

**Supplementary Information**

**Remaking the chlor-alkali electrolysis process to co-generate useful reduction products from CO<sub>2</sub>**

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## 1. Determination of uncompensated resistance

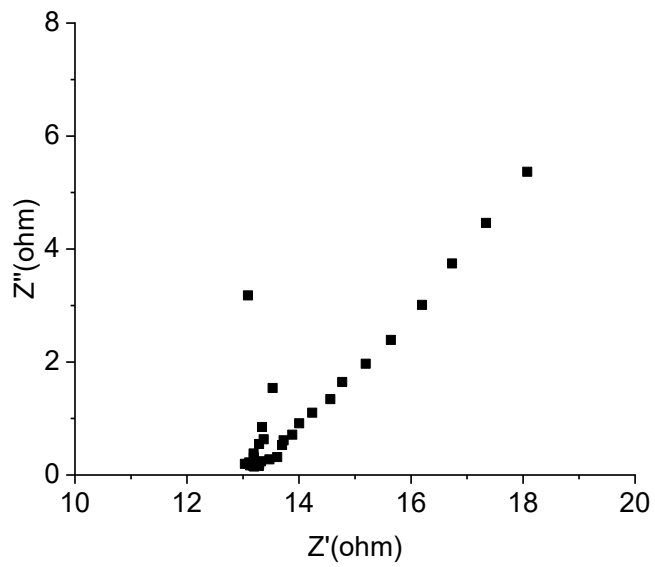


Fig. S1: Potentiostatic electrochemical impedance spectroscopy of the overall cell has resistance of 13  $\Omega$ . The electrochemical cell with the cathode, anode, membrane and electrolyte is described in the experimental section.

## 2. Linear Sweep Voltammetry

Linear sweep voltammetry was conducted to compare the performance of Ni<sub>2</sub>P in electrolyte comprised of CO<sub>2</sub>-saturated 0.5 M KHCO<sub>3</sub> (red lines) and in Ar-saturated 0.25 M Na<sub>4</sub>P<sub>2</sub>O<sub>7</sub> (black lines) at the same starting pH, as shown in **Fig. S2**. At more negative potentials, the former gives lower current density, indicating that CO<sub>2</sub> binding to the cathode blocks the HER active sites. However, at applied potentials more positive than -0.2V, the red trace shows higher current density, indicating the relatively higher activity of CO<sub>2</sub>RR over HER. This behavior is consistent with thermodynamic control of the product distribution, as we previously reported.

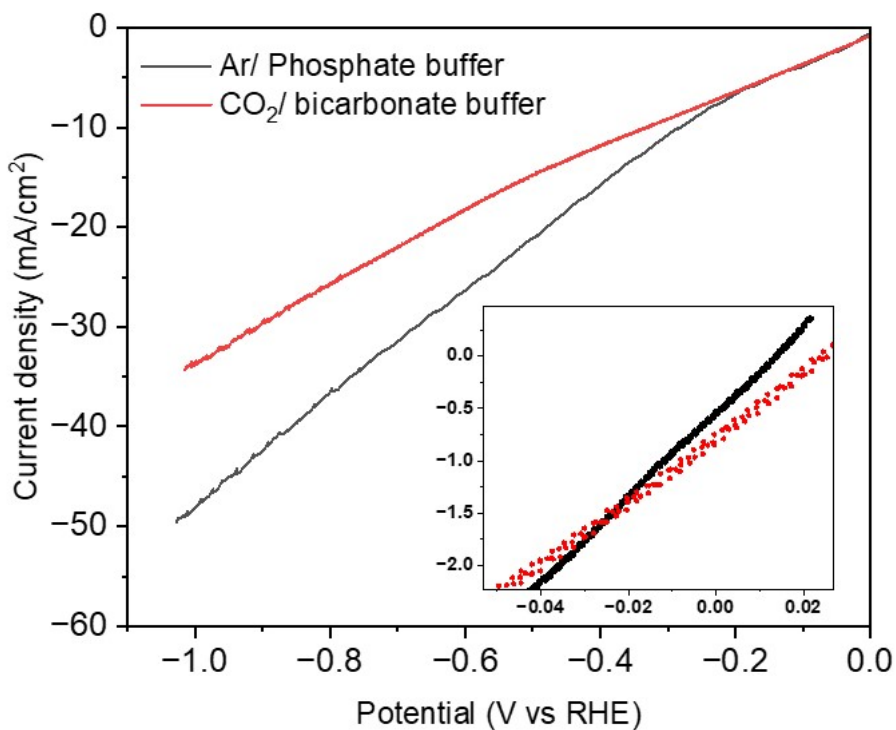


Fig. S2: Linear sweep voltammetry (LSV) for Ni<sub>2</sub>P in CO<sub>2</sub>-saturated 0.5 M KHCO<sub>3</sub> (red lines) and in Ar-saturated 0.25 M Na<sub>4</sub>P<sub>2</sub>O<sub>7</sub> (black lines) at 10mV/s in cathodic direction.

