Supporting Information

Machine-learned Charge State Prediction Enables Nitrogen Reduction Reaction on Transition Metals Doped C₃B

Chengwei Yang^{a,b,†}, Chao Yang^{c,†}, Yunxia Liang^a, Hongxia Yan^a, Aodi Zhang^d, Guixian Ge^{a,*}, Wentao Wang^{b,c,*}, Pengfei Ou^{c,*}

^a Key Laboratory of Ecophysics and Department of Physics, College of Science, Shihezi University, Shihezi 832003, PR China

^b Guizhou Provincial Key Laboratory of Computational Nano-Material Science, Guizhou Education University, Guiyang 550018, PR China.

^c Department of Chemistry, National University of Singapore, Singapore, 117543, Singapore

^d School of Physics and Electronics, Henan University, Kaifeng 475004, China

[†]These authors contributed equally.

* Corresponding authors: pengf.ou@nus.edu.sg (P. Ou); wtwang@gznc.edu.cn (W. Wang); geguixian@126.com (G. Ge)

Note S1. The formation energy of defect

The defect formation energy $E_f(D^q)$ in 2D C₃B for a defect *D* in a charge state *q* is evaluated as ¹:

$$E_{f}(D^{q}) = E_{tot}(D^{q}) - E_{tot}(bulk) + \sum_{i} n_{i}\mu_{i} + q(E_{F} + E_{VBM})$$
(S1)

- $E_{tot}(D^q)$ and $E_{tot}(bulk)$ are the calculated total energies of the supercells with defect X in charge state q and without the defect, respectively.
- n_i is the number of atoms of type *i* (n_i<0, if atoms are added; and n_i>0, if atoms are removed).
- µ_i is the corresponding atomic chemical potential. The chemical potentials of C and B are thermodynamic constraints that can be used to represent experimental synthesis conditions. Under the thermal equilibrium growth conditions, the stable generation of the host material C₃B should satisfy the equation:

$$3\mu_{\rm C} + \mu_{\rm B} = \mu_{\rm C3B} \tag{S2}$$

Where μ_C and μ_B represent the chemical potentials of C and B in C₃B and μ_{C3B} is the chemical potential of C₃B. The optimized static energy E_{C3B} for per formula unit in C₃B can be expressed as

$$E_{C3B} = 3E_{C} + E_{B} + \Delta H_{f}(C_{3}B)$$
 (S3)

Where E_C is the energy of a C atom in the graphene, E_B is the energy of a C atom in the borophene, and $\Delta H_f(C_3B)$ is the formation enthalpy per formula C_3B . Under the thermal equilibrium growth conditions, $E_{C3B} = \mu_{C3B}$; then, from **Eqs. (S2)** and **(S3)**, we have the equation

$$\Delta H_{f}(C_{3}B) = 3(\mu_{C} - E_{C}) + (\mu_{B} - E_{B}) = 3\Delta \mu_{C} + \Delta \mu_{B}$$
(S4)

where, $\Delta \mu_{C} = \mu_{C} - E_{C}$ and $\Delta \mu_{B} = \mu_{B} - E_{B}$ are defined as the chemical potential changes referenced to the elemental bulk/gaseous state, depending on the growth environment. With the **Eqs. (3)** and **(4)**, the calculated PBE formation enthalpies per formula unit C₃B is -0.68 eV. Then, under C-poor (B-rich) conditions $\Delta \mu_{B}=0$, $\mu_{B}=E_{B}$, $\Delta \mu_{C}=\Delta H_{f}(C_{3}B)/3$, thus, we can get μ_{C} . Similarly, we can get the chemical potential of C/B atom under B-rich (C-poor) conditions. The acquired chemical potentials for μ_{B} and μ_{C} under the two extreme growth conditions are shown in **Table S1**.

Table S1. The chemical potentials calculated using the PBE functional at the representative limits imposed by the formation of competing impurity–related phases for transition metal single atoms are presented. (unit:eV)

	B rich	C rich
μ(B)	6.70	5.78
μ(C)	8.91	9.22
μ(Ti)	11.02	12.87
μ(V)	11.21	13.06
μ(Cr)	10.57	12.39
μ(Mn)	10.00	10.92
μ(Fe)	8.99	9.92
μ(Co)	7.84	8.77
μ(Ni)	5.95	6.88
μ(Cu)	3.71	3.71
μ(Zr)	11.58	13.42
μ(Nb)	14.99	14.68
μ(Mo)	12.31	13.23
μ(Ru)	10.10	11.95
μ(Rh)	8.05	8.98
μ(Pd)	5.57	6.12
μ(Ag)	1.16	3.01
μ(Hf)	12.98	14.83

μ(Ta)	13.75	15.60
μ(W)	13.44	14.37
μ(Re)	13.68	15.53
μ(Os)	11.86	13.71
μ(Ir)	9.27	10.20
μ(Pt)	6.47	7.39
μ(Au)	1.22	3.07

When the extrinsic element doping is studied theoretically, the initial step also involves determining the chemical potential of the doping element^{2, 3}. According to the general rules²⁻⁷, to prevent the precipitation of source elements during doping using TM elements, it is required that

$$\Delta \mu_{\rm X} < 0 \tag{S5}$$

Where, $\Delta \mu_X = {}^{\mu_X} {}^{-E_X}$, is defined as the change in the chemical potential of the reference element in the bulk/gas state, depending on the growth environment. E_X is the energy of one X atom in an X crystal. We take Ti doping as an example, the upper bound is $\Delta \mu_{Ti} < 0$ as. On the other hand, to prevent the formation of secondary phases. Such as TiC, TiB, and TiB₂, because of the extrinsic doping in C₃B, it is required that

$$\Delta \mu_{\rm Ti} + \Delta \mu_{\rm C} < \Delta H_{\rm f}({\rm TiC}) \tag{S6}$$

$$\Delta \mu_{\rm Ti} + \Delta \mu_{\rm B} < \Delta H_{\rm f}({\rm TiB}) \tag{S7}$$

$$\Delta \mu_{\rm Ti} + 2\Delta \mu_{\rm B} < \Delta H_{\rm f} ({\rm TiB}_2)$$
(S8)

In which, $\Delta H_f(TiB)$, $\Delta H_f(TiB_2)$ and $\Delta H_f(TiC)$ are the formation enthalpies of per molecular unit of TiB, TiB₂ and TiC, respectively, and it can be calculated by^{5, 8}

$$\Delta H_{f}(TiC) = E(TiC) - E_{Ti} - E_{C}$$
(S9)

$$\Delta H_{f}(TiB) = E(TiB) - E_{Ti} - E_{B}$$
(S10)

$$\Delta H_{f}(TiB2) = E(TiB2) - E_{Ti} - 2E_{B}$$
(S11)

Where, E(TiC), E(TiB) and E(TiB₂) are the total energy per formula TiC, TiB and TiB₂. The control of chemical potentials of TM anchored on the C₃B monolayer is crucial to avoid undesired phase formation^{4, 8, 9}. We conducted PBE functional calculations using experimental lattice constants of impurity–related phases for TMs such as Ti, V, Cr, Mn, Fe, Co, Ni, Cu, Zr, Nb, Mo, Ru, Rh, Pd, Ag, Hf, Ta, W, Re, Os, Ir, Pt, and Au anchored on the C₃B monolayer, as summarized in **Table S2**. The PBE calculated formation enthalpy (given per formula unit) for the

impurity-related phases are shown in **Table S3**, together with the Oqmd^{10, 11} (the Open Quantum Materials Database will be freely available for public use at <u>https://oqmd.org/</u>) values. It is found that the calculated values are in good agreement with the Oqmd ones.

