @ 0.58% average grade
- Uranium present mainly as:
  - Uraninite (UO<sub>2</sub>)
  - Coffinite  $((USiO_4)_{1-x}(OH)_{4x})$  lesser amounts
- High in carbonates
  - Ankerite (Ca(Fe,Mg,Mn)(CO<sub>3</sub>)<sub>2</sub>)
  - Dolomite  $(CaMg(CO_3)_2)$



# **Metallurgical Testwork**

# History

- Acid Leach Pilot Plant operated at ANSTO in 1997
  - Ore upgraded to 2%  $\rm U_3O_8$  by radiometric sorting and gravity separation
  - 7 campaigns
  - Direct uranium precipitation yielded on-spec product
- Alkaline Leaching subsequently investigated by Cameco, but acid route selected



# **Metallurgical Testwork**

# **Objectives of Work Discussed Today**

- Optimisation of leach conditions
  - Maximise U extraction
  - Minimise acid and oxidant consumption
- Evaluate leaching of variability samples
- Effect of ore type and leach conditions on settling, filtration and rheology
- Neutralisation and radionuclide deportment in tailings
- Solvent Extraction Pilot Plant
  - Evaluation of ammonia and strong acid stripping



## **Leach Testwork**



Automated – Computer controlled pH and ORP



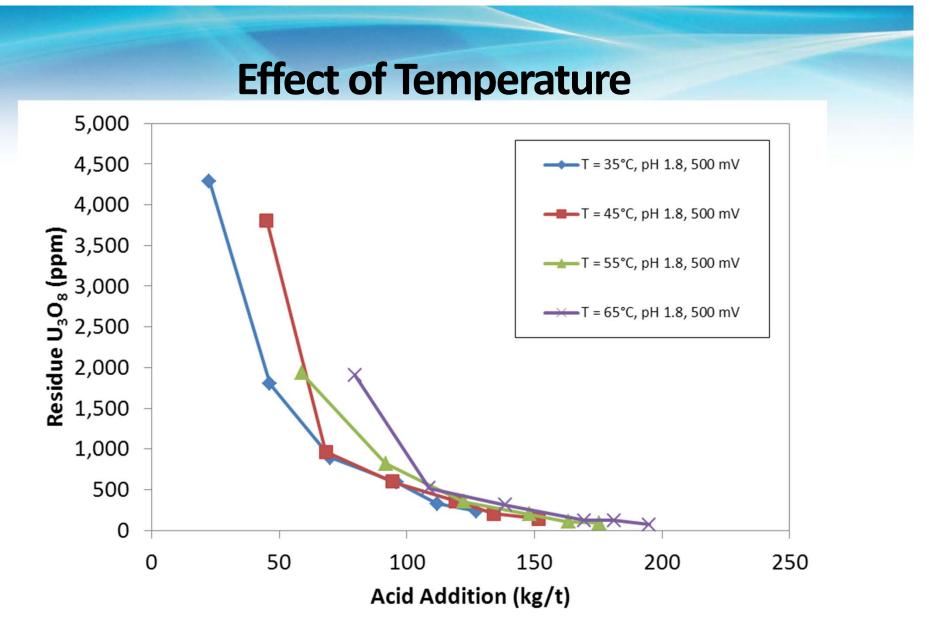
# **Leach Optimisation Program**

Composite sample prepared to represent average of U and CO<sub>3</sub> in orebody

Species	wt%	Species	wt%
CO <sub>3</sub>	9.55	Mg	7.3
Са	3.03	Si	25.6
Fe	8.2	U <sub>3</sub> O <sub>8</sub>	0.52

- Leach Testwork:
  - 50 wt% solids
  - 18 h leach time
  - NaMnO<sub>4</sub> as oxidant
- Variables investigated:
  - Temperature 35-65°C
  - pH 1.8-2.5
  - ORP 450-550 mV (vs. Ag/AgCl)
  - P<sub>80</sub> 250-710 μm