### 3. Gas Chromatography

The volatile gas products were analyzed by Gas Chromatograph (GC; SRI model 8610C) with flame ionization detector (FID) and thermal conductivity detector (TCD) in series. Here, the headspace of the cell was directly connected to the GC for online analysis. Volatile gases were separated using a molecular sieve 5A porous layer open tubular capillary column (RT-Msieve 5A) prior to the detectors. The apparatus was purged before reaction to check for air leakage, and then every 1 hour during the measurements.

Samples were taken before reaction to check for air presence and then every hour thereafter. The hydrogen calibration was done with *in situ* generated gas through electrolysis of water on platinum, under argon purge, and diluted post-reaction with CO<sub>2</sub>.

The oxygen calibration was done with *in situ* generated gas through electrolysis of water on platinum.

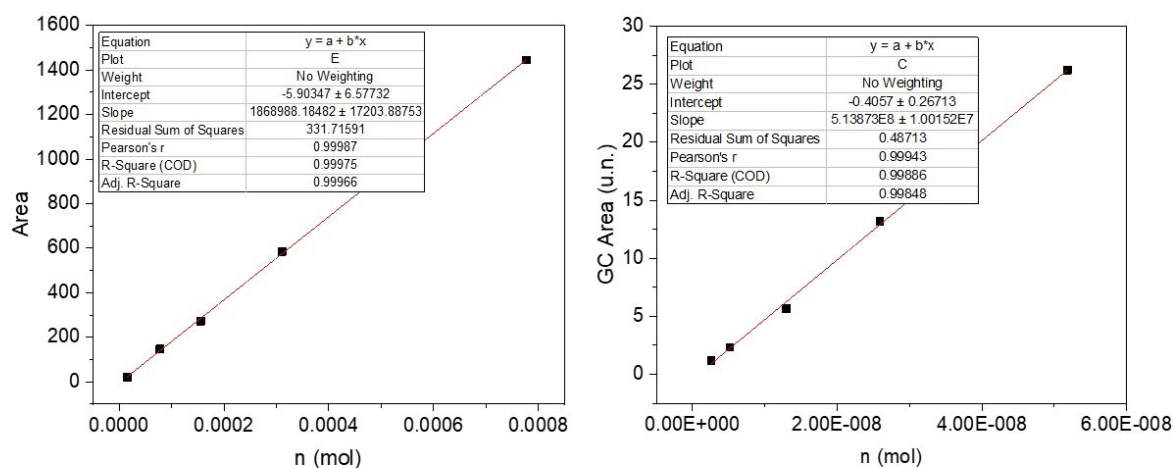


Fig. S3: (a) Calibration curve for hydrogen, the only product detected in the gas phase at the cathode side; (b) Calibration curve for oxygen, the only detected gas product apart from chlorine in the gas phase at the anode side.

#### 4. Liquid Product Analysis by $^1\text{H}$ NMR

The  $^1\text{H}$  NMR sample was made by mixing the 450  $\mu\text{L}$  catholyte taken after the reaction from the working electrode compartment with 150  $\mu\text{L}$  of  $\text{D}_2\text{O}$ .  $^1\text{H}$  NMR spectra was recorded using a Bruker Avance Neo 500 MHz NMR spectrometer. For the specific details of the assignment of the NMR peaks, please refer to the literature<sup>1</sup>.