Table S2. Experimental lattice constants of impurity-related phases (Ti, V, Cr, Mn, Fe, Co, Ni, Cu,

	Space group	Experiment			
TiB ₂	P6/mmm	a=b=3.029, c=3.221; α=β=90°, γ=120° ¹²			
TiB	Pnma	a= 6.111, b=3.045, c=4.557; α = β = γ =90° ¹³			
TiC	$Fm\bar{3}m$	a=b=c=4.331; α = β = γ =90° ¹⁴			
VB_2	P6/mmm	a=b=2.994, c=3.018; α=β=90°, γ=120° ¹⁵			
VB	Cmcm	a=b=4.286, c=2.963; α = β =90°, γ =138.454° ¹⁶			
VC	$Fm\bar{3}m$	a=b=c=2.935; α = β = γ =60°17			
CrB ₂	P6/mmm	a=b=2.956, c=3.023; α = β =90°, γ =120° ¹⁸			
CrB	I4 ₁ /amd	a=b=c=8.100; α = β =159.256°, γ =29.502° ¹⁹			
Cr_3C_2	Pnma	a=5.477, b=2.786, c=11.453; α = β = γ =90° ²⁰			
Mn_7C_3	Pnma	a=4.439, b=6.794, c=11.721; α=β=γ=90° ²¹			
MnB	Pnma	a=5.445, b=2.983, c=4.114; $\alpha = \beta = \gamma = 90^{\circ 22}$			
Fe ₂ B	I4/mcm	a= b=5.052, c= 4.231; $\alpha=\beta=\gamma=90^{\circ23}$			
FeB	I4 ₁ /amd	a=b=c=7.758; α = β =158.258°, γ =30.938° ²⁴			
Fe ₃ C	Pnma	a=5.30, b=6.711, c=4.475; α=β=γ=90° ²⁵			
FeC	Fm ³ m	a=b=c=3.985; α = β = γ =90° ²⁶			
CoB	Pnma	a=5.152, b=3.058, c=3.917; α = β = γ =90° ²⁷			
Co ₂ C	Pnma	a=4.359, b=4.363, c=2.874; α = β = γ =90° ²⁸			
Ni ₃ B	Pnma	a= 5.202, b= 6.641, c= 4.389; $\alpha = \beta = \gamma = 90^{\circ 29}$			
Ni ₂ B	I4/mcm	$a=b=c=4.119; \alpha=\beta=105.592^{\circ}, \gamma=117.544^{\circ23}$			
NiB	Cmcm	a=b=3.973, c=2.991; α=β=90°, γ=136.575° ³⁰			
Ni ₃ C	R ³ c	a=b=c=5.079; α = β = γ =53.819° ³¹			
ZrB_2	P6/mmm	a=b=3.167, c=3.529; α = β =90°, γ =120° ¹²			
ZrC	$Fm\bar{3}m$	a=b=c=4.692; α = β = γ =90° ³²			
NbB ₂	P6/mmm	a=b=3.102, c=3.321; α=β=90°, γ=120° ³³			
NbB	Cmcm	a=3.309, b=8.792, c=3.171; $\alpha = \beta = \gamma = 90^{\circ 16}$			
NbC	Fm ³ m	a=b=c=4.473; α = β = γ =90° ³⁴			
MoB ₂	R ³ m	a=b=3.02, c=20.02; α = β =90°, γ =120° ³⁵			
MoB	I4 ₁ /amd	a=b=3.126, c=3.221; α = β = γ =90° ³⁶			
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Ru, Rh, Pd, and Ag) are used for the PBE functional calculations. The values a, b, and c are in Å.

Mo ₂ C	P ³ m2	a=b=2.884, c=5.462; α = β =90°, γ =120° ³⁷
MoC	рб _{т2}	a=b=2.903, c=2.828; α=β=90°, γ=120° ³⁸
RuB_2	Pmmn	a=4.645, b=2.865, c=4.045; α = β = γ =90° ³⁹
Ru_2B_3	P6 ₃ /mmc	a=b=2.917, c=12.856; α = β =90°, γ =120° ⁴⁰
RuC	pēm2	a=b=2.908, c=2.822; α=β=90°, γ=120° ⁴¹
RhB	Cmcm	a=3.382, b=5.86, c=4.157; α=β=γ=90°42
Rh_2B	Pnma	a=5.36, b=3.836, c=6.813; $\alpha = \beta = \gamma = 90^{\circ 43}$
Pd_5B_2	C12/c1	a=12.776, b=4.95, c=5.47; α =90°, β =90.05°, γ =90°44
Pd ₃ B	Pnma	a=5.464, b=7.563, c=4.85; α = β = γ =90°45
PdC	Fm ³ m	a=b=c=4.432; $\alpha = \beta = \gamma = 90^{\circ 46}$
AgB ₂	P6/mmm	a=b=3.016, c=4.079; α=β=90°, γ=120°47
HfB_{2}	P6/mmm	a=b=3.141, c=3.484; α=β=90°, γ=120°48
HfC	Fm ³ m	a=b=c=3.284; α=β=γ=60°49
TaB_2	P6/mmm	a=b=3.10, c=3.318; α = β =90°, γ =120° ¹²
TaC	Fm ³ m	a=b=c=3.166; $\alpha = \beta = \gamma = 60^{\circ 50}$
TaB	Cmcm	a=b=4.662, c=3.164; α=β=90°, γ=138.638° ⁵¹
W_2B	I4/mcm	a=b=c=4.621; α = β = γ =117.737° ⁵¹
WB	Cmcm	a=b=4.662, c=2.853; α = β =90°, γ =138.638° ⁵¹
WC	рб _{m2}	a=b=2.923, c=3.221; α = β =90°, γ =120° ⁵²
ReB	pē _{m2}	a=b=2.897, c=2.941; α=β=90°, γ=120° ⁵³
OsB ₂	Pmmn	a=2.884, b=4.693, c=4.082; α = β = γ =90° ³⁹
OsB	pē _{m2}	a=b=2.895, c=2.876; α = β =90°, γ =120° ⁵⁴
OsC	pē _{m2}	a=b=2.994, c=2.730; α=β=90°, γ=120° ⁵⁵
IrB	Cmcm	a=b=3.553, c=3.939; α=β=90°, γ=111.102° ⁵⁶
IrC	pē _{m2}	a=b=3.045, c=10.431; α = β =90°, γ =120° ⁵⁷
Pt ₂ B	P6 ₃ /mmc	a=b=2.936, c=3.221; α=β=90°, γ=120° ⁵⁸
PtB	I4/mmm	a=b=c=4.651; α = β =131.311°, γ =71.319°59
PtC	Fm ³ m	a=b=c=3.163; α = β = γ =60° ⁶⁰
AuB_2	P6/mmm	a=b=3.005, c=4.233; α=β=90°, γ=120°47

Table S3. The formation enthalpy per formula unit for impurity–related phases in $TM@C_3B$ (B and C sites) was calculated using the PBE functional. For comparison, we referenced 67 values from the Open Quantum Materials Database (OQMD), which are publicly accessible at OQMD^{10, 11}. (unit: eV)

PBE	Oqmd

\triangle Hf(TiB ₂)	3.18	3.20
\triangle Hf(TiB)	1.67	1.70
\triangle Hf(TiC)	1.64	1.61
\triangle Hf(VB ₂)	2.23	2.20
\triangle Hf(VB)	1.71	1.67
\triangle Hf(VC)	0.87	0.78
\triangle Hf(CrB ₂)	1.04	1.01
\triangle Hf(CrB)	1.07	1.05
\triangle Hf(Cr ₃ C ₂)	0.66	0.52
\triangle Hf(Mn ₇ C ₃)	0.83	0.85
\triangle Hf(MnB)	1.00	1.00
\triangle Hf(Fe ₂ B)	0.92	0.94
\triangle Hf(FeB)	0.75	0.76
\triangle Hf(Fe ₃ C)	0.21	0.19
\triangle Hf(FeC)	1.20	1.20
\triangle Hf(CoB)	0.82	0.82
\triangle Hf(Co ₂ C)	0.27	0.28
△Hf(Ni ₃ B)	1.07	1.07
△Hf(Ni ₂ B)	0.89	0.88
\triangle Hf(NiB)	0.49	0.49
△Hf(Ni ₃ C)	0.22	0.24
$ riangle Hf(ZrB_2)$	3.05	2.98
\triangle Hf(ZrC)	1.72	1.62
\triangle Hf(NbB ₂)	2.21	2.09
\triangle Hf(NbB)	1.61	1.55
\triangle Hf(NbC)	4.47	4.06
\triangle Hf(MoB ₂)	2.72	2.61
\triangle Hf(MoB)	1.02	1.00
\triangle Hf(Mo ₂ C)	1.64	1.61
\triangle Hf(MoC)	0.19	0.18
\triangle Hf(RuB ₂)	1.72	1.66
\triangle Hf(Ru ₂ B ₃)	1.64	1.61
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\triangle Hf(RuC)	1.35	1.30
\triangle Hf(RhB)	0.78	0.78
$ riangle Hf(Rh_2B)$	0.62	0.63
\triangle Hf(Pd ₅ B ₂)	1.83	1.32
\triangle Hf(Pd ₃ B)	1.05	1.04
∆Hf(PdC)	2.91	2.97
\triangle Hf(AgB ₂)	1.54	1.56
\triangle Hf(HfB ₂)	3.05	3.07
\triangle Hf(HfC)	1.89	1.90
\triangle Hf(TaB ₂)	1.94	1.95
∆Hf(TaC)	1.16	1.17
\triangle Hf(TaB)	1.63	1.63
$ riangle Hf(W_2B)$	0.80	0.78
\triangle Hf(WB)	0.50	0.71
\triangle Hf(WC)	0.29	0.25
\triangle Hf(ReB)	0.56	0.56
\triangle Hf(OsB ₂)	0.62	0.64
\triangle Hf(OsB)	0.46	0.47
\triangle Hf(OsC)	1.82	1.82
\triangle Hf(IrB)	0.42	4.28
\triangle Hf(IrC)	2.46	2.45
\triangle Hf(Pt ₂ B)	0.58	0.36
\triangle Hf(PtB)	0.38	0.36
\triangle Hf(PtC)	2.92	2.95
\triangle Hf(AuB ₂)	1.98	2.00

• q is the defect charge state (q > 0, when electrons are removed and q < 0, when electrons are added).