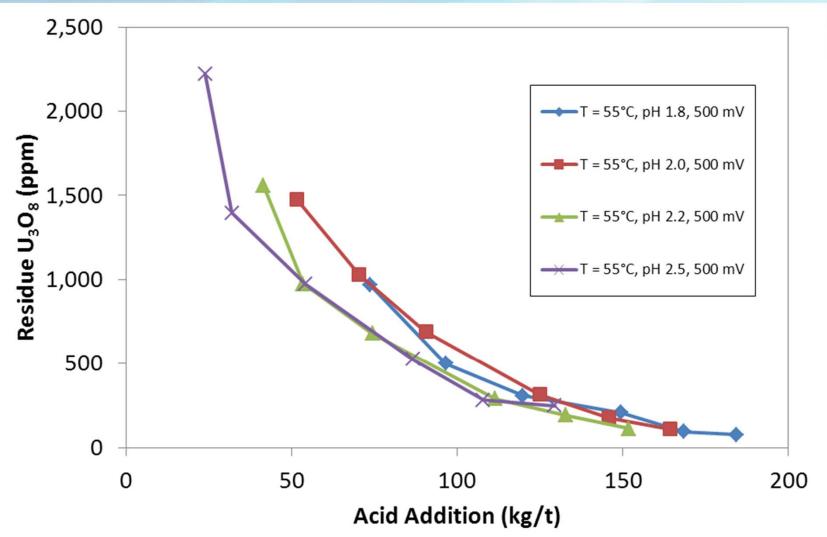




- U leached at varying rates, but final extractions similar
- 55°C selected as optimum temperature



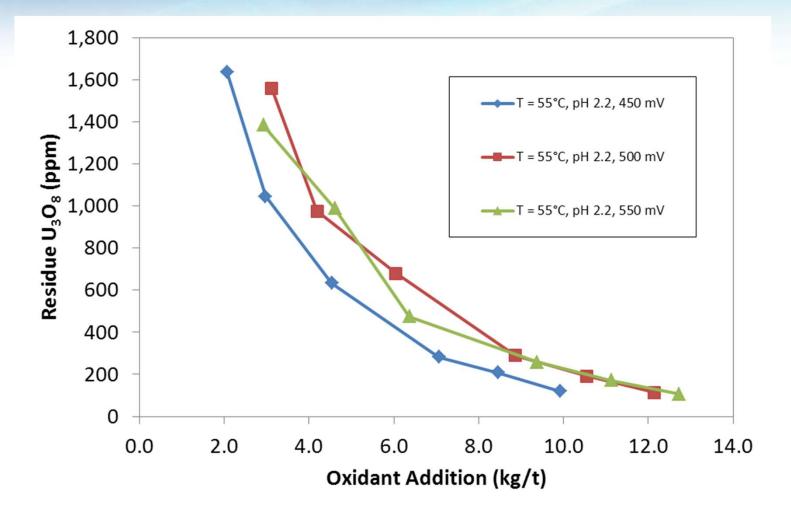
# Effect of pH



- Final U extractions similar at all pHs except 2.5
- pH 2.2 selected as optimum



## **Effect of ORP**



- 450 mV selected as optimum ORP
- U present mainly as Uraninite leaching not driven by ORP as long as Fe<sup>3+</sup> concentration is sufficiently high