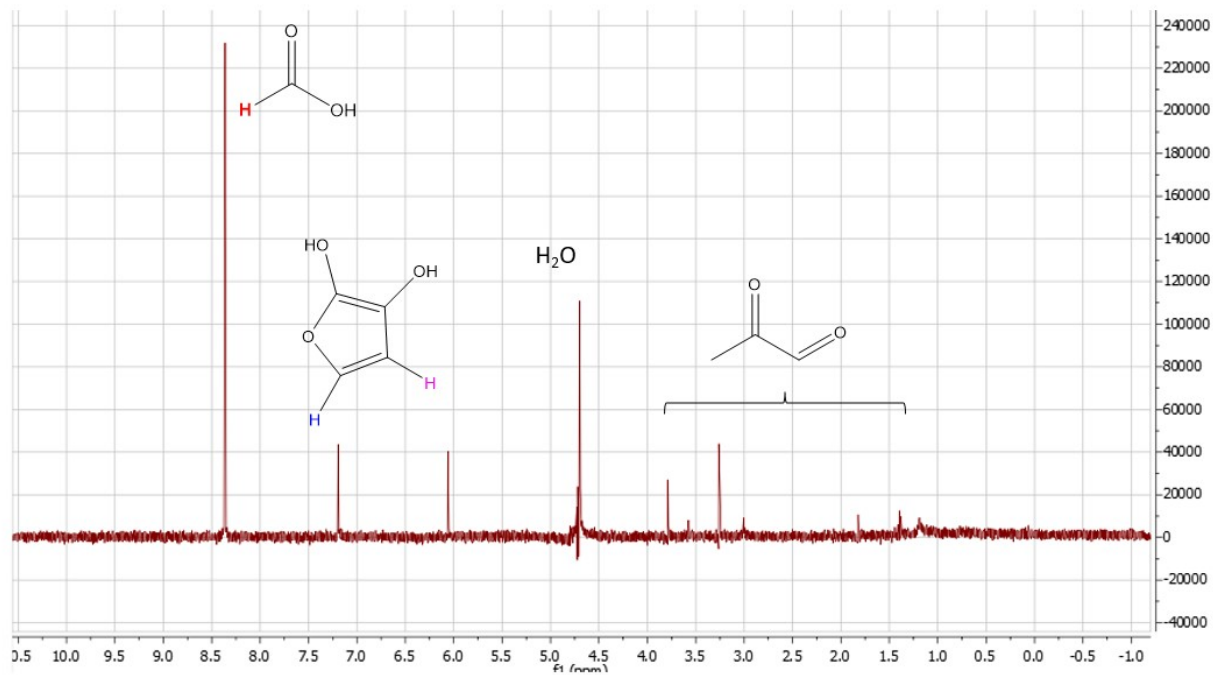


Fig. S4:  $^1\text{H}$  NMR of a catholyte sample from the electroreduction of  $\text{CO}_2$  on  $\text{Ni}_2\text{P}$ (3mA).

## 5. High Performance Liquid Chromatography

The liquid products were detected and quantified by an offline high-performance liquid chromatography (HPLC; Perkin Elmers Flexer) with an autosampler, UV-vis detector and refractive index detector (RID). Separation was performed on a polymer-based matrix (polystyrene divinylbenzene) column (BioRad HPX 87H Aminex), with injection volumes of 20  $\mu\text{L}$ . The runtime was 60 minutes at a flow rate of 0.3  $\text{mL min}^{-1}$  and 45  $^{\circ}\text{C}$ . The calibration ( $R^2 > 0.99$ ) for furandiol, HCOOH and methyl glyoxal was conducted with standards of concentrations ranging from 0.1 to 50 mM. Product assignment was established by retention time and confirmed by  $^1\text{H}$  NMR (chemical shifts and splitting multiplicity), as described in detail in the ESI and in Calvinho et al.<sup>1</sup>. All measurements were repeated three times on identically prepared and independent samples for statistical significance.