E_F is the Fermi energy, which varies in the band gap range of the defect-free supercell. E_{VBM} is referenced to the defect-free supercell's valence-band maximum (VBM). The obtained PBE band gap is used as the changing range of Fermi energy.

From the calculated defect formation energy, the transition level ($\varepsilon(q_1/q_2)$) is defined as the Fermi energy, for which the formation energies of the two charge states q_1 and q_2 of a defect are equal, and given as⁶¹

$$\varepsilon(q_1/q_2) = \frac{E_f(D^{q_1}) - E_f(D^{q_2})}{q_2 - q_1}$$
(S12)

where $E_f(D^q)$ is the above–defined defect formation energy for defect *D* in a charge state *q*. The transition level of a defect is the Fermi level position, where the defect can accept/donate electrons. A deep–level means that removing a hole or electron from the trap to the valence or conduction band requires relatively high energy^{61, 62}.

Note S2 Computational hydrogen electrode

The pathways of the alternating, distal and enzymatic have been summarized in detail. In an acid electrolyte, the distal mechanism can be decomposed into the following elementary steps:

$$* + N_2(g) \rightarrow *N_2 \tag{S13-a}$$

$$*N_2 + H^+ + e^- \rightarrow *NNH$$
 (S13-b)

*NNH + H⁺ + e⁻
$$\rightarrow$$
 *NNH₂ (S13-c)

*NNH₂ + H⁺ + e⁻
$$\rightarrow$$
 *N (S13-d)

$$*N + H^+ + e^- \rightarrow *NH$$
 (S13-e)

$$*NH + H^+ + e^- \rightarrow *NH_2$$
 (S13-f)

$$*NH_2 + H^+ + e^- \rightarrow *NH_3$$
 (S13-g)

The alternating or enzymatic mechanisms can be decomposed into the following elementary steps:

$$* + N_2(g) \rightarrow *N_2 \tag{S14-a}$$

$$*N_2 + H^+ + e^- \rightarrow *NNH$$
 (S14-b)

*NNH + H⁺ + e⁻
$$\rightarrow$$
 *NHNH (S14-c)

*NHNH + H⁺ + e⁻
$$\rightarrow$$
 *NHNH₂ (S14-d)

$$*NHNH_2 + H^+ + e^- \rightarrow *NH_2NH_2$$
(S14-e)

$$*\mathrm{NH}_{2}\mathrm{NH}_{2} + \mathrm{H}^{+} + \mathrm{e}^{-} \rightarrow *\mathrm{NH}_{2}$$
(S14-f)

*
$$NH_2 + H^+ + e^- \rightarrow *NH_3$$
 (S14-g)

where * represents the active site on the C₃B catalysts, and g represent the gas phase, respectively.

To compute the Gibbs free energy change (ΔG) of each elementary step of

electrochemical NH₃ synthesis, we adopted the computational hydrogen electrode (CHE) model developed by Nørskov et al.⁶³, according to which the ΔG of an electrochemical reaction is defined as:

$$\Delta G = \Delta E - T\Delta S + \Delta ZPE + \Delta GU + \Delta G_{pH}$$
(S15)

Where ΔE is the reaction energy directly obtained from DFT calculations, and T is the temperature (T=298.15 K). ΔZPE and ΔS are the differences between the adsorbed species and the gas phase molecules in zero-point energy and entropy, respectively. ΔG_U is the free energy contribution related to applied potential U, which can be determined as $\Delta G_U = -eU$. The U_L is defined by the equation U_L = $-\Delta G_{max}/e$, where maximum Gibbs free energy change (ΔG_{max}) represents the highest positive Gibbs free energy change during the reduction process. In each NRR pathway, the step with the ΔG_{max} is referred to as the potential-determining step (PDS). ΔG_{pH} is the correction of the free energy of H⁺ ions by the concentration dependence of the entropy: $\Delta G_{pH} = 2.303 \times k_B \times pH$, where k_B is the Boltzmann constant, and the value of pH is set to be 0.

Thus, the Gibbs free energy change for each step of the distal mechanism can be described as follows:

$$\Delta G_1 = \Delta G_{*NN} - \Delta G_{N2}(g) - G^* \tag{S16-a}$$

$$\Delta G_2 = \Delta G_{*NNH} - \Delta G_{*N2} - 1/2G^*H_2 \qquad (S16-b)$$

$$\Delta G_3 = \Delta G_{*NNH2} - \Delta G_{*NNH} - 1/2G^*H_2 \tag{S16-c}$$

$$\Delta G_4 = \Delta G_{\text{NH3}} + \Delta G_{\text{*N}} - \Delta G_{\text{*NNH2}} - 1/2G^{\text{*}}H_2 \tag{S16-d}$$

$$\Delta G_5 = \Delta G_{*NH} - \Delta G_{*N} - 1/2G^*H_2 \qquad (S16-e)$$

$$\Delta G_6 = \Delta G_{*_{NH2}} - \Delta G_{*_{NH}} - 1/2G^*H_2$$
 (S16-f)

$$\Delta G_7 = \Delta G_{*NH3} - \Delta G_{*NH2} - 1/2G^*H_2$$
(S16-g)

The Gibbs free energy change for each step of the alternating or enzymatic mechanism can be described as follows:

$$\Delta G_1 = \Delta G_{*N2} - \Delta G_{N2}(g) - G^* \tag{S17-a}$$

$$\Delta G_2 = \Delta G_{*NNH} - \Delta G_{*N2} - 1/2G^*H_2 \tag{S17-b}$$

$$\Delta G_3 = \Delta G_{*NHNH} - \Delta G_{*NNH} - 1/2G^*H_2 \tag{S17-c}$$

$$\Delta G_4 = \Delta G_{*NHNH2} + \Delta G_{*N} - \Delta G_{*NHNH} - 1/2G^*H_2$$
(S17-d)

$$\Delta G_5 = \Delta G_{*NH2NH2} - \Delta G_{*NHNH2} - 1/2G^*H_2$$
(S17-e)

$$\Delta G_6 = \Delta G_{*NH2} - \Delta G_{*NH2NH2} - 1/2G^*H_2 \qquad (S17-f)$$

$$\Delta G_7 = \Delta G_{*NH3} - \Delta G_{*NH2} - 1/2G^*H_2$$
 (S17-g)

The zero-point energies and entropies of the reaction species were calculated from

the vibrational frequencies. During these frequency calculations, all atoms of substrate were rigidly constrained so that no additional degrees of freedom, due to the catalyst, are introduced in to the reacting system. For gas phase molecule, the entropy term can be expressed as the sum of the translational, rotational and vibrational contributions, whereas for adsorbates the translational and rotational entropy were not taken into account due to negligible contributions. The thermodynamic quantities for the N₂, H₂, NH₃ species in the gas phase are shown in **Table S4**.

Table S4. Computed total energies (E) and thermodynamic quantities, in eV, for the gas phase N_2 , H_2 and NH_3 species. (T = 298.15 K, P = 1 bar), at the 298.15 K, 0.035 bar. In comparison, the experimental entropies (TS_{exp}) of the gas phase N_2 , H_2 and NH_3 are also shown, which are from NIST standard reference database (https://doi.org/10.18434/T4D303).

Species	Е	TS	ZPE	G	TSexp
N ₂	-16.63	0.59	0.15	-17.06	0.59
H_2	-6.77	0.40	0.27	-6.90	0.41
NH ₃	-19.53	0.60	0.91	-19.22	0.60

Note S3 Machine learning (ML)

All the ML algorithms were carried out utilizing the open-source code Scikit-learn in Python3 environment⁶⁴. The ML process was conducted using the Gradient Boosted Regression (GBR), Random Forest Regression (RFR), Support Vector Regression, and the K-Neighbour Regression (KNR) models.

a. The computational method of the GBR regression algorithm

The best performing Gradient Boosted Regression (GBR) model is an integrated ML algorithm that is generated by the integration of weak regression trees^{65, 66}. Given the training samples D = (x1, y1), (x2, y2), ... (xn, yn) the number of leaf nodes in every regression tree is J. We divided the input data into J disjoint areas and defined each regression tree as $t_m(x)$. The training goal of GBR is to minimize the loss function L, and the parameters of decision tree θ_m are determined through empirical risk

minimization:

$$\theta_{m} = \operatorname{argmin}_{\theta} \sum_{i=1}^{n} L(y_{i}, f_{m-1}(x_{i}) + t_{m}(x_{i}))$$
(S18)

The process of GBR training is as follows:

(a) Initialize a regression tree function $f_0(x)$.

(b) Train GBR in the gradient decline direction, and compute the negative gradient value of the loss function as the estimated value of the residual. For the m_{th} iteration, GBR generates a regression tree according to the residual and updates the current function $f_m(x)$.