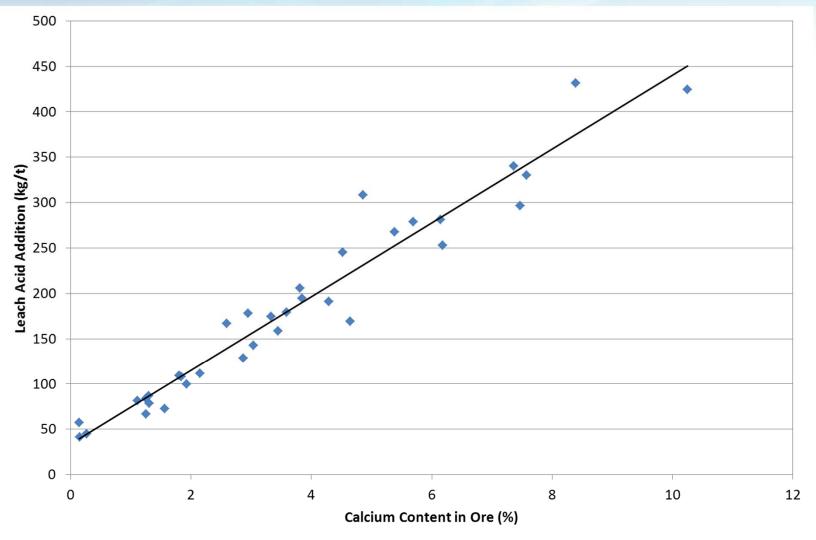
# **Variability Samples Leaching**

- Performed under optimum conditions determined:
  - pH 2.2
  - ORP 450 mV
  - $-P_{80} 500 \, \mu m$
  - 50 wt% Solids
  - 18 h Leach Time

	% U Extn.	Acid Addn. (kg/t)	Equiv. MnO2 Addn. (kg/t)
Max	99.2	432	22.3
Min	86.2	42	2.8
Avg.	95.5	187	11.1



# **Variability Samples Leach Results**





# **Other Work from Leaches**

- QEMScan on leach feeds and residues
- Effect of ore type and particle size on settling and filtration rate
- Effect of solids density on rheology for feed and residue slurries and neutralised slurries
- Tailings Neutralisation and radionuclide deportment



# **Bulk Leaching**

#### • 2000 kg of ore leached over three campaigns





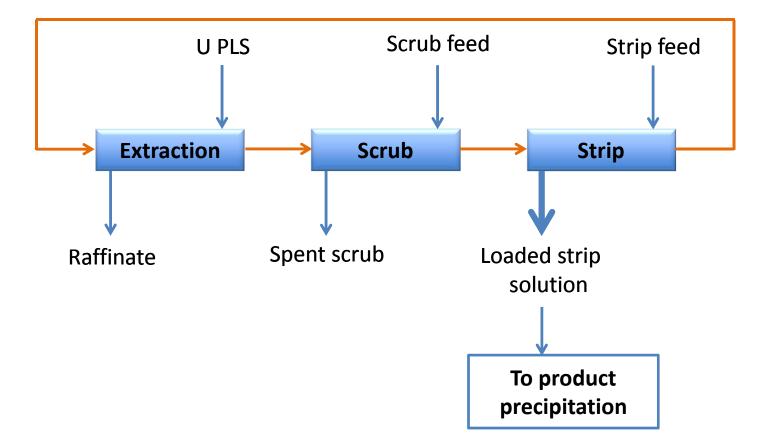
# **Solvent Extraction Testwork Objectives**

To carry out equilibrium batch testwork and to operate solvent extraction mini-pilot plant:

- To compare the performance of the ammonia and acid options for uranium stripping;
- To further define the solvent extraction process unit operation for the Kintyre Uranium Project to obtain engineering design data for a DFS.
- Confirm/optimise operating conditions;
- Identify potential operational issues;
- To trace the deportment of impurities.



### **Uranium Solvent Extraction – General Process**





# **Mini-Plant Continuous Operation**

- Two separate continuous solvent extraction mini plant trials operated for a total of 6 days;
- Total of ~1000 L of PLS was treated
- Alamine 336 (5 vol. %) + 2.5% iso-decanol in Shellsol 2046
- Stripping methods:
  - ✓ Ammonia stripping process  $(NH_4OH/(NH_4)_2SO_4)$ ,
  - ✓ Acid stripping process  $(H_2SO_4)$