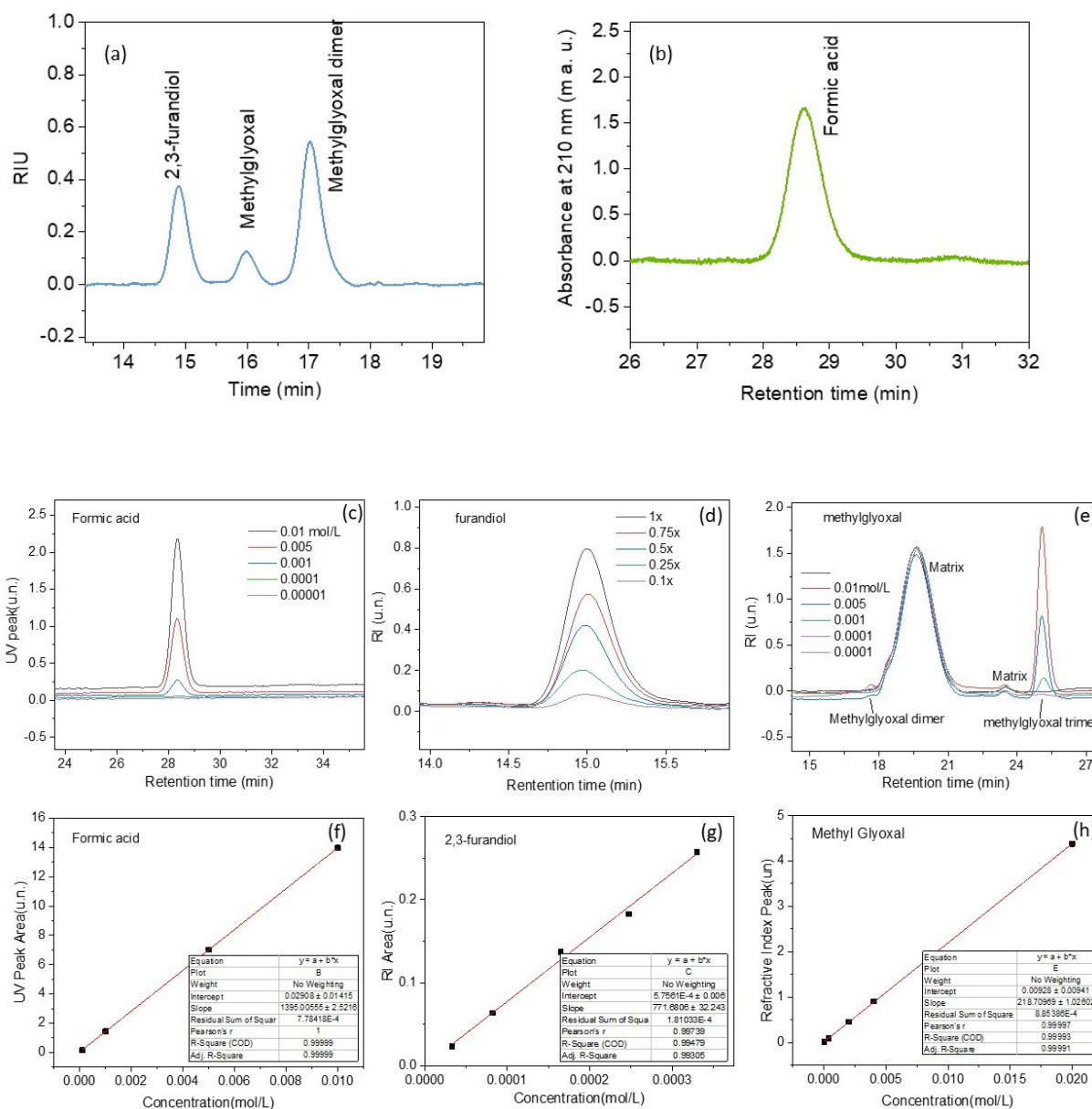


Fig. S5: (a) and (b) Chromatographs for a catholyte sample for CO<sub>2</sub>RR on Ni<sub>2</sub>P at the current of 1mA obtained with the refractive index detector (left) and the UV absorbance detector (right); (c-h) The HPLC peak for standards of formic acid, 2,3-durandiol and methyl glyoxal and its calibration curves, respectively. The mean error associated with the HPLC quantification was determined to be <1%. The calibration method for methylglyoxal and 2,3-furandiol are the same as the published literature<sup>1</sup>.

## 6. Faradaic Efficiency

The Faradaic Efficiency for the liquid CO<sub>2</sub> reduction products, hydrogen and oxygen was calculated using the equations below, respectively:

$$FE_{CO_2RR} = \text{Concentration} \cdot V_{\text{electrolyte}} \cdot \# \text{electrons} \cdot F / \text{Charge}$$

$$FE_{H_2} = \text{mols of } H_2 \cdot \# \text{electrons} \cdot F / (\text{current} \cdot \text{time})$$

$$FE_{O_2} = \text{mols of } O_2 \cdot \# \text{electrons} \cdot F / (\text{current} \cdot \text{time})$$

In order to test the remaining charge of oxidation process, a one-time experiment was conducted: argon gas at the flow rate of 20 sccm was purged into the system, went through the 40% NaOH solution and finally connected to the GC to test the oxygen amount, while the chlorine was dissolved in the NaOH solution. At the current of 3 mA, it was calculated that FE(O<sub>2</sub>) to be 28%, matches with the FE(Cl<sub>2</sub>) of 70%.

Table S1: Faradaic efficiency for all catalysts at the potentials tested  $\pm$  standard deviation between at least three chronoamperometry experiments.

Current (mA)	Formate Faradic Efficiency (%)	2,3-furandiol Faradic Efficiency (%)	Methylglyoxal Faradic Efficiency (%)	CO <sub>2</sub> RR FE (%)	Hydrogen Faradic Efficiency (%)	Total FE (%)
-1	3.15 $\pm$ 0.49	20.25 $\pm$ 0.77	77.45 $\pm$ 0.22	100.85	0	100.85
-2	2.62 $\pm$ 0.14	17.36 $\pm$ 1.34	74.6 $\pm$ 0.52	94.58	5.8 $\pm$ 4.1	100.38
-3	3.69 $\pm$ 0.1	17.21 $\pm$ 0.94	56.67 $\pm$ 3.7	77.51	25 $\pm$ 7	102.51



## 7. XRD

The diffraction peaks at 17.4, 26.3, 30.5, 31.7, 35.4, 40.7, 44.6, 47.4, 54.2, 54.9, 57.4, 61.2, 63.5, 66.4, 69.6, 72.7, 74.8, 80.2, 82.9, 86.2, 88.2 and 88.8 are assigned to the (100), (001), (110), (101), (200), (111), (201), (210), (300), (211), (102), (301), (220), (310), (221), (311), (400), (302), (401), (320), (003), (410) and (321) crystal phases, respectively.