(c) The final regression model is the weighted sum of several weak regression trees, which is defined as:

$$f_{M}(x) = \sum_{m=1}^{M} t(x;\theta_{m})$$
(S19)

b. The computational method of the RFR regression algorithm

The Random Forest Regression (RFR) model is an ensemble learning method for regression by constructing a multitude of decision trees at training time and outputting the class that is the mode of the classes (classification) or mean/average prediction (regression) of the individual trees⁶⁷. Random decision forests correct for decision trees' habit of overfitting to their training set. The training algorithm for random forests applies the general technique of bootstrap aggregating, or bagging, to tree learners. Given a training set $X = x_1... x_n$ with responses $Y = y_1... y_n$, bagging repeatedly (B times) selects a random sample with replacement of the training set and fits trees to these samples:

(a) For b = 1... B: Sample, with replacement, n training examples from X, Y; call these X_b, Y_b; Train a regression tree f_b on X_b, Y_b.

(b) After training, predictions for unseen samples x' can be made by averaging the predictions from all the individual regression trees on x'

$$\hat{f} = \frac{1}{B} \sum_{b=1}^{B} f_{b}(\mathbf{x})$$
 (S20)

c. The computational method of the SVR regression algorithm

The Support Vector Regression (SVR) model is a supervised learning model with associated learning algorithms that analyze data used for regression analysis⁶⁸. Given a

set of training examples, each marked as belonging to one or the other of two categories, an SVM training algorithm builds a model that assigns new examples to one category or the other, making it a non-probabilistic binary linear classifier (although methods such as Platt scaling exist to use SVM in a probabilistic classification setting). An SVM model is a representation of the examples as points in space, mapped so that the examples of the separate categories are divided by a clear gap that is as wide as possible. New examples are then mapped into that same space and predicted to belong to a category based on the side of the gap on which they fall. In addition to performing linear classification, SVMs can efficiently perform a non-linear classification using what is called the kernel trick, implicitly mapping their inputs into high-dimensional feature spaces. Training the original SVR to:

(a)
$$\begin{array}{c} \text{Minimize} \frac{1}{2} \\ \|\mathbf{w}\|^2 \end{array}$$
(S21)

(b) Subject to $|yi - \langle w, x_i \rangle - b| \le \varepsilon$ (S22)

where x_i is a training sample with a target value y_i . The inner product plus intercept $\langle w, x_i \rangle - b$ is the prediction for that sample, and ε is a free parameter that serves as a threshold: all predictions have to be within a ε range of the true predictions. Slack variables are usually added into the above to allow for errors and to allow approximation in the case the above problem is infeasible.

d. The computational method of the KNR regression algorithm

The K-Neighbour Regression (KNR) model is a non-parametric method proposed by Thomas Cover used for classification and regression⁶⁹. In both cases, the input consists of the k closest training examples in the feature space. The output depends on whether k-NN is used for classification or regression. In k-NN regression, the output is the property value for the object. This value is the average of the values of k nearest neighbors.

In the classification phase, k is a user-defined constant, and an unlabeled vector (a query or test point) is classified by assigning the label which is most frequent among the k training samples nearest to that query point. A commonly used distance metric for continuous variables is Euclidean distance. For discrete variables, such as for text classification, another metric can be used, such as the overlap metric (or Hamming distance). Often, the classification accuracy of k–NN can be improved significantly if the distance metric is learned with specialized algorithms such as Large Margin Nearest

Neighbor or Neighborhood components analysis. This algorithm works as follows:

(a) Compute the Euclidean or Mahalanobis distance from the query example to the labeled examples.

(b) Order the labeled examples by increasing distance.

(c) Find a heuristically optimal number k of nearest neighbors, based on RMSE. This is done using cross-validation

(d) Calculate an inverse distance weighted average with the k-nearest multivariate neighbors.



Ľx



Fig. S1. (a) Uniformly scaled supercell of C₃B monolayer with the lateral lengths equal. (b)(c) (d)(e) Convergence of defect (B and C vacancies in C₃B in -1 charge state) formation energies as a function of uniformly scaled supercell size in all the directions (lateral length: $L_x = L_y = L_z = 20.68$ Å). The bottom x-axis represents the supercell size, which * denotes the cubic supercell size. For example, 4*4 illustrates 4×4 ×1 supercell size.

Our calculated lattice parameters of the C₃B monolayer are a = b = 5.17 Å, with C-C and C-B bond lengths of 1.42 Å and 1.56 Å, respectively, which align well with previously reported data^{70, 71}. In this paper, different charge states of the system are modeled by introducing a uniform background charge (positive or negative) in the supercell. However, this method may induce spurious Coulomb interactions, leading to errors in calculating defect formation energy in supercells. In order to correct the spurious Coulomb interaction, our calculations for the defect formation energies in the C₃B monolayer are performed uniformly in a supercell with $L_x = L_y = L_z^{72, 73}$. This method has been used successfully to investigate the defect formation energies of several 2D materials in the field of electrocatalysis.74-77 To assess the impact of supercell sizes on defect formation energy, we calculate the formation energies of C and B vacancy defects in the -1 charge state using different supercell models: 1*1 (8 atoms), 2*2 (32 atoms), 3*3 (72 atoms), 4*4 (128 atoms), and 5*5 (200 atoms) with equal lateral lengths ($L_x = L_y = L_z$). Here, * denotes the cubic supercell size. For example, 4*4 illustrates $4 \times 4 \times 1$ supercell size. As shown in Fig. S1, our results show that the defect formation energy converges quite well at the supercell size of 4 * 4 ($L_x = L_y = L_z = 20.68$ Å). Thus, we use a supercell size of $4 \times 4 \times 1$ (128 atoms) for all calculations. Correspondingly, the vacuum layer in this supercell is 20.68 Å, which can effectively avoid the periodic interactions along the vacuum direction.



Fig. S2. Band structure of pristine unit cell of C_3B monolayer calculated with PBE functional. The band gap is 0.68 eV, and the Fermi level is set to 0 eV.

Note S4 GGA+U

Considering the strong correlation effect of transition metals, we used the GGA+U method to further refine the overpotential of $V_C^{1+}@C_3B$. In this paper, U is introduced as an empirical parameter, with $U_{eff} = 3.1 \text{ eV}$ applied to the 3d orbital of V, as derived from earlier computational and experimental studies⁷⁸. The Gibbs free energy diagrams for $V_C^{1+}@C_3B$, calculated using the GGA + U method, are presented in **Fig. S3** and compared with results obtained from the PBE method. In the first hydrogenation step, the energy difference between PBE and PBE+U is 0.39 eV and 0.63 eV, respectively, while in the subsequent steps, the difference between them is small. The overpotential of $V_C^{1+}@C_3B$ for the PBE approach is 0.39 V, whereas it is 0.63 V for the PBE + U approach. These findings suggest that the PBE method is adequate for describing the electrocatalytic activity, in line with previous research^{79, 80}. However, small variations in the U-value of transition metals can significantly affect the NRR overpotential. As a result, the U value has a substantial impact on the NRR overpotential. Currently, the specification of U is often weak, with many authors arbitrarily selecting U values for transition metals, leading to confusion in the conclusions of NRR overpotential

calculations. To avoid this ambiguity, in this paper we employed the PBE method for our calculations and focused on screening systems with promising NRR activity.



Fig. S3. Gibbs free energy diagrams of $V_C^{1+} @C_3B$ using PBE calculation and PBE +U calculation.



Fig. S4. The formation energies of $3d-TM_B@C_3B$ under B-rich and C-rich conditions are plotted as a function of the Fermi level with respect to the VBM. The Fermi level (E_F) at the VBM and the conduction band minimum (CBM) is set to 0.00 and 0.68 eV, respectively. The solid line represents the charge state with the lowest formation energy of all possible charge states.



Fig. S5. The formation energies of $3d-TM_C@C_3B$ under B-rich and C-rich conditions are plotted as a function of the Fermi level with respect to the VBM. The Fermi level (E_F) at the VBM and the conduction band minimum (CBM) is set to 0.00 and 0.68 eV, respectively. The solid line represents the charge state with the lowest formation energy of all possible charge states.



Fig. S6. The formation energies of $4d-TM_B@C_3B$ under B-rich and C-rich conditions are plotted as a function of the Fermi level with respect to the VBM. The Fermi level (E_F) at the VBM and the conduction band minimum (CBM) is set to 0.00 and 0.68 eV, respectively. The solid line represents the charge state with the lowest formation energy of all possible charge states.



Fig. S7. The formation energies of $4d-TM_C@C_3B$ under B-rich and C-rich conditions are plotted as a function of the Fermi level with respect to the VBM. The Fermi level (E_F) at the VBM and the conduction band minimum (CBM) is set to 0.00 and 0.68 eV, respectively. The solid line represents the charge state with the lowest formation energy of all possible charge states.