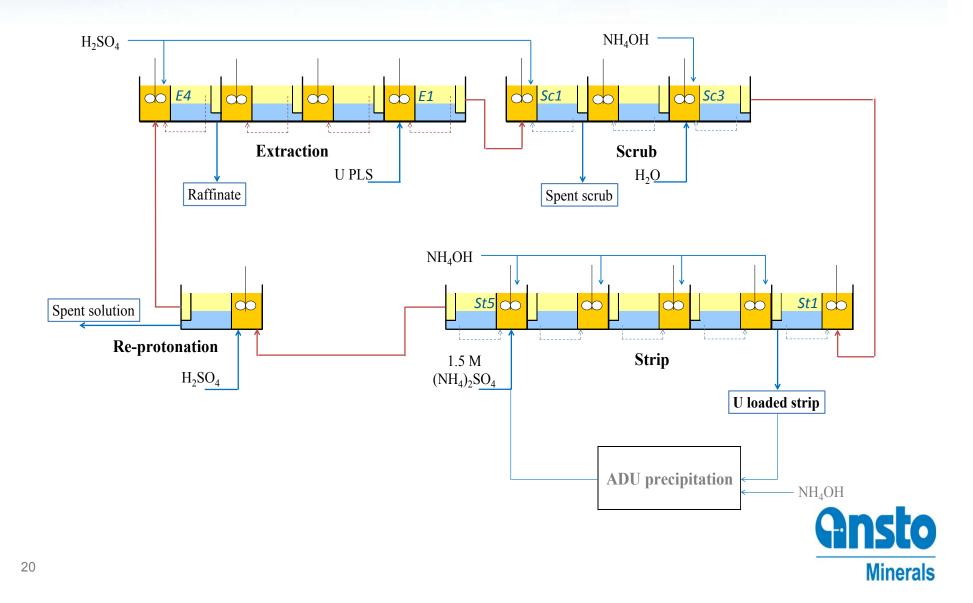


### **SX Mini-Plant Setup**

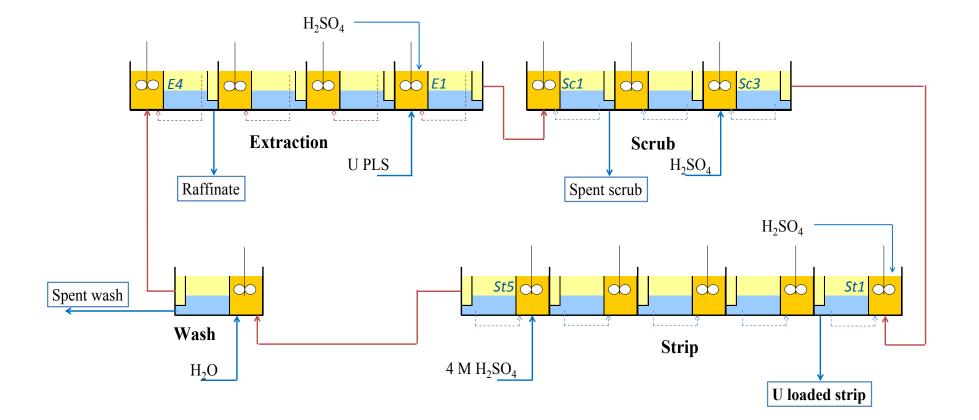




#### **Uranium Solvent Extraction - Ammonia Strip Process**



#### **Uranium Solvent Extraction - Acid Strip Process**





# **Feed Liquor**

	g/L	Elements	g/L	Elements	g/L
U	2.7	As	0.002	AI	1.6
S	53	Мо	<0.001	Са	0.57
		Si	0.3	Cu	0.02
рН	1.8	Zr	0.01	Mg	25
ORP (mV)	420	V	<0.001	Mn	16
				Na	1
		Fe	1.5	Ni	0.02
				Zn	0.03



# Extraction

#### Operating Conditions:

- pH 1.9-2.1 in E1 (loaded organic)
- 1.6-1.8 in E4 (raffinate)
- Temperature = 35-45°C
- O/A = 0.65

#### Performance:

- U extraction = 99.7-99.96 %
- [U]<sub>raffinate</sub> = 1-9 mg/L
- [U]<sub>loaded solvent</sub> = 4.2-4.3 g/L