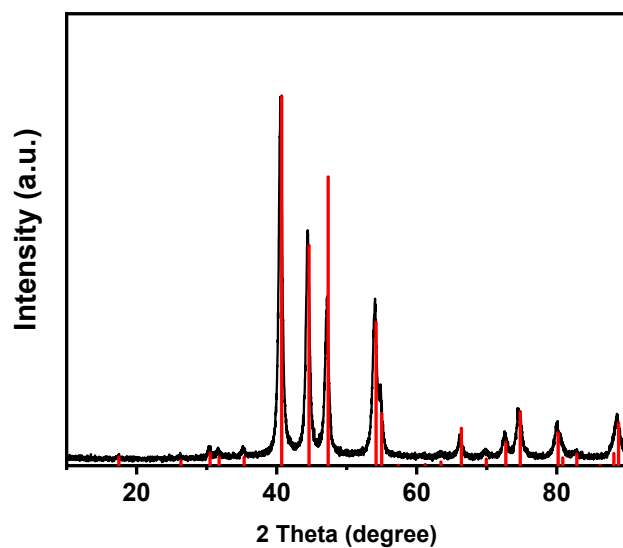


Fig S6. XRD pattern of the Ni<sub>2</sub>P synthesized by soft-templated method (black) and reference peak: ICDD # 01-074-1385 (red).

## 8. XPS

By comparing catalysts without acid washing, we found that Cl still exists as shown in below **Fig. S7a**, and the atomic percentage is  $1.34 \pm 0.47 \%$ , similar to the  $\text{Ni}_2\text{P}$  catalyst without acid washing ( $1.49 \pm 0.65 \%$ ). Therefore, it confirms that the Cl is from the nickel chloride of the synthesis procedure. By further washing the catalyst repeated for 10 times with water, the Cl remains as shown in **Fig. S7b**, indicating the Cl bind with nickel phosphides tightly through chemical binding.

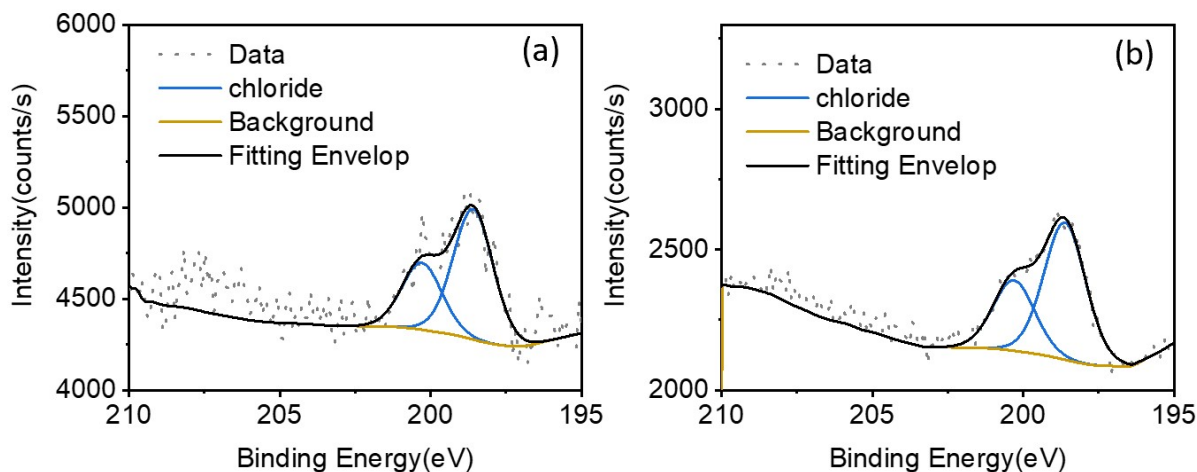


Fig. S7 XPS spectra of Cl 2p of  $\text{Ni}_2\text{P}$  catalyst without acid wash(a);  $\text{Ni}_2\text{P}$  catalyst without acid wash which is further washed with  $\text{H}_2\text{O}$  for 10 times (b).

Table S2. The calculated atomic percentage through XPS analysis was listed for pristine electrodes (sample 1-3); electrodes after catalytic turnover (sample 4-6); pristine electrodes without acid wash (Sample 7-8).

	Pristine electrode (with acid wash)			Electrode after catalytic turnover (with acid wash)			Pristine electrode (without acid wash)	
	Sample 1	Sample 2	Sample 3	Sample 4	Sample 5	Sample 6	Sample 7	Sample 8
Ni	12.42	8.72	10.63	6.94	11.3	14.86	11.22	10.04
O	45.11	42.61	45.67	33.32	41.87	45.82	47.45	45.2
N	1.34	0.7	1.2	8.04	3.96	3.74	1.36	1.15
C	25.54	35.88	28.67	47.01	40.91	32.61	29.8	33.17
Cl	0.63	1.64	2.19	2.84	0.42	0.95	1.8	0.87
P	14.97	10.45	11.64	1.85	1.54	2.01	8.37	9.57

## 9. Techno-Economic Feasibility

An initial techno-economic feasibility (TEF) model was developed to assess the economic impact of co-generation of methylglyoxal (MG) and chlorine ( $\text{Cl}_2$ ). A TEF model provides a total cost and profit assessment of a plant in the units of \$ per day. The costs are categorized into capital costs and operating costs. In this preliminary TEF study we used polarization data obtained at NREL for a single stack MEA type electrolyzer and the same nickel phosphide catalyst supported in a gas diffusion electrode<sup>2</sup>. The capital costs consider only the cost of the electrolyzer and therefore do not include the costs of infrastructure or other downstream or upstream equipment costs. The operating costs include the electricity cost, estimated maintenance cost, and the feedstock costs. The operational lifetime of the plant is assumed to be over 20 years, which is generally considered a conservative estimation for electrochemical processes<sup>3, 4</sup>. The profits are calculated based on the production of all products including MG,  $\text{Cl}_2$ ,  $\text{KHCO}_3$  and  $\text{H}_2$ .