Fig. S8. The formation energies of $5d-TM_B@C_3B$ under B-rich and C-rich conditions are plotted as a function of the Fermi level with respect to the VBM. The Fermi level (E_F) at the VBM and the conduction band minimum (CBM) is set to 0.00 and 0.68 eV, respectively. The solid line represents the charge state with the lowest formation energy of all possible charge states.



Fig. S9. The formation energies of $5d-TM_C@C_3B$ under B-rich and C-rich conditions are plotted as a function of the Fermi level with respect to the VBM. The Fermi level (E_F) at the VBM and the conduction band minimum (CBM) is set to 0.00 and 0.68 eV, respectively. The solid line segments represent the charge state with the lowest formation energy of all possible charge states.



Fig. S10. The formation energies of 3*d* (a-d), 4*d* (e-f), and 5*d* (i-l) TM doped B and C sites C₃B monolayer under B-rich and C-rich conditions as a function of the $E_{\rm F}$. C₃B monolayer stabilized charge states at the B and C sites of 3*d* (m), 4*d* (n), and 5*d* (o) doped TMs. The $E_{\rm F}$ varies between the valence band maximum (VBM, 0.00 eV) and the conduction band minimum (CBM, 0.68 eV). The TM atoms and their stable charge states considered in this paper are shown.



Fig. S11. The charge transfer of 3*d*, 4*d*, and 5*d* TM@C₃B (B and C sites) in the 0 and +1 charge states.



Fig. S12. Alternating, distal, and enzymatic reaction pathways for NRR.



Fig. S13. Calculated free energy diagrams for NRR through alternating and distal mechanism catalyzed by $TM@C_3B$ ($TM = Ta_B^{1+}$, W_B^0 , W_B^{1+} , Re_B^0 , Os_B^{1+} , and W_C^{1+}).





Fig. S14. Calculated free energy diagrams for NRR through enzymatic mechanism catalyzed by TM@C₃B (TM = V_B^{1+} , Nb_B^0 , Mo_B^0 , Mo_B^{1+} , Ta_B^0 , Ta_B^{1+} , W_B^0 , W_B^{1+} , Re_B^0 , Re_B^{1+} , Ti_C^0 , Ti_C^{1+} , Zr_C^0 , Nb_C^0 , Mo_C^{0+} , Mo_C^{1+} , Hf_C^0 , Hf_C^{1+} , Ta_C^0 , W_C^0 , W_C^{1+} , Re_C^0 , and Re_C^{1+}).



Fig. S15. Calculated free energy diagrams for NRR through enzymatic mechanism catalyzed by (a) $V_C^0@C_3B$, and (b) $V_C^{1+}@C_3B$.

We explore the details of the enzymatic mechanism of NRR on the promising $V_C^0@C_3B$ and $V_C^{1+}@C_3B$, as shown in **Fig. S15**. In the reduction process, adsorbed *N₂ can interact with H⁺/e⁻ pairs to form *NNH. The PDS occurs during the first hydrogenation step, forming the *NNH species with ΔG values of 0.45 and 0.39 eV, respectively. This small barrier can easily be overcome at room temperature or decreased with more negative voltage, which is superior to electrocatalysts reported in the literature, such as V@MoP (-0.65 V)⁸¹, RuB@GY (-0.53 V)⁸² and Fe@Nb₂C (-0.47 V)⁸³.



Fig. S16. Adsorption energies of N_2 and H atom for the HER and NRR on various catalysts.



Fig. S17. Pearson correlation coefficients (PCC) plot of 13 input features with (a) ΔE_{*N2} (end-on), (b) ΔE_{*N2} (side-on), (c) $\Delta G_{N2 \rightarrow N2H}$ (side-on), (d) $\Delta G_{N2 \rightarrow N2H}$ (end-on) and (e) $\Delta G_{NH2 \rightarrow NH3}$. Red, blue, and white colors represent direct, indirect, and no correlations, respectively. (f) |PCC| of the 13 input features.



Fig. S18. (a)–(e) ΔE_{*N2} (end–on), ΔE_{*N2} (side–on), $\Delta G_{N2\rightarrow NNH}$ (enzymatic), $\Delta G_{N2\rightarrow NNH}$ (distal), $\Delta G_{NH2\rightarrow NH3}$ of DFT calculation vs ML predicted energy in the neutral and charged state.



Fig. S19. (a)–(e) ΔE_{*N2} (end–on), ΔE_{*N2} (side–on), $\Delta G_{N2\rightarrow NNH}$ (enzymatic), $\Delta G_{N2\rightarrow NNH}$ (distal), $\Delta G_{NH2\rightarrow NH3}$ of DFT calculation vs ML predicted energy in the neutral state.



Fig. S20. (a)–(e) ΔE_{*N2} (end–on), ΔE_{*N2} (side–on), $\Delta G_{N2\rightarrow NNH}$ (enzymatic), $\Delta G_{N2\rightarrow NNH}$ (distal), $\Delta G_{NH2\rightarrow NH3}$ of DFT calculation vs ML predicted energy in the charged state.



Fig. S21. The band structure of C₃B (unit cell) using the HSE06 method with different Hartree–Fock mixing parameters α , (a) α =0.15, (b) α =0.25, and (c) α = 0.27. The valence band maximum (VBM) position is placed at zero. (d) The bandgap as a function of Hartree–Fock mixing parameters α .

The HSE hybrid functional is adopted to accurately describe the PDOS of $V_{c}^{0}@C_{3}B$ and $V_{c}^{1+}@C_{3}B$ with the screening parameter (ω) set to 0.2 Å⁻¹. We test the bandgap of pure C₃B by adjusting the Hartree–Fock mixing parameter (α) from 0.15 to 0.25. As shown in **Fig. S21**, we find that when $\alpha = 0.27$, the calculated bandgap is 1.85 eV for pure C₃B, close to the bandgap (1.84 eV) obtained by Zhang et al.⁸⁴.



Fig. S22. (a) The partial density of states (PDOS) of V-3*d* in $V_C^0@C_3B$ and $V_C^{1+}@C_3B$ before N_2 adsorption; the PDOS of N_2 is also given for comparison. (b) The PDOS of V-3*d* and N-2*p* in $V_C^0@C_3B$ and $V_C^{1+}@C_3B$ after N_2 adsorption. (c) The PDOS of V-3*d* and N-2*p* in $V_C^0@C_3B$ and $V_C^{1+}@C_3B$ after NNH adsorption. PDOS uses the Heyd–Scuseria–Ernzerhof (HSE) hybrid functional approach, where the E_F is set to zero.

In **Fig. S22a**, the frontier orbitals of the N₂ can be observed, labeled as $2\sigma^*$, 2π , 3σ , and $2\pi^*$, respectively. For the free N₂, the $2\sigma^*$, 2π , and 3σ orbitals lie below the E_F , while the antibonding $2\pi^*$ orbital is located above the E_F , accounting for the high stability of the N₂. However, after N₂ adsorption (**Fig. S22b**), the $2\pi^*$ orbital of N₂ obviously decreases, and there is a significant hybridization between the N-2*p* orbitals and the V-3*d* orbitals, especially below the E_F . This hybridization primarily involves the transfer of electrons from N₂ to the vacant V-3*d* orbitals. Simultaneously, electrons are transferred from the V-3*d* orbitals to the vacant $2\pi^*$ orbitals of N₂, causing the $2\pi^*$ orbitals to split and hybridize with the 3*d* orbitals of V, which forms occupied (unoccupied) $2\pi^*$ orbitals located above (below) the E_F . This "donation and backdonation" mechanism⁸⁵ strengthens the TM-N bond while weakening the N=N bond. After NNH adsorption, in the V_C⁰@C₃B and V_C¹⁺@C₃B, there is a significant hybridization between the N-2*p* and V-3*d* orbitals, as shown in **Fig. S22c**.



Fig. S23. schematic diagram of three moieties for $V_C^0 @C_3B$ and $V_C^{1+} @C_3B$.



Fig. S24. Adsorption free energies of intermediates (*NNH, *NHNH, *NHNH₂, *NH₂NH₂, *NH₂, *NH₃) for TM@C₃B (V_B¹⁺, Nb_B⁰, Mo_B⁰, Mo_B¹⁺, Ta_B⁰, Ta_B¹⁺, W_B⁰, W_B¹⁺, Re_B⁰, Re_B¹⁺, Ti_C⁰, Ti_C¹⁺, V_C⁰, V_C¹⁺, Zr_C⁰, Nb_C⁰, Mo_C⁰⁺, Hf_C⁰⁺, Hf_C¹⁺, Ta_C⁰, W_C⁰⁺, W_C⁰⁺, Re_C⁰⁺, Re_C¹⁺).

Note S5 Solvation effect

Given that the NRR process typically occurs in aqueous electrolyte solutions, this study takes the solvation effect into account. Using $V_C{}^0@C_3B$ and $V_C{}^{1+}@C_3B$ as examples, the enzymatic reaction pathways in the solution are calculated. An implicit solvation model, as implemented in VASPsol, is employed in this paper. As indicated in **Table S5**, the PDS during the hydrogenation process remains the first protonation step when considering the solvation effect. However, the limiting potentials are generally reduced, with the maximum reduction being 0.19 V for $V_C{}^0@C_3B$. This indicates that while the catalytic activity of TM@C_3B ($V_C{}^0$ and $V_C{}^{1+}$) in solution slightly improves.