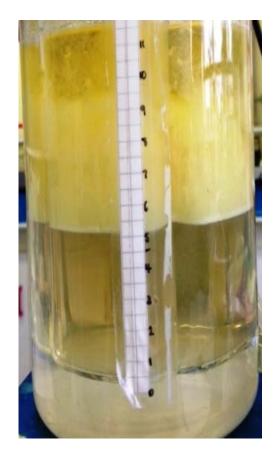


## **Phase Disengagement – Extraction**

#### Organic Continuous



#### Aqueous Continuous





### Scrub

#### **Operating Conditions:**

	рН		
Stago	Sc1	Sc3	
Stage	(spent scrub)	(scrubbed organic)	
Ammonia strip process	1.5	2.5	
Acid strip process	1.7	1.6	
Temperature 30-40°C		O/A ~10	

#### Performance:

- $\geq$  50 % of entrained Mg and Mn removal;
- Negligible scrubbing of As, Ca, Cu, Si and Zr;
- U scrubbing = 0.1-0.3 %



### **Operating Conditions**

#### Ammonia Strip:

- [(NH<sub>4</sub>)<sub>2</sub>SO<sub>4</sub>] = 1.5 M
- pH gradient:

pH 3 in St1 (product stream) pH 5 in St5 (stripped organic)

- **Strong Acid Strip:**
- H<sub>2</sub>SO<sub>4</sub> Concentration:

390 g/L in St1 (product stream)

400 g/L in St5 (stripped organic).

- Temperature = 30-40°C
- O/A = 5

- Temperature = 30-40°C
- O/A = 10



# **Mini-Plant Performance - Comparison**

#### Ammonia Strip:

	[M] <sub>product stream</sub>	[M] <sub>stripped solvent</sub>	Strip
Element	g/L	mg/L	%
U	18	30	99.3
S	54	-	

#### **Strong Acid Strip:**

	[M] <sub>product stream</sub>	[M] <sub>stripped solvent</sub>	Strip
Element	g/L	mg/L	%
U	41	15	99.6-99.8
S	126	3,700	



## **Mini-Plant Performance - Comparison**

	U Normalised Recovery (%)			
	To Raffinate	To Spent Scrub		
Ammonia Strip	0.3	0.1	99.6	
Strong Acid Strip	0.04-0.35	0.1-0.3		99.6-99.8



### **Impurities in the Product Stream**

	Ammonia strip	Acid strip	Max. U Concentrate Limit**
		% of U	
As	0.03	0.02	0.10
Са	0.16	0.29	1.0
Fe	0.04	0.01	1.0
Mg	0.07	< 0.002	0.50
Мо	< 0.03	< 0.01	0.30
Si	< 0.03	< 0.01	
V	< 0.03	< 0.01	0.30
Zr	0.06	0.15*	0.10

\* Zr rejected by  $H_2O_2$  precipitation

\*\* Without rejection (ASTM C967-13 Standard Specification for Uranium Ore Concentrate)



### Acid vs. Ammonia Stripping Methods

- Both methods are equally effective, achieving > 99 % stripping of U;
- The acid circuit is easier to control;
- Higher U concentration in the loaded acid strip solution;
- Acid route more options for uranium final product;
- Environmental issues associated with ammonia;
- Use of corrosive reagent.



## Conclusions

- Acid leaching has been identified as a robust process for Kintyre ores
- Testwork showed that optimum leach conditions for Kintyre are:
  - pH 2.2
  - ORP 450 mV
  - $-P_{80} \, 500 \, \mu m$
  - 18 h Residence time
- Average Uranium Extraction of 95.5% for Variability <sub>31</sub> Samples



### Conclusions

- Two fully integrated U SX mini-pilot plants were successfully operated for 6 days testing ammonium strip and acid strip processes;
- High U recoveries from feed to product stream were achieved:
  - [U]raffinate < 10 mg/L;</li>
  - acid strip process: 99.8 %;
  - o ammonia strip process: 99.6 %.
- Comparable stripping efficiency:
  - ammonia strip process: 99.3 %;
  - acid strip process: 99.6-99.8 %.
- Potential operational issue with ammonia strip stable emulsion formation (extraction) and crud.