We start by calculating the electricity and electrolyzer capital for a plant producing 50,000 kg MG/day. Cell performance upper limit estimate is as follows: cell operating potential of 3.19 V at 50 mA/cm<sup>2</sup> total current density with a 50% FE of MG which will be used as the basis in the following calculations<sup>2</sup>. The cell potential for combined  $\text{CO}_2\text{RR}$  and  $\text{Cl}_2$  evolution is estimated starting from the  $\text{CO}_2\text{RR}$  electrolysis using the identical confirmation as follows: 3.06V overall cell potential to operate at 50mA/cm<sup>2</sup> for  $\text{CO}_2\text{RR}$  electrolysis<sup>2</sup>, to this is added the additional potential required to produce  $\text{Cl}_2$  rather than  $\text{O}_2$  at the anode. This added potential is estimated as the standard potential difference between chloride oxidation to  $\text{Cl}_2$  (1.36V vs NHE) and water oxidation to  $\text{O}_2$  (1.23V)<sup>5</sup>. This calculation implicitly assumes that the overpotential required for chlorine evolution reaction (CER) and oxygen evolution reaction (OER) are similar, together this

gives a cell voltage of 3.19V. Considering CER's faster kinetics (2 electron reaction) compared with OER (4 electron reaction), this is considered a conservative assumption. It should be noted that a far more efficient water electrolysis configuration has recently been demonstrated<sup>6</sup> and this cell type may be applicable to CO<sub>2</sub>RR in the future with corresponding decreased operating costs.

As outlined in the main text, the anode current density is actually doubled under the optimized condition to reach up to 100mA/cm<sup>2</sup> due to the smaller anode size. The electrode area used for the electrolyzer capital cost calculations are based on the cathode for simplicity assuming the anode will simply be reduced in size while the rest of the cell structure remains the same.

To estimate the electrolyzer cost we use the basis of the capital costs reported by the hydrogen model (H2A) of the United States Department of Energy for the production of hydrogen via water electrolysis, we make a first level TEF with the cost of the electrolyzer stack is \$423/kW.<sup>7</sup> This model has been widely used in previous CO<sub>2</sub> electroreduction techno-economic analysis<sup>4, 8, 9</sup>. Due to the improvements in water electrolysis costs recent studies estimate the cost of CO<sub>2</sub>RR electrolyzers at \$300/kW, making the above a conservative estimate<sup>4</sup>. The parameters used for the TEF calculation are listed in the below Table S3.

**Table S3.** Parameters for the combined CO<sub>2</sub>RR and CER system

Parameter	Value
Overall cell voltage, V	3.19
CO <sub>2</sub> electrolysis current density, mA/cm <sup>2</sup>	50
Methylglyoxal faradic efficiency, %	50
Cl <sub>2</sub> faradic efficiency, %	100

Electrolyzer cost, \$/kW	423
Electricity cost, ¢/kWh	5.8
Daily methylglyoxal production, kg/day	50,000

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### (1) Mass and Energy Flow

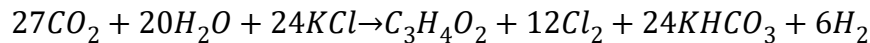
The daily mass of several products (Cl<sub>2</sub>, KHCO<sub>3</sub>, H<sub>2</sub>) is calculated by the followed equations:

$$\text{Daily product mass} \left( \frac{kg}{day} \right) = \frac{\text{daily methylglyoxal production} \left( \frac{kg}{day} \right) * \text{product molecular weight} \left( \frac{g}{mol} \right) *}{\text{methylglyoxal molecular weight} \left( \frac{g}{mol} \right)}$$

The daily reactant mass (CO<sub>2</sub>, H<sub>2</sub>O, KCl) consumed is calculated by the equation below:

$$\text{Daily reactant mass} \left( \frac{kg}{day} \right) = \frac{\text{daily methylglyoxal production} \left( \frac{kg}{day} \right) * \text{reactant molecular weight} \left( \frac{g}{mol} \right) *}{\text{methylglyoxal molecular weight} \left( \frac{g}{mol} \right)}$$

where the product to methylglyoxal ratio/ reactant to methylglyoxal ratio is obtained from the followed equation (considering the production of methylglyoxal and H<sub>2</sub> at the cathode side, each with 50% faradic efficiency):