Table S5. The $\Delta G_{N2 \rightarrow NNH}$, $\Delta G_{NNH \rightarrow NHNH}$, $\Delta G_{NHNH \rightarrow NHNH2}$, $\Delta G_{NHNH2 \rightarrow NH}$, $\Delta G_{NH \rightarrow NH2}$, $\Delta G_{NH2 \rightarrow NH3}$ are calculated using the PBE (without solvation) and PBE+SOL (with solvation) methods in $V_{C}^{0}@C_{3}B$ and $V_{C}^{1+}@C_{3}B$ systems (unit: eV).

тм	$\Delta G_{N2 \rightarrow}$	$\Delta G_{NNH \rightarrow NHN}$	$\Delta G_{\rm NHNH \rightarrow NHN}$	$\Delta G_{NHNH2 \rightarrow N}$	$\Delta G_{NH \rightarrow N}$	$\Delta G_{NH2 \rightarrow N}$
1 171	NNH	Н	H2	Н	H2	H3
V _{C⁰(PBE)}	0.45	-0.15	-0.42	-0.01	-1.58	-0.14
$V_{C}{}^{0}_{(PBE+SOL)}$	0.34	-0.34	-0.83	-0.56	-0.55	-0.78
$V_{C}{}^{1+}_{(PBE)}$	0.39	-0.23	-0.27	-0.40	-1.26	-0.32
V_{C}^{1+} (PBE+SOL)	0.12	-0.27	-0.64	-0.42	-0.73	-0.95

Table S6. The ELF map of 3d, 4d, and 5d TM@C₃B (B and C sites) in the 0 and +1 charge states.

	TM _B (@C ₃ B	TM _C @C ₃ B		
ТМ	Charge state: 0	Charge state: +1	Charge state: 0	Charge state: +1	
Ti					
V					
Cr					
Mn					
Fe					
Co					





9		,s) in the o	und i on	ange states			
-	TM _B @C ₃ B	Cł	narge state	: 0	Ch	arge state:	+1
-	TM	d _{TM-C1}	d _{TM-C2}	d _{TM-C3}	d _{TM-C1}	d _{TM-C2}	d _{TM-C3}
_	Ti	1.97	1.97	1.97	1.97	1.97	1.97
	V	1.90	1.90	1.90	1.89	1.89	1.89
	Cr	1.89	1.89	1.89	1.89	1.89	1.89
	Mn	1.88	1.88	1.88	1.88	1.88	1.88
	Fe	1.81	1.81	1.84	1.83	1.83	1.83
	Co	1.80	1.80	1.80	1.80	1.80	1.80
	Ni	1.84	1.84	1.84	1.84	1.84	1.84
	Cu	1.91	1.91	1.92	1.91	1.91	1.92
	Zr	2.07	2.07	2.09	2.07	2.08	2.09
	Nb	2.01	2.01	2.01	2.00	2.00	2.00
	Мо	1.99	1.99	1.99	2.00	2.00	2.00
	Ru	1.90	1.90	1.90	1.90	1.90	1.90
	Rh	1.91	1.91	1.91	1.91	1.91	1.91
	Pd	1.96	1.96	1.96	1.97	1.97	1.97
	Ag	2.09	2.09	2.09	2.10	2.10	2.10
	Hf	2.06	2.06	2.06	2.06	2.06	2.06
	Та	1.99	1.99	1.99	1.99	1.99	1.99
	W	1.96	1.96	1.96	1.94	1.94	1.94
	Re	1.92	1.92	1.92	1.94	1.94	1.94
	Os	1.90	1.90	1.90	1.90	1.90	1.90
	Ir	1.92	1.92	1.92	1.92	1.92	1.92
	Pt	1.95	1.95	1.95	1.95	1.95	1.95
	Au	2.04	2.04	2.04	2.04	2.04	2.04
-	TM _C @C ₃ B	Cł	narge state	: 0	Ch	arge state:	+1
-	ТМ	d _{TM-B}	d _{TM-C1}	d _{TM-C2}	d _{TM-B}	d _{TM-C1}	d _{TM-C2}
_	Ti	2.34	1.92	1.92	2.60	1.90	1.90
	V	2.23	1.83	1.83	2.22	1.83	1.83
	Cr	2.02	1.85	1.85	2.14	1.87	1.87
	Mn	2.03	1.85	1.85	2.04	1.83	1.83
	Fe	1.89	1.82	1.82	1.89	1.80	1.81
	Co	1.86	1.78	1.78	1.88	1.78	1.78
	Ni	1.89	1.81	1.81	1.92	1.82	1.82
	Cu	2.00	1.90	1.90	2.01	1.90	1.90

Table S7. The bond lengths between TM and the adjacent atoms (C and B) at the 3*d*, 4*d*, and 5d-TM@C₃B (B and C sites) in the 0 and +1 charge states.

Zr	2.47	2.06	2.05	2.67	2.04	2.04
Nb	2.32	1.94	1.94	2.32	1.94	1.94
Mo	1.95	1.91	2.02	2.12	1.92	1.93
Ru	1.94	1.88	1.88	1.94	1.89	1.89
Rh	1.96	1.91	1.91	1.98	1.90	1.90
Pd	2.00	1.97	1.97	2.02	1.98	1.98
Ag	2.16	2.12	2.12	2.17	2.13	2.13
Hf	2.40	2.04	2.04	2.61	2.01	2.01
Та	2.28	1.93	1.93	2.28	1.94	1.94
W	2.05	1.90	1.90	2.07	1.91	1.91
Re	1.96	1.89	1.89	2.04	1.89	1.89
Os	1.96	1.88	1.88	1.96	1.88	1.88
Ir	1.97	1.90	1.90	1.98	1.90	1.90
Pt	1.99	1.95	1.95	2.01	1.96	1.96
Au	2.09	2.06	2.06	2.10	2.07	2.07

 Table S8. Physical properties for the transition metal atoms.

TM	χm	Ei	r _d	N _m	m	Ne	θ_d
Ti	1.54	6.83	1.46	0.08	47.88	4	2
V	1.63	6.74	1.35	0.53	50.94	5	3
Cr	1.66	6.76	1.28	0.67	52	6	5
Mn	1.55	7.43	1.26	-0.5	54.94	7	5
Fe	1.83	7.9	1.27	0.15	55.85	8	6
Co	1.88	7.86	1.25	0.67	58.93	9	7
Ni	1.91	7.63	1.25	1.16	58.69	10	8
Cu	1.9	7.72	1.28	1.24	63.55	11	10
Zr	1.33	6.95	1.6	0.43	91.22	4	2
Nb	1.6	6.77	1.47	0.89	92.91	5	4
Мо	2.16	7.18	1.4	0.75	95.96	6	5
Ru	2.2	7.36	1.34	1.06	101.1	8	7
Rh	2.28	7.46	1.35	1.14	102.9	9	8
Pd	2.2	8.33	1.38	0.56	106.4	10	10
Ag	1.93	7.57	1.45	1.31	107.9	11	10
Hf	1.3	7	1.58	0.01	178.5	4	2
Та	1.5	7.88	1.47	0.32	180.9	5	3
W	2.36	7.98	1.41	0.82	183.9	6	4
Re	1.9	7.87	1.38	0.15	186.2	7	5
Os	2.2	8.7	1.35	1.1	190.2	8	6

Pt2.288.961.392.14195.1109Au2.549.221.442.321971110	Ir	2.2	9	1.36	1.57	192.2	9	7
Au 2.54 9.22 1.44 2.32 197 11 10	Pt	2.28	8.96	1.39	2.14	195.1	10	9
	Au	2.54	9.22	1.44	2.32	197	11	10

Table S9. Characteristics of TM@C₃B (B and C sites) in the 0 and +1 charge states. magnetic moment (μ_{TM} , μ_B), *s*-band center (ε_s , eV), p-band center (ε_p , eV), *d*-band center (ε_d , eV),