The KCl usage should be 24 eq of methylglyoxal production due to the according to the equation above (the molecular weight of methylglyoxal is 72g/mol, the molecular weight of KCl is 74.5g/mol):

$$\text{mass (KCl)} = 50000 \frac{\text{kg}}{\text{day}} \times \frac{\text{mol}}{72\text{g}} \times 24 \times 74.5 \frac{\text{g}}{\text{mol}} = 1.24 \times 10^6 \frac{\text{kg}}{\text{day}}$$

The CO<sub>2</sub> usage for methylglyoxal:

$$\text{mass (CO}_2, \text{MG)} = 50000 \frac{\text{kg}}{\text{day}} \times \frac{\text{mol}}{72\text{g}} \times 3 \times 44 \frac{\text{g}}{\text{mol}} = 9.17 \times 10^4 \frac{\text{kg}}{\text{day}}$$

The CO<sub>2</sub> usage for KHCO<sub>3</sub>:

$$\text{mass (CO}_2, \text{KHCO}_3) = 50000 \frac{\text{kg}}{\text{day}} \times \frac{\text{mol}}{72\text{g}} \times 24 \times 44 \frac{\text{g}}{\text{mol}} = 7.33 \times 10^5 \frac{\text{kg}}{\text{day}}$$

The production of H<sub>2</sub>:

$$\text{mass (H}_2) = 50000 \frac{\text{kg}}{\text{day}} \times \frac{\text{mol}}{72\text{g}} \times 6 \times 2 \frac{\text{g}}{\text{mol}} = 8.33 \times 10^3 \frac{\text{kg}}{\text{day}}$$

The usage of H<sub>2</sub>O:

$$\text{mass (H}_2\text{O)} = 50000 \frac{\text{kg}}{\text{day}} \times \frac{\text{mol}}{72\text{g}} \times 20 \times 18 \frac{\text{g}}{\text{mol}} = 2.5 \times 10^5 \frac{\text{kg}}{\text{day}}$$

We neglect the charge loss of OER considering the high FE of CER<sup>10</sup>, the production rate of Cl<sub>2</sub> should be:

$$\text{mass (Cl}_2) = 50000 \frac{\text{kg}}{\text{day}} \times \frac{\text{mol}}{72\text{g}} \times 71 \frac{\text{g}}{\text{mol}} \times 12 = 5.92 \times 10^5 \frac{\text{kg}}{\text{day}}$$

The KHCO<sub>3</sub> product should be:

$$\text{mass (KHCO}_3) = 50000 \frac{\text{kg}}{\text{day}} \times \frac{\text{mol}}{72\text{g}} \times 100 \frac{\text{g}}{\text{mol}} \times 24 = 1.67 \times 10^6 \frac{\text{kg}}{\text{day}}$$

The separation cost for KHCO<sub>3</sub> could possibly be reduced by utilizing the heat of the *iR* drop from the electrolyzer, however modelling this cost is outside the defined scope of this initial TEF.

## (2) Capital cost

The current needed for one day of operation for this 12-electron reduction reaction is:

$$\begin{aligned}
 \text{Current}(A) &= \frac{\text{methylglyoxal production}(g) * 12e \frac{\text{mol}}{\text{molCO}_2} * 96500C/m}{\text{methylglyoxal molecular weight} \left( \frac{g}{\text{mol}} \right) * \text{faradic efficiency} * 3600} \\
 &= 50000 \frac{kg}{day} \times 1000 \frac{g}{kg} \times \frac{day}{3600 \times 24s} \times \frac{mol}{72g} \times 12e \frac{mol}{molCO_2} \times 96500 \\
 &= 1.86 \times 10^7 A
 \end{aligned}$$

The power consumed in a day can be calculated from an estimated operating cell potential of 3.19 V for 50 mA/cm<sup>2</sup> overall current density,

$$\text{Energy consumed (kW)} = \text{Current}(A) * \text{Potential}(V) * (\text{kW}/1000W)$$

$$P = 1.86 \times 10^7 A \times 3.19 V \times \frac{kW}{1000W} = 5.93 \times 10^4 kW$$

As stated above we choose to use a conservative estimate for electrolyzer costs from B. James et al. developed for water splitting in this first level TEF<sup>11</sup>. The cost of the electrolyzer stack is therefore taken as \$423/kW. The electrolyzer is assumed operated at 0.05 A/cm<sup>2</sup> and 3.19V. The installation factor is 1.2 based on the value from B. James et al. Thus, the cost of an electrolyzer capable of producing 50 metric tons/day of MG is:

*Electrolyzer Capital cost*

$$= \text{Energy consumed}(kW) * \text{electrolyzer cost} \left( \frac{\$}{kW} \right) * \text{installation factor}$$

$$\text{Electrolyzer Capital cost} = 5.93 \times 10^4 kW \times \frac{\$423}{kW} \times 1.2 = 3.01 \times 10^7 \$$$

While water splitting and CO<sub>2</sub>RR requires different balance of plant (BoP) technology, the detailed evaluation of this is outside the scope of this initial TEF and the BoP capital cost is considered 10% of the total cost<sup>11</sup>, giving a BoP capital cost of

$$BoP = \$1.78 \times 10^7 \times \frac{0.1}{0.9} = \$1.98 \times 10^6$$

Here we used a mature ethanol purification technology to estimate the downstream purification capital cost, which includes purification (distillation, rectifier column, stripping column, molecular sieve, heat exchanger, product tank, and pump) and recycle streams, based on similar mature technology, has a scaled capital cost of \$67226<sup>12</sup>.