,	TM _B @C ₃ I	3	Charge state: 0				
TM	μ_{TM}	\mathcal{E}_{s}	$\varepsilon_{ m p}$	ε _d	R _{TM}	Qe	
Ti	0.87	-0.82	1.57	1.05	1.97	1.47	
V	2.00	-1.53	1.05	0.52	1.90	1.33	
Cr	3.00	-1.49	1.41	0.51	1.89	1.11	
Mn	2.00	-2.15	0.85	-0.83	1.88	1.02	
Fe	1.00	-2.38	0.72	-0.79	1.82	0.77	
Co	0.00	-2.36	0.36	-1.24	1.80	0.58	
Ni	0.86	-2.89	-0.19	-2.58	1.84	0.56	
Cu	1.96	-2.44	-0.03	-4.06	1.91	0.59	
Zr	0.90	-0.98	1.78	1.16	2.08	1.66	
Nb	2.00	-0.58	2.07	0.50	2.01	1.49	
Mo	3.00	-1.35	1.54	-0.09	1.99	1.06	
Ru	1.00	-1.48	1.92	-0.93	1.90	0.62	
Rh	0.00	-4.09	-0.40	-2.30	1.91	0.42	
Pd	0.87	-3.30	-0.71	-3.84	1.96	0.40	
Ag	1.96	-2.40	-0.20	-5.68	2.09	0.50	
Hf	0.86	-1.80	2.11	1.52	2.06	1.65	
Та	1.99	-1.62	1.87	0.36	1.99	1.59	
W	3.00	-2.11	1.74	-0.08	1.96	1.31	
Re	2.00	-2.89	1.43	-0.83	1.92	0.98	
Os	1.00	-2.89	1.71	-1.01	1.90	0.69	
Ir	0.00	-3.04	1.21	-1.95	1.92	0.43	
Pt	0.87	-3.18	0.31	-3.67	1.95	0.36	
Au	1.96	-2.38	0.40	-5.30	2.04	0.37	
	TM _B @C ₃ I	3		Charge s	state: +1		
ТМ	μ_{TM}	\mathcal{E}_{s}	ε _p	ε _d	R _{TM}	Qe	
Ti	0.00	-0.70	1.69	0.98	1.97	1.50	
V	1.00	-1.14	1.52	0.33	1.89	1.39	
Cr	2.52	-1.06	1.73	0.72	1.89	1.15	
Mn	3.00	-1.53	1.09	-0.77	1.88	1.06	

Fe	2.00	-2.08	1.01	-0.92	1.83	0.84
Co	-0.36	-2.07	0.82	-0.61	1.80	0.61
Ni	0.00	-2.77	-0.01	-2.68	1.84	0.62
Cu	0.98	-2.60	-0.04	-4.23	1.91	0.60
Zr	0.00	-0.14	2.04	1.39	2.08	1.69
Nb	1.00	-0.27	2.04	0.46	2.00	1.56
Mo	2.03	-1.00	1.80	-0.22	2.00	1.14
Ru	1.75	-1.56	1.90	-1.02	1.90	0.68
Rh	0.00	-1.96	0.77	-1.49	1.91	0.44
Pd	0.00	-3.15	-0.23	-3.55	1.97	0.42
Ag	0.98	-2.21	0.14	-5.60	2.10	0.52
Hf	0.00	-1.24	2.50	1.74	2.06	1.75
Та	1.00	-2.23	2.03	0.82	1.99	1.65
W	2.00	-0.69	2.21	0.03	1.94	1.40
Re	2.99	-2.38	1.48	-0.90	1.94	1.04
Os	1.85	-2.64	1.78	-1.13	1.90	0.75
Ir	0.00	-2.39	1.64	-1.50	1.92	0.46
Pt	0.00	-2.74	0.75	-3.40	1.95	0.37
Au	0.98	-1.63	0.73	-5.19	2.04	0.39
	TM _C @C ₃ E	3		Charge	state: 0	
TM		Es.	Ep	ε _d	R _{TM}	Qe
	μ_{TM}	5				
Ti	μ _{TM} 0.00	0.09	1.64	1.35	2.06	1.38
Ti V	0.00 0.93	0.09 -0.58	1.64 1.27	1.35 0.17	2.06 1.96	1.38 1.21
Ti V Cr	0.00 0.93 -2.00	0.09 -0.58 -1.41	1.64 1.27 1.51	1.35 0.17 0.27	2.06 1.96 1.91	1.38 1.21 0.87
Ti V Cr Mn	0.00 0.93 -2.00 3.00	0.09 -0.58 -1.41 -1.85	1.64 1.27 1.51 1.28	1.35 0.17 0.27 -0.36	2.06 1.96 1.91 1.91	1.38 1.21 0.87 0.79
Ti V Cr Mn Fe	0.00 0.93 -2.00 3.00 2.00	0.09 -0.58 -1.41 -1.85 -2.61	1.64 1.27 1.51 1.28 0.87	1.35 0.17 0.27 -0.36 -1.01	2.06 1.96 1.91 1.91 1.84	1.38 1.21 0.87 0.79 0.50
Ti V Cr Mn Fe Co	μTM 0.00 0.93 -2.00 3.00 2.00 1.00	0.09 -0.58 -1.41 -1.85 -2.61 -2.40	1.64 1.27 1.51 1.28 0.87 0.90	1.35 0.17 0.27 -0.36 -1.01 -1.21	2.06 1.96 1.91 1.91 1.84 1.81	1.38 1.21 0.87 0.79 0.50 0.29
Ti V Cr Mn Fe Co Ni	0.00 0.93 -2.00 3.00 2.00 1.00 0.00	$\begin{array}{r} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \end{array}$	1.64 1.27 1.51 1.28 0.87 0.90 0.90	1.35 0.17 0.27 -0.36 -1.01 -1.21 -1.63	2.06 1.96 1.91 1.91 1.84 1.81 1.84	1.38 1.21 0.87 0.79 0.50 0.29 0.25
Ti V Cr Mn Fe Co Ni Cu	0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91	$\begin{array}{r} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \end{array}$	1.64 1.27 1.51 1.28 0.87 0.90 0.90 0.35	$ \begin{array}{r} 1.35 \\ 0.17 \\ 0.27 \\ -0.36 \\ -1.01 \\ -1.21 \\ -1.63 \\ -3.66 \end{array} $	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93	1.38 1.21 0.87 0.79 0.50 0.29 0.25 0.38
Ti V Cr Mn Fe Co Ni Cu Zr	0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00	$\begin{array}{r} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \end{array}$	1.64 1.27 1.51 1.28 0.87 0.90 0.90 0.35 1.60	$ \begin{array}{r} 1.35 \\ 0.17 \\ 0.27 \\ -0.36 \\ -1.01 \\ -1.21 \\ -1.63 \\ -3.66 \\ 1.46 \end{array} $	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19	1.38 1.21 0.87 0.79 0.50 0.29 0.25 0.38 1.54
Ti V Cr Mn Fe Co Ni Cu Zr Nb	0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00 0.92	$\begin{array}{r} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \\ -0.32 \end{array}$	1.64 1.27 1.51 1.28 0.87 0.90 0.90 0.90 0.35 1.60 1.44	$\begin{array}{c} 1.35\\ 0.17\\ 0.27\\ -0.36\\ -1.01\\ -1.21\\ -1.63\\ -3.66\\ 1.46\\ 0.33\end{array}$	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19 2.07	1.38 1.21 0.87 0.79 0.50 0.29 0.25 0.38 1.54 1.36
Ti V Cr Mn Fe Co Ni Cu Zr Nb Mo	0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00 0.92 0.00	$\begin{array}{r} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \\ -0.32 \\ -1.17 \end{array}$	$ 1.64 \\ 1.27 \\ 1.51 \\ 1.28 \\ 0.87 \\ 0.90 \\ 0.90 \\ 0.35 \\ 1.60 \\ 1.44 \\ 1.80 $	$\begin{array}{c} 1.35\\ 0.17\\ 0.27\\ -0.36\\ -1.01\\ -1.21\\ -1.63\\ -3.66\\ 1.46\\ 0.33\\ -0.26\end{array}$	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19 2.07 1.96	1.38 1.21 0.87 0.79 0.50 0.29 0.25 0.38 1.54 1.36 0.84
Ti V Cr Mn Fe Co Ni Cu Zr Nb Mo Ru	0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00 0.92 0.00 0.00	$\begin{array}{c} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \\ -0.32 \\ -1.17 \\ -2.08 \end{array}$	$ 1.64 \\ 1.27 \\ 1.51 \\ 1.28 \\ 0.87 \\ 0.90 \\ 0.90 \\ 0.35 \\ 1.60 \\ 1.44 \\ 1.80 \\ 1.55 $	$\begin{array}{c} 1.35\\ 0.17\\ 0.27\\ -0.36\\ -1.01\\ -1.21\\ -1.63\\ -3.66\\ 1.46\\ 0.33\\ -0.26\\ -1.52\end{array}$	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19 2.07 1.96 1.90	$ \begin{array}{c} 1.38\\ 1.21\\ 0.87\\ 0.79\\ 0.50\\ 0.29\\ 0.25\\ 0.38\\ 1.54\\ 1.36\\ 0.84\\ 0.16\\ \end{array} $
Ti V Cr Mn Fe Co Ni Cu Zr Nb Mo Ru Rh	0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00 0.92 0.00 0.00 1.00	$\begin{array}{r} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \\ -0.32 \\ -1.17 \\ -2.08 \\ -2.74 \end{array}$	$ 1.64 \\ 1.27 \\ 1.51 \\ 1.28 \\ 0.87 \\ 0.90 \\ 0.90 \\ 0.35 \\ 1.60 \\ 1.44 \\ 1.80 \\ 1.55 \\ 0.58 $	$\begin{array}{c} 1.35\\ 0.17\\ 0.27\\ -0.36\\ -1.01\\ -1.21\\ -1.63\\ -3.66\\ 1.46\\ 0.33\\ -0.26\\ -1.52\\ -2.02\end{array}$	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19 2.07 1.96 1.90 1.93	$ \begin{array}{c} 1.38\\ 1.21\\ 0.87\\ 0.79\\ 0.50\\ 0.29\\ 0.25\\ 0.38\\ 1.54\\ 1.36\\ 0.84\\ 0.16\\ 0.04 \end{array} $
Ti V Cr Mn Fe Co Ni Cu Zr Nb Mo Ru Ru Rh Pd	μTM 0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00 0.92 0.00 1.00 0.00 0.92 0.00 1.00 0.00	$\begin{array}{c} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \\ -0.32 \\ -1.17 \\ -2.08 \\ -2.74 \\ -2.61 \end{array}$	1.64 1.27 1.51 1.28 0.87 0.90 0.90 0.35 1.60 1.44 1.80 1.55 0.58 0.58	$\begin{array}{c} 1.35\\ 0.17\\ 0.27\\ -0.36\\ -1.01\\ -1.21\\ -1.63\\ -3.66\\ 1.46\\ 0.33\\ -0.26\\ -1.52\\ -2.02\\ -3.03\end{array}$	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19 2.07 1.96 1.90 1.93 1.98	$ \begin{array}{c} 1.38\\ 1.21\\ 0.87\\ 0.79\\ 0.50\\ 0.29\\ 0.25\\ 0.38\\ 1.54\\ 1.36\\ 0.84\\ 0.16\\ 0.04\\ 0.07\\ \end{array} $
Ti V Cr Mn Fe Co Ni Cu Zr Nb Mo Ru Rh Pd Ag	μ-гм 0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00 0.92 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.84	$\begin{array}{c} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \\ -0.32 \\ -1.17 \\ -2.08 \\ -2.74 \\ -2.61 \\ -2.01 \end{array}$	1.64 1.27 1.51 1.28 0.87 0.90 0.90 0.35 1.60 1.44 1.80 1.55 0.58 0.58 0.50	$\begin{array}{c} 1.35\\ 0.17\\ 0.27\\ -0.36\\ -1.01\\ -1.21\\ -1.63\\ -3.66\\ 1.46\\ 0.33\\ -0.26\\ -1.52\\ -2.02\\ -3.03\\ -5.10\end{array}$	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19 2.07 1.96 1.90 1.93 1.98 2.13	$ \begin{array}{c} 1.38\\ 1.21\\ 0.87\\ 0.79\\ 0.50\\ 0.29\\ 0.25\\ 0.38\\ 1.54\\ 1.36\\ 0.84\\ 0.16\\ 0.04\\ 0.07\\ 0.30\\ \end{array} $
Ti V Cr Mn Fe Co Ni Cu Zr Nb Mo Ru Rh Pd Ag Hf	 µтм 0.00 0.93 -2.00 3.00 2.00 1.00 0.00 0.91 0.00 0.92 0.00 0.00 1.00 0.00 0.00 0.84 0.00 	$\begin{array}{c} 0.09 \\ -0.58 \\ -1.41 \\ -1.85 \\ -2.61 \\ -2.40 \\ -2.68 \\ -2.35 \\ 0.31 \\ -0.32 \\ -1.17 \\ -2.08 \\ -2.74 \\ -2.61 \\ -2.01 \\ -0.69 \end{array}$	1.64 1.27 1.51 1.28 0.87 0.90 0.90 0.35 1.60 1.44 1.80 1.55 0.58 0.58 0.50 2.04	$\begin{array}{c} 1.35\\ 0.17\\ 0.27\\ -0.36\\ -1.01\\ -1.21\\ -1.63\\ -3.66\\ 1.46\\ 0.33\\ -0.26\\ -1.52\\ -2.02\\ -3.03\\ -5.10\\ 1.68\end{array}$	2.06 1.96 1.91 1.91 1.84 1.81 1.84 1.93 2.19 2.07 1.96 1.90 1.93 1.98 2.13 2.16	$\begin{array}{c} 1.38\\ 1.21\\ 0.87\\ 0.79\\ 0.50\\ 0.29\\ 0.25\\ 0.38\\ 1.54\\ 1.36\\ 0.84\\ 0.16\\ 0.04\\ 0.07\\ 0.30\\ 1.55\end{array}$

W	0.00	-2.12	1.93	-0.16	1.95	0.95		
Re	1.00	-2.77	1.85	-0.80	1.92	0.49		
Os	0.00	-3.03	1.54	-1.44	1.90	0.19		
Ir	0.99	-3.28	1.26	-2.18	1.92	-0.03		
Pt	0.00	-3.22	1.04	-3.11	1.97	-0.09		
Au	0.81	-3.04	0.52	-4.85	2.07	0.06		
TM _C @C ₃ B				Charge state: +1				
TM	μ_{TM}	\mathcal{E}_{S}	Еp	ε _d	R _{TM}	Qe		
Ti	1.00	0.36	1.36	0.98	2.14	1.44		
V	0.00	-0.71	1.36	0.03	1.96	1.21		
Cr	3.00	-0.66	1.89	0.54	1.96	0.97		
Mn	2.00	-1.62	1.38	-0.54	1.90	0.80		
Fe	1.04	-2.00	1.28	-0.90	1.83	0.52		
Co	0.00	-2.27	0.97	-1.19	1.81	0.34		
Ni	0.95	-2.04	1.00	-1.80	1.85	0.32		
Cu	0.00	-2.08	0.61	-3.61	1.94	0.41		
Zr	1.00	0.26	1.37	1.35	2.25	1.63		
Nb	0.00	0.00	1.62	0.42	2.07	1.38		
Mo	1.00	-1.17	1.89	-0.55	1.99	0.91		
Ru	-0.74	-1.45	2.16	-1.12	1.90	0.25		
Rh	0.00	-2.55	0.86	-2.15	1.93	0.13		
Pd	0.97	-2.35	0.75	-2.91	1.99	0.12		
Ag	0.00	-1.71	0.81	-4.99	2.14	0.32		
Hf	1.00	-0.41	1.82	1.66	2.21	1.63		
Та	0.00	-1.44	1.66	0.46	2.05	1.42		
W	1.00	-1.89	2.03	-0.31	1.97	1.00		
Re	1.99	-2.59	1.84	-0.89	1.94	0.67		
Os	0.78	-2.64	1.85	-1.27	1.91	0.23		
Ir	0.00	-2.93	1.47	-2.19	1.93	0.04		
Pt	0.97	-2.83	1.24	-2.96	1.98	-0.05		
Au	0.00	-2.70	0.80	-4.84	2.08	0.10		

Regressors	ΔE_{*N2} (end-on)			ΔE_{*N2} (side–on)				
	R ²	RMSE	R ²	RMSE	R ²	RMSE	R ²	RMSE
	(train)	(train)	(test)	(test)	(train)	(train)	(test)	(test)
GBR	0.999	0.002	0.943	0.057	0.999	0.001	0.942	0.033
KNR	0.819	0.096	0.770	0.117	0.813	0.061	0.771	0.062
RFR	0.924	0.056	0.884	0.066	0.902	0.038	0.759	0.061
SVR	0.819	0.096	0.770	0.117	0.690	0.079	0.688	0.068
Regressors		$\Delta G_{N2 \rightarrow NNI}$	_H (distal)		Δ	$G_{N2 \rightarrow NNH}$	(enzymati	c)
	R ²	RMSE	R ²	RMSE	R ²	RMSE	R ²	RMSE
	(train)	(train)	(test)	(test)	(train)	(train)	(test)	(test)
GBR	0.999	0.001	0.960	0.096	0.999	0.001	0.929	0.071
KNR	0.901	0.122	0.861	0.178	0.712	0.203	0.702	0.135
RFR	0.952	0.006	0.088	0.913	0.843	0.113	0.819	0.109
SVR	0.949	0.090	0.891	0.141	0.750	0.183	0.649	0.224
Regressors		ΔG_{*NH2}	2→NH3					
	R ²	RMSE	R ²	RMSE				
	(train)	(train)	(test)	(test)				
GBR	0.999	0.001	0.902	0.220				
KNR	0.801	0.254	0.791	0.249				
RFR	0.913	0.146	0.866	0.161				
SVR	0.721	0.305	0.715	0.249				

Table S10. Summary of prediction results of R² scores and RMSE for charge and neutral state of ΔE_{*N2} (end on), ΔE_{*N2} (side on), $\Delta G_{N2 \rightarrow NNH}$ (distal), $\Delta G_{N2 \rightarrow NNH}$ (enzymatic) and $\Delta G_{NH2 \rightarrow NH3}$ based on machine learning models.

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