### (3) Operating Costs and Profits

The current electricity cost is calculated from the power requirement and the price of electricity (¢5.8 /kWh based on the average electricity costs in cheap US states like Arkansas, Kentucky, Louisiana, Montana, Oklahoma, Texas, and Washington)<sup>13</sup>:

$$Electricity\ cost\left(\frac{\$}{day}\right) = \frac{Energy\ consumed(kW)}{3600s} * 3600 * \frac{24s}{day} * Electricity\ price\left(\frac{\$}{kWh}\right)$$

$$Electricity\ cost = 5.93 \times 10^4 kW \times \frac{24s}{day} \times \frac{kWh}{3600 \times 1000W} \times \frac{\$0.058}{kWh} = \$8.24 \times 10^4/day$$

The maintenance cost is assumed 3% of capital cost per year based on H2A<sup>7</sup>:

$$\$3.208 \times 10^7 \times \frac{0.025}{year} \times \frac{year}{365day} = \$2,636/day$$

The cost of CO<sub>2</sub> is calculated based on a cost of \$130 from<sup>14</sup>:

$$CO_2\ cost = 8.25 \times 10^5 kg/day \times \frac{\$130}{ton} \times \frac{ton}{1000kg} = \$1.07 \times 10^5/day$$



The cost of H<sub>2</sub>O consumption is based on costs of \$1.6/ton from<sup>15</sup>:

$$H_2O \text{ cost} = 2.5 \times 10^5 \text{kg/day} \times \frac{\$1.6}{\text{ton}} \times \frac{\text{ton}}{1000\text{kg}} = \$4.0 \times 10^2/\text{day}$$

The \$202/ton cost of KCl is based on 2021 numbers due to the abnormal higher costs of KCl in 2023 which are expected to return to normal levels<sup>16</sup>:

$$KCl \text{ cost} = 1.24 \times 10^6 \text{kg/day} \times \frac{\$202}{\text{ton}} \times \frac{\text{ton}}{1000\text{kg}} = \$2.50 \times 10^5/\text{day}$$

The \$122/ton price of Cl<sub>2</sub> is based on <sup>17</sup> and yields a daily cost of:

$$Cl_2 \text{ cost} = 5.92 \times 10^5 \frac{\text{kg}}{\text{day}} \times \frac{\$122}{\text{ton}} \times \frac{\text{ton}}{1000\text{kg}} = \$7.22 \times \frac{10^4}{\text{day}}$$

Using the wholesale price of methylglyoxal as 1.9 \$/kg<sup>18</sup>, the daily price of methylglyoxal from this process is:

$$MG \text{ revenue} = 5 \times 10^4 \text{kg/day} \times \frac{\$1.9}{\text{kg}} = \$9.5 \times 10^4/\text{day}$$

The price of H<sub>2</sub> is estimated from<sup>19</sup>:

$$H_2 \text{ revenue} = 8.33 \times 10^3 \text{kg/day} \times \frac{\$1350}{\text{ton}} \times \frac{\text{ton}}{1000\text{kg}} = \$1.12 \times 10^4/\text{day}$$

The \$ 1,089/ton price of KHCO<sub>3</sub> is based on <sup>20</sup>:

$$KHCO_3 \text{ cost} = 1.67 \times 10^6 \text{kg/day} \times \frac{\$1089}{\text{ton}} \times \frac{\text{ton}}{1000\text{kg}} = \$1.82 \times 10^6/\text{day}$$

According to the handbook of chlor-alkali technology, the cost of chlorine purification/processing is \$23,400/day when scaled for a 592 ton/day production.<sup>21</sup>

The scaled operating cost for purification technology of methylglyoxal through distillation column is \$2669/day. <sup>12</sup>

Thus, the net per day profit is:

$$\begin{aligned} & \$ \\ & (7.22 \times 10^4 + 9.5 \times 10^4 + 1.82 \times 10^6 + 1.12 \times 10^4 - 2.5 \times 10^5 - 1.0 \times 10^5) \\ & /day = \$1.53 \times 10^6/day \end{aligned}$$

This initial TEF therefore indicated that the proposed integrated system for co-product generation is competitive in the market at current costs of electricity and feedstocks. Future developments may further improve the commercial impact of this production.

